# Hydrothermal Alteration and Lithogeochemistry of the Boundary Volcanogenic Massive Sulfide (VMS) Deposit, Central Newfoundland, Canada

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## Abstract

The Boundary volcanogenic massive sulfide (VMS) deposit (0.45 Mt @ 3.4% Cu, 4.0% Zn, 1.0 % Pb, 34 g/t Ag) is located in the Tally Pond group (~510 Ma), central Newfoundland, Canada. The deposit is hosted by rhyolitic rocks that are interpreted to have formed within a rifted continental arc on the leading edge of Ganderia. Mineralization consists of massive pyrite, chalcopyrite, and sphalerite lenses. The basal portion of these lenses contain lapilli tuff clasts indicative of replacement style mineralization. Three hydrothermal alteration assemblages are recognized at Boundary: quartz-sericite, chlorite-sericite, and intense chlorite. Lithogeochemical data are useful in identifying key element associations and alteration assemblages. Short-wave infrared (SWIR) spectroscopy data provide an effective vector for Zn mineralization and correlate with whole rock geochemistry. Lithogeochemical and SWIR data have been used to recreate a three-dimensional alteration model of the Boundary deposit that may be useful in further mineral exploration.

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# List of Abbreviations

AAT	Annieopsquotch Accretionary Tract
AI	Hashimoto alteration index
Alt	Alteration
A.N.D. Co.	Anglo Newfoundland Development
	Company
Ana	Anatase
aq	Aqueous
Arg	Argillite
Ave	Average
BBL	Baie Verte Brompton Line
BX	Breccia
BVOT	Baie Verte Oceanic Tract
Carb	Carbonate
CC	Chaotic carbonate
Сср	Chalcopyrite
ССРІ	Chlorite-carbonate-pyrite alteration index
Chl	Chlorite
cm	Centimeter
CREAIT	Core Research Equipment and Instrument
	Training
DBL	Dog Bay Line
DF	Dover Fault
EDX	Energy Dispersive X-Ray
Eq	Equation
EPMA	Electron probe microanalysis
Fig(s)	Figure(s)
Frag(s)	Fragment(s)
FW	Footwall
g/t	Grams per ton
Gal	Galena
GBF	Green Bay fault
Gpa	Giga pascal
HFSE	High field strength elements
Hg-FIMS	Mercury-flow injection mercury system
HREE	Heavy rare earth elements

HW	Hanging wall		
ICP-ES	Inductively coupled emission-mass		
	spectroscopy		
ICP-MS	Inductively coupled plasma-mass		
	spectrometry		
Int	Intrusion		
Int Chl	Intense chlorite		
km	Kilometer		
Kspar, K-feldspar	Potassium feldspar		
LBOT	Lushs Bight Oceanic Tract		
LCF	Lobster Cove fault		
LFSE	Light field strength elements		
LOI	Loss on ignition		
LREE	Light rare earth elements		
LRF	Llovds River fault		
Lt	Lapilli tuff		
m	Meters		
mm	Millimeter		
Mon	Monazite		
MS	Massive sulfide		
Mt	Million tonnes		
nm	Nanometers		
РСА	Principal component analysis		
Pheno(s)	Phenocryst(s)		
ddd	Parts per billion		
ppm	Parts per million		
Pv	Pvrite		
Otz	Ouartz		
REE	Rare earth elements		
RIL	Red Indian Line		
SEM-BSE	Scanning electron microscope back scatter		
	electron		
Ser	Sericite		
Sph	Sphalerite		
SWIR	Short-wave infrared		
VMS	Volcanogenic massive sulfide		
WL	wavelength		
wt %	Weight percent		
Xcutting	Crosscutting		
2D. 3D	Two, three dimensional		
,	,		

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# Chapter 1: Introduction to the Boundary volcanogenic massive sulfide deposit, central Newfoundland, Canada

The Exploits Subzone represents the peri-Gondwanan remnants of the ancient Iapetus Ocean and is composed of continental and intra-oceanic, arc-back arc, and ophiolite complexes (Williams et al., 1988). The subzone is host to a number of volcanogenic massive sulfide (VMS) deposits in a region known as the Victoria Lake Supergroup, in central Newfoundland (Williams, et al., 1988; Swinden, 1991; Evans and Kean, 2002; Piercey and Hinchey 2012). Within the Victoria Lake supergroup, the Tally Pond group hosts the Duck Pond and Boundary bimodal felsic VMS deposits (combined tonnage of 4.1 Mt @ 3.3% Cu, 5.7% Zn, 0.9% Pb, 59 g/t Ag, and 0.8 g/t Au; Squires and Moore, 2004; Piercey and Hinchey, 2012; Piercey et al., 2014). Various regional and deposit scale studies have been undertaken on these deposits, mostly focusing on the general setting and stratigraphy of mineralization (i.e., Squires et al., 2001; Squires and Moore, 2004; Piercey et al., 2014). Despite the regional and depositspecific research, no detailed study of the petrology, chemostratigraphy, and 2D-3D alteration, alteration mineralogy, or lithogeochemistry has been undertaken. This is the primary objective of this thesis. The following sections provide background information on the exploration history of the area, regional tectonic and geological setting, local geology, and a review of the geology of VMS deposits.

#### **1.1-Exploration History**

In 1871 Alexander Murray of the Geological Survey of Newfoundland undertook a general geological survey of the area. The findings of Murray's survey were followed up by the Anglo Newfoundland Development Company (A.N.D.Co.) who, in 1905, secured a 99-year resource lease that included timber, water, and mineral rights to an area over 3,700 km<sup>2</sup> (Evans and Kean, 2002). Intense exploration by A.N.D.Co and partners led to the discovery of the Buchans Mine and several other promising prospects in the early 1900s. Through a series of partnerships and acquisitions, Noranda entered into a joint exploration agreement with Abitibi-Price in 1979, and this led to the discovery of the Boundary (1981) and Duck Pond (1987) VMS deposits (Evans and Kean, 2002). In 1993 Noranda purchased the mineral rights to both the Duck Pond and Boundary deposits along with many other prospects in the region. After five years of further exploration, Noranda gave up its rights to low mineral potential properties and sold the more promising prospects to several junior exploration companies (Evans and Kean, 2002). Thundermin Resources purchased the rights to the Duck Pond and Boundary deposits in March 1999. Upon the completion of feasibility studies for both deposits, Thundermin Resources sold the rights of Duck Pond and Boundary to Aur Resources in 2002, which began mining Duck Pond in 2007 (Evans and Kean, 2002; Mineral Information, 2007). Later in 2007, Teck Cominco (now Teck Resources) purchased 93% of Aur Resources (Teck News Release 2007) and is the current owner of the Duck Pond and Boundary deposits, which are now undergoing closure and reclamation.

#### **1.2 Tectonic Evolution and Regional Geology**

The Appalachian orogeny in Newfoundland is divided into four tectonostratigraphic zones based upon rock type, age, geophysical signatures, and metallogeny (Wiliams, 1979; Williams et al., 1988; Williams, 1995). From west to east, these zones are the Humber Zone, Dunnage Zone, Gander Zone, and the Avalon Zone (Fig. 1.1). These zones represent a sequence of arcs, back arcs, ophiolite sequences, and rifted blocks generated by the opening and closing of both the Iapetus and Rheic oceans during the Taconic, Penobscot, Salinic, Acadian, and Neo-Acadian orogenies (Williams, 1979; Swinden, 1991; van Staal et al., 1998; Zagorevski et al., 2010).

The Taconic Orogeny is divided into three separate events, which took place from the late Cambrian (~500 Ma) until the late Ordovician (~450 Ma), and occured within the Laurentian Humber Zone and the peri-Laurentian side of the Dunnage Zone (Fig. 1.2)(van Staal et al., 2007). The first event (500-493 Ma) was the accretion of the Lushs Bight Oceanic Tract (LBOT) onto the Dashwoods microcontinent as the Taconic Seaway closed (van Staal et al., 2007). This was followed by the obduction of the Baie Verte Oceanic Tract (BVOT) onto the Humber margin as the Taconic Seaway finished closing (~488-460 Ma). The third event (~460-450 Ma) involved the collision the Annieopsquotch Accretionary Tract (AAT), which represents the leading edge of the peri-Laurentian margin, with the Victoria Arc, the leading edge of peri-Gondwana margin (van Staal et al., 2007; Zagorevski et al., 2007b).

The Penobscot Orogeny roughly coincided with the Taconic Orogeny though there is no direct link between the two orogenic events as there were over 2,500 km separating peri-Gondwanan and peri-Laurentian crustal fragments on opposite sides of the Iapetus Ocean (van Staal and Zagorevski, 2012). Extension resulted in the separation of the Penobscot arc from the edge of Gondwana (~509-501 Ma), which created the Rheic Ocean (Neuman, 1967; Colman-Sadd et al., 1992; van Staal and Zagorevski, 2012). The Penobscot arc then began to undergo back-arc extension (~500-485 Ma), which separated the Penobscot arc from the Gander margin until terminal closure (486-478 Ma)(Fig. 1.3)(Colman-Sadd et al., 1992; Zagorevski, et al., 2010).

As subduction continued to close the Iapetus Ocean, a collision between the younger, more buoyant Victoria Arc and the older, denser Penobscot Arc left the Victoria Arc emplaced above the Penobscot Arc (van Staal et al., 2007; Zagorevski, 2007b). Further subduction closed the Iapetus Ocean permanently as the peri-Laurentian AAT collided with the peri-Gondwanan Victoria Arc (460-455 Ma)(van Staal et al., 2007).

The Dunnage Zone of the Appalachians represents the vestiges of the Iapetus Ocean and remnants of continental and intra-oceanic, arc, back-arc, and ophiolite complexes that developed during the third Taconic event and the Penobscot Orogeny (Williams et al., 1988; Swinden, 1991; Zagorevski et al., 2007a). It is divided into two subzones: the Notre Dame Subzone (peri-Laurentian margin) and the Exploits Subzone (peri-Gondwanan margin)(Williams et al., 1988; Williams, 1995; Zagorevski, 2007b). These subzones are juxtaposed against one another along the Red Indian Line, a suture that represents the accretion of peri-Gondwanan and peri-Laurentian elements (~460-455 Ma)(van Staal et al., 1998; van Staal et al., 2007; Zagorevski, 2007a; Zagorevski, 2007b).

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The Victoria Lake supergroup is situated east of the Red Indian Line within the Exploits Subzone (Fig. 1.4). It is bound to the east by the Noel Paul's Line, and to the northeast, it is overlain or in fault contact with Ordovician to Silurian sedimentary rocks of the Badger Group (Zagorevski et al., 2007b). Initially the Victoria Lake supergroup was divided into two main volcanic belts: the Tally Pond and Tulks volcanic belts; however, more detailed geochronological, lithological, and geochemical studies have resulted in the division into six fault-bound packages (Kean and Jayasinghe, 1980; van Rogers et al., 2005; Zagorevski et al., 2007b; Zagorevski et al., 2010). From west to east, these groups are the Wigwam Brook group (~453 Ma)(Zagorevski et al., 2007b), the Pats Pond group (~488 Ma)(Zagorevski et al., 2007b), the Sutherlands Pond group (~462 Ma)(Dunning et al., 1987), the Tulks group (~498 Ma)(Evans et al., 2007b), and the Tally Pond group (~513-509 Ma) (Dunning et al., 1991; McNicoll et al., 2010).

The Tally Pond group, host to the Boundary deposit, is further broken down into two separate formations: the Lake Ambrose formation and Bindon's Pond formation (Fig. 1.5)(Evans and Kean, 2002; Rogers and van Staal, 2002; Squires and Moore, 2004). The Lake Ambrose formation (~513 Ma) is a sequence of basalt-dominated pillowed and massive flows with minor felsic and sedimentary rocks (Evans and Kean, 2002; McNicoll et al, 2010; Piercey et al., 2014). The Bindon's Pond formation (~509 Ma) is predominantly comprised of felsic breccia, tuffs, quartz porphyry, crystal tuff, and flowbanded rhyolite and hosts the Duck Pond, Boundary, and Lemarchant VMS deposits

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(Evans and Kean, 2002; Rogers and van Staal, 2002; Squires and Moore, 2004; Rogers et al., 2005).

#### **1.3 Volcanogenic Massive Sulfide (VMS) Deposits**

Volcanogenic massive sulfide (VMS) deposits are accumulations of Zn-Pb-Cu-(Ag-Au)-bearing sulfide deposits that form at or near the seafloor and are associated with contemporaneous, submarine volcanism in a subaqueous environment (Franklin et al., 1981, 2005; Lydon 1984, 1988; Large, 1992; Barrie and Hannington, 1999; Galley et al., 2007). They form in a variety of tectonic environments, including mid-ocean ridges, intraoceanic and continental arcs, and back-arc basins; and in all cases, they form in extensional environments associated with high heat flow (Franklin et al., 1981, 2005; Lydon 1984, 1988; Large, 1992; Barrie and Hannington, 1999; Galley et al. 2007). The deposits typically occur as polymetallic, massive sulfide lenses containing Cu, Zn, Pb, Au, and Ag that form through the precipitation of metals from heated hydrothermal fluids as they mix with cool seawater (i.e., Franklin et al., 1981, 2005; Lydon 1984, 1988; Large, 1992; Barrie and Hannington, 1999; Galley et al., 2007). In addition to typical exhalative-type deposition, the Boundary deposit exhibits textures indicative subseafloor replacement deposition (Piercey et al., 2014).

Volcanogenic massive sulfide deposits are divided into six groups based on the lithology of the main volcanic or sedimentary units present at the time of deposition (Table 1; Barrie and Hannington, 1999; Franklin et al., 2005; Galley et al., 2007; Gibson et al., 2007). The Boundary deposit is a bimodal-felsic VMS deposit (Fig. 1.6), as are many deposits within the Dunnage Subzone (Piercey, 2007, McNicoll et al., 2008; Piercey and Hinchey, 2012). Bimodal-felsic deposits are polymetallic and typically exhibit high base metal grades, especially for Zn and Pb (Galley et al., 2007; Franklin et al., 2005). The Boundary and Duck Pond deposits also show elevated Cu concentrations, up to 3.3%, in comparison to other Canadian bimodal felsic VMS deposits (Table 1).

Volcanogenic massive sulfide deposits have extensive zones of hydrothermal alteration that vary in mineralogy, shape, and size. Alteration mineralogy and composition varies with proximity to mineralization. Hydrothermal alteration associated with bimodal-felsic deposits can be divided into pipe-like or discordant alteration and semi-conformable or recharge-related alteration (Riverin and Hodgson, 1980; Franklin et al., 1981, 2005; Gemmell and Large, 1992; Large, 1992; Galley, 1993, Piercey, 2009).

Semi-conformable alteration results from lateral fluid flow as seawater is drawn down through fractures. As the water is heated, it interacts with footwall rocks leaching metals and certain other elements at different temperatures (Galley, 1993). The stacking of hydrothermal alteration zones with different mineralogy is a function of the geothermal gradient, often forming zones of spilitization, silicification, epidote-quartz, and carbonatization/potassic alteration (Galley, 2003; Piercey, 2009). Semi-conformable alteration is generally patchy but can extend laterally up to a couple hundred kilometers (Galley, 1993; Franklin et al., 2005).

Pipe-like alteration develops as heated hydrothermal fluids are discharged through vents and crosscuts stratigraphy (Fig. 1.6)(Riverin and Hodgson, 1980; Franklin et al., 1981; Piercey, 2007). The resulting alteration zone is often constrained to only a few hundred meters wide and is controlled by synvolcanic fault structures (Riverin and

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Hodgson, 1979; Lydon, 1984, 1988; Gemmell and Large, 1992; Franklin et al., 2005; Galley et al., 2007; Piercey 2007). Chlorite-quartz alteration makes up the innermost zone that gives way to a middle chlorite-sericite zone, and eventually grades to an outer sericite-quartz zone (Riverin and Hodgson, 1980; Franklin et al., 1981, 2005; Lydon, 1984; Lydon, 1988; Galley et al., 2007).

#### **1.4 Deposit Geology**

The Boundary deposit consists of three separate sulfide lenses, the North zone, South zone, and Southeast zone and contains approximately 0.45 Mt grading 3.5% Cu, 4.0% Zn, 1.0% Pb, and 34.0 g/t Ag (Fig. 1.7)(Squires and Moore, 2004; Piercey and Hinchey, 2012; Piercey et al., 2014). Regional studies and recent airborne magnetic studies show that the North zone and South zone lenses are separated by the Wagner fault (Wagner, 1993; Hennessey, pers. comm.).

Stratigraphically, the Boundary deposit occurs at the contact between an impermeable hanging wall consisting of flow-banded rhyolite flows and breccia, and a permeable footwall made up primarily of felsic lapilli tuff and lesser tuff (Piercey and Hinchey 2012; Piercey et al., 2014). Below the lapilli tuff are interlayered, aphyric rhyolite flows, flow breccias, and aphyric rhyolite jigsaw-fit breccias. The jigsaw-fit breccias are similar to those found in the Duck Pond deposit (Squires and Moore, 2004; Piercey et al., 2014). Deeper in the stratigraphy, pillow basalts are interlayered with aphyric rhyolites and breccias (Piercey and Hinchey, 2012; Piercey et al., 2014). An altered, quartz-feldspar porphyritic intrusion cross-cuts all units in a similar manner as observed at the Duck Pond deposit (Piercey and Hinchey 2012; Piercey et al., 2014).

Mineralization in each of the three lenses is relatively consistently comprised of Cu-Zn massive sulfide mineralization, dominated by chalcopyrite and sphalerite, with variable pyritic sulfides and minor galena (McNicoll et al., 2010; Piercey and Hinchey 2012; Piercey et al., 2014). The mineralization occurs at the hanging wall-footwall contact and locally extends into the porous lapilli tuff for approximately 10 meters (Piercey and Hinchey 2012; Piercey et al., 2014). The lenses are comprised of a series of conformable to semi-conformable stacked sheets that center around synvolcanic structures, resulting in a tree-branch-type morphology (Piercey and Hinchey 2012; Piercey et al., 2014). Mineralization displays five main facies (Piercey and Hinchey 2012; Piercey et al., 2014):

- 1) stringer sulfides with clasts;
- 2) clast-rich massive sulfides;
- 3) massive pyritic sulfides with chalcopyrite stringers;
- 4) chalcopyrite-sphalerite-rich massive sulfide;
- 5) bedded sulfides

The abundance of clast-rich mineralization has led to the interpretation that mineralization formed by both replacement and exhalation processes (Squires and Moore, 2004; McNicoll et al., 2010, Piercey and Hinchey 2012; Piercey et al., 2014).

Alteration consists predominantly of chlorite, variably intense quartz-sericite, and carbonate that occurs in both the hanging wall and footwall (McNicoll et al., 2010; Piercey and Hinchey 2012; Piercey et al., 2014). The footwall exhibits all three types of alteration. The lapilli tuff unit provides the best measure of alteration as it ranges from being relatively fresh, to quartz-sericite altered, to quartz-sericite-chlorite altered, to intense chlorite altered with "chaotic carbonate" or dolomite (Piercey and Hinchey 2012; Piercey et al., 2014). Intense chlorite alteration occurs in discrete vertical zones thought to represent feeder pipes running along the synvolcanic structures that fed the replacement style mineralization (Piercey and Hinchey 2012; Piercey et al., 2014). The hanging wall alteration is less diverse and is dominated by patchy quartz-sericite alteration with minor chlorite (Piercey and Hinchey 2012; Piercey et al., 2014). Spotty iron carbonate alteration is found throughout the region, in the Boundary, Duck Pond, and Lemarchant deposits as an overprint thought to be related to either a late VMS hydrothermal activity or a younger regional metamorphism (van Staal et al, 2007; Piercey and Hinchey 2012; Piercey et al., 2014).

Limited lithochemical analyses have been performed on the Boundary deposit rocks. The most recent by Piercey et al. (2014) consisted of two downhole lithogeochemical profiles from holes thought to be representative of the hydrothermal alteration. A majority of the samples show strong Na<sub>2</sub>O depletions along with high Hashimoto and chlorite-carbonate-pyrite indices (Piercey and Hinchey, 2012; Piercey et al., 2014). The "Duck Pond alteration signature" (Collins, 1989), defined by anomalous Hg contents and high Hg/Na<sub>2</sub>O and Ba/Sr, is also present at the Boundary deposit (Piercey et al., 2014). Limited immobile element analyses revealed signatures of LREE enrichment, flat HREE, and negative Nb, Ti, and Eu anomalies, petrogenetic signatures common in arc environments or re-melted arc basement (Piercey, 2011; Piercey et al., 2014).

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#### **1.5 Thesis Objectives**

Though the Boundary deposit was discovered over thirty years ago, a complete thorough study of this deposit's alteration signature has never been undertaken. An indepth study of the stratigraphy, alteration type and distribution, and lithogeochemistry will be used herein to understand the controls on alteration and deposit formation in order to create a detailed deposit model. The main objectives are as follows:

- build upon the stratigraphic work of Piercey et al. (2014) and study the lithofacies and their relationship to various alteration zones;
- document the mineralogy and spatial distribution of alteration within the deposit through graphic core logging, petrography, lithogeochemistry, and infrared spectroscopy. This will lead to a constrained hydrothermal alteration distribution model for the deposit;
- map mineralogical variations and compositional variations among micas and chlorite by utilizing short-wave infrared (SWIR) spectroscopy;
- understand textural and compositional variations among the different alteration facies to gain a better understanding of alteration history and ore deposition;
- 5) utilize lithogeochemistry to obtain a chemostratigraphy for the deposit. This can be used to calculate mass changes and quantify elemental gains and losses associated with individual alteration zones;

- combine all the above to create 2D and 3D representations of the deposit that may be used to create custom alteration indices for other mineralized zones within the Tally Pond belt;
- 7) determine what factors influenced ore formation. This will aid in understanding why alteration is more heavily concentrated in certain areas and why sulfides were deposited where they were; and
- use the stratigraphic findings and textural evidence to determine the order of events that led to the deposition of the Boundary deposit.

#### **1.6 Methods**

A comprehensive model of the Boundary deposit will be defined using the following methods: 1) identification of stratigraphic units and alteration assemblages documented in drill core; 2) careful sample selection that encompasses all alteration types and lithologies followed by thin section petrography and scanning electron microscope (SEM) analyses, major and trace element analyses, and hyperspectral analysis; 3) 2D-3D modeling of the alteration and lithogeochemistry of the Boundary deposit. A detailed description of each analysis is described below.

#### **1.6.1 Core Logging**

Detailed drill hole logging took place during August-September 2013 and June 2014 at the Duck Pond core storage area. A total of 62 holes were logged in a manner that allowed for complete coverage of the North and South zones (Fig. 1.7). Entire holes were logged (~50 m) in which lithology, mineralization, and alteration were documented. Hand-drawn core logs were created in the field with detailed notes and photographs.

Core logs were later digitized using Adobe Illustrator software (Appendix A). A series of 2-D fence diagrams was also created in which the alteration was extrapolated between holes in order to gain a better understanding of the alteration surrounding the Boundary deposit (Chapter 2).

A total of 237 samples from 51 drill holes were selected based on major lithology or alteration intensity/assemblage changes. Samples (~10 cm) were halved using a water-cooled rock saw in the Core Research Equipment and Instrument Training (CREAIT)-TERRA facility at Memorial University and subsequently photographed before being cut to size for thin sections and geochemistry. All samples also underwent hyperspectral analyses. The results are discussed in Chapter 2.

#### **1.6.2** Lithogeochemistry

All samples were analyzed for major elements and trace elements, rare earth elements, base metals, and trace metals. Samples were sent to Actlabs in Ancaster, Ontario to analyze major oxides and select trace elements. Prior to shipment, the samples were cut to size and washed to avoid cross-contamination. Samples were crushed and pulverized using mild steel at Actlabs. The samples were analyzed using a pre-analysis lithium metaborate/tetraborate fusion, dissolution of the fused bead in nitric acid, and subsequent analysis using inductively coupled plasma emission spectroscopy (ICP-ES). Mercury analyses were also undertaken at Actlabs using cold vapor flow injection mercury spectrometer (CV-FIMS) where a sample is digested with aqua regia to leach out soluble compounds and then analyzed by CV-FIMS. Other trace elements, including rare earth elements, high field strength elements, and volatile metals were obtained at Memorial University using multi-acid digestion and sodium peroxide dissolution with subsequent analyses of solutions by inductively coupled plasma mass spectrometry (ICP-MS)(see Chapter 2).

#### **1.6.3 Petrography and Scanning Electron Microscopy**

A total of 45 samples were cut down to standard microscope slide size (25 mm x 45 mm) before being sent to Vancouver Petrographics, in October 2013, for polished thin sections to be made. Polished thin sections were fabricated for samples that exhibited unique textures and representative lithologies or alteration types. Petrographic work was carried out using a Nikon LV100POL polarizing microscope at Memorial University. Basic petrography, mineral assemblages, and textures were documented using reflected and transmitted light. Scanning electron microscope (SEM) analyses were used to document finer scale textures, trace minerals, and semi-quantitative chemical information and textural relationships for sulfide phases. SEM analyses were undertaken at Memorial University's CREAIT-Micro-Analysis Facility, Bruneau Innovation Centre (MAF-IIC) using the FEI MLA 650F SEM. Characteristic mineral phases and textures were identified using backscatter electron imaging, whereas semi-quantitative compositional data were collected using point analyses and Energy Dispersive X-Ray (EDX) analysis mapping. The SEM analyses were critical in identifying all the mineral phases present within the Boundary deposit (Appendix E).

#### 1.6.4 Short-wave Infrared (SWIR) Spectroscopy

Hyperspectral analysis using short wave infrared (SWIR) spectroscopy was used to map mineralogical variations in drill core samples and compositional variations in

micas and chlorite throughout the deposit. Analyses were completed at Memorial University using a Terraspec<sup>TM</sup> mineral spectrometer with a Hi-Brite Muglight. Minimal sample preparation was needed as SWIR absorption spectra were collected from fresh drill core that had been recently cut and cleaned. Optimization and white references were collected every  $\sim 20$  samples or 40 min to avoid drifts in data calibration. Analyses were only performed in a naturally sunlit room to avoid any interference from artificial lighting. Reference sample measurements were made after each optimization and white reference calibration to measure both precision and accuracy. Reflectance spectra were collected using the RS<sup>3</sup> Spectral Acquisition software. The resultant spectra were processed using The Spectral Geologist Hotcore (TSG) v 7.1.55 software in which the hull correction was applied. Hyperspectral scalars were calculated using a fourth order polynomial fitting curve applied over the ranges 2120-2245 nm (focused at 2180-2230 nm) for the AlOH wavelengths and 2230-2270 nm (focused at 2240-2260 nm) for FeOH. Spectral depth filters were applied to remove background noise: 0.015% for AlOH and 0.010% for FeOH. Due to the variable clast-matrix mineralogy, a minimum of three analyses per sample were made to ensure representative measurements. The SWIR absorption data for each sample are also reported in Appendix B.

#### 1.6.5 Modeling of 2D-3D Alteration

Stratigraphy, lithofacies, and alteration zones were modeled in 2D and 3D. Additional lithogeochemical analyses (elemental gains and losses, metallurgical assays, alteration indices, etc.) were included to identify unique relationships throughout the deposit. Modeling was initially performed using the program Geosoft Target linked to

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ArcGIS. Final models were produced using Leapfrog Geo<sup>™</sup>. Lithofacies zones were created using the logged drill hole data, ore zones used assay data, mass change interpolants used the mass change data from Appendix C. Hyperspectral data points are from Appendix B.

#### **1.7 Co-Authorship Statement**

The design of this research project is attributed to Dr. Stephen J. Piercey. The author conducted the primary research, which includes core logging, sample collection, and optical microscopy. The hyperspectral analysis, Geosoft Target, and Leapfrog Geo aspects of the project were conducted by the author with valuable assistance provided by Jonathan Cloutier. The Principal Component Analysis (PCA) was carried out by the author with assistance from Praise Nyade. Scanning electron microscopy was conducted by the author under the supervision of David Grant. The primary editor of this manuscript is Dr. Stephen J. Piercey, with secondary editing performed by Dr. Graham Layne and Dr. John Hinchey.

#### **1.8 Thesis Presentation**

This thesis consists of three chapters plus supplementary appendices. Chapter 1 serves as an introductory chapter that presents the purpose and format of this thesis by providing background information about the exploration history, regional and local geology, previous work on the deposit, and the methods used in this study.

Chapter 2 is the main body of the thesis and is intended for publication in *Economic* 

Geology or a similar scientific journal. This chapter contains detailed

descriptions of the lithology and alteration assemblages. Through the use of various geochemical techniques and hyperspectral analysis, this chapter presents evidence pertaining to the tectonic setting, a complete deposit alteration model, and details on the Boundary deposit's evolution.

Chapter 3 is a summary of the conclusions drawn from Chapter 2 in addition to additional information provided in the appendices. This chapter also summarizes unresolved questions and directions for future research.

The appendices provide supplementary material not included in Chapter 2. They include digitized drill logs, complete geochemical and hyperspectral datasets, and a compilation of scanning electron microscope (SEM) analyses.

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Туре	Lithology	Tectonic Setting	Ore	Examples
			Grade*	
Pelitic-Mafic	Mature back-arc successions with mafic sills (up to 25%). Less than 5% felsic volcanic rocks.	Sedimented mid- ocean ridges, transforms, or back- arcs	Cu: 1.6% Zn: 2.6% Pb: 0.36% Ag: 29 g/t Au: <0.9 g/t	Windy Craggy, Canada; Besshi district Japan
Back-Arc Mafic	Dominantly mafic flows with up to 10% felsic flows/domes. Mafic sills and dikes are common.	Mature intra-oceanic back-arcs	Cu: 3.2% Zn: 1.9% Pb: 0.0% Ag: 15 g/t Au: 2.5 g/t	South Urals, Russia; Semail, Oman
Bimodal-Mafic	Dominantly mafic flows with up to 25% felsic volcanic rocks.	Rifted oceanic arcs	Cu: 1.7% Zn 5.1% Pb: 0.6% Ag: 45 g/t Au: 1.4 g/t	Flin Flon, Canada; Tambo Grande, Peru
Bimodal-Felsic	Felsic volcanic rocks with 10- 40% mafic flows and sills. Less than 10% terrigenous sediment. May exhibit subaerial or shallow water facies.	Continental margin arcs and related back- arcs	Cu: 1.3% Zn: 6.1% Pb: 1.8% Ag: 123 g/t Au: 2.2 g/t	Boundary, Newfoundland, Canada; Pontides, Turkey
Felsic Siliciclastitc	Siliciclastic rocks up to 75% with felsic volcanic rocks (up to 25%). Less than 10% mafic flows and sills. May exhibit subaerial or shallow water facies.	Mature epicontinental back- arcs	Cu: 1.0% Zn: 4.7% Pb: 2.0% Ag: 53 g/t Au: 0.9 g/t	Bathurst, Canada; Iberian Pyrite Belt, Spain
Hybrid Bimodal-Felsic	Felsic volcaniclastic and siliciclastic rocks.	Combinations of shallow water VMS and epithermal mineralization	NA	Manus Basin, Pacific Ocean

Table 1.1 Classification, type associations, and setting of VMS deposits

\*Average Canadian ore grade Table data compiled from Franklin et al., (2005); Gibson et al., (2007); Galley et al., (2007); and references therein.



Figure 1.1 Geological map of the Newfoundland Appalachians with tectonostratigraphic zones, accretionary tracts, and associated belts. Tectonostratigraphic divisions modified from van Stall (2007), and van Staal and Barr (2012). Abbreviations: BBL=Baie Verte Brompton Line; DF=Dover Fault; DBL=Dog Bay Line; GBF=Green Bay fault; LCF=Lobster Cove fault; LRF=Lloyds River fault; RIL=Red Indian Line.


Figure 1.2 Evolution of the Taconic Orogeny. The first two events are related to the closure of the Taconic/Humber seaway, which caused the obduction of the Lushs Bight Oceanic Tract onto the Dashwoods (A,B; Taconic 1) and the Baie Verte Oceanic Tract onto the Humber margin (C,D; Taconic 2). Taconic 3 (E) involves the formation of the Annieopsquotch Accretionary Tract (AAT) and the location of the eventual collison between the AAT and Victoria Arc, marked by the Red Indian Line (modified from van Staal and Barr, 2012; van Staal et al., 2007).



Figure 1.3 Evolution of the Penobscot and Victoria arc systems. Development of the Penobscot arc-back arc system (A,B). The closure of the Penobscot back-arc led to the Penobscot Orogeny (C). Further closure of the Iapetus ocean (D) prompted the formation of the Popelogan Victoria arc and the corresponding Tetagouche-Exploits back-arc basin (modified from van Staal and Barr, 2012; Zagorevski et al., 2010).







Figure 1.5 Geology of the Tally Pond group and related rocks. Thrust faults, U-Pb zircon age dates, and mines/prospects are shown (modified from Piercey et al., 2014; McNicoll et al., 2010).



Figure 1.6 Ideal cross section of a bimodal-felsic VMS deposit (modified from Galley et al., 2007).



Figure 1.7 Boundary deposit surface geology plan map with distribution of sulfide mineralization. Modified from Hennessey (2013)(Unpublished).

# Chapter 2: Hydrothermal Alteration and Lithogeochemistry of the Boundary Volcanogenic Massive Sulfide (VMS) Deposit, central Newfoundland, Canada

# 2.1 Abstract

The Boundary volcanogenic massive sulfide (VMS) deposit (0.45 Mt @ 3.4% Cu, 4.0% Zn, 1.0% Pb, 34 g/t Ag) is hosted by the Tally Pond group (~510 Ma) of the Victoria Lake supergroup, central Newfoundland, Canada. The Boundary deposit is a type example of a subseafloor replacement-style VMS deposit. The deposit has a felsic hanging wall composed of flow banded, quartz-phyric flows and breccias. The immediate footwall consists of aphyric, felsic lapilli tuff that contains interlayered aphyric, felsic flow breccias and massive flows. Mineralization predominantly occurs within the upper ten meters of the footwall and at the hanging wall-footwall contact. It consists of a broad pyrite lens and smaller individual chalcopyrite and sphalerite lenses. The basal portions of these lenses commonly contain clasts of the host lapilli tuff, typical of replacement-type mineralization. Three distinct hydrothermal alteration assemblages are present at Boundary: quartz-sericite, chlorite-sericite, and intense or pervasive chlorite. The intense chlorite assemblage occurs within the footwall lapilli tuff as both discordant pipes and conformable sheets underlying the ore. The chlorite-sericite assemblage typically forms around the periphery of the intense chlorite assemblage and occurs within all lithofacies, though rarely in the hanging wall. The quartz-sericite

assemblage occurs throughout the deposit, but its intensity increases proximal to Zn-rich mineralization.

The hanging wall and footwall rhyolites have no significant immobile element differences, suggesting a common petrological history. The rhyolites have FIIIa signatures, transitional Zr/Y ratios (2.8-4.5), low Nb/Y ratios, low Zr contents (Zr<200ppm), and low upper crust-normalized La/Sm ratios (<1) consistent with the formation of felsic rocks derived from melting of juvenile to weakly evolved crust within an extensional rift environment. When coupled with previous Nd-Pb isotopic data and zircon inheritance patterns, these immobile element data suggest that the Boundary host rocks were formed in a rifted peri-continental arc environment.

All rocks at the Boundary deposit are altered and contain elevated alteration indices, including high Ba/Sr, Hg/Na<sub>2</sub>O, chlorite-carbonate-pyrite index (CCPI), and Hashimoto alteration index (AI) values. Mass changes and principal component analysis (PCA) identify key element associations related to mineralization and alteration. Intense chlorite alteration exhibits significant mass gains in MgO and Cu, variable Fe<sub>2</sub>O<sub>3</sub> and SiO<sub>2</sub> behavior, and depletions in K<sub>2</sub>O and Ba. The quartz-sericite assemblage has mass gains in SiO<sub>2</sub>, K<sub>2</sub>O, Ba, and Fe<sub>2</sub>O<sub>3</sub>, but no significant enrichment in Cu or MgO. Chlorite-sericite alteration contains variable gains and losses of SiO<sub>2</sub>, K<sub>2</sub>O, Ba, MgO, and Fe<sub>2</sub>O<sub>3</sub> depending on the dominant matrix mineral phase (i.e., chlorite vs. sericite). The principal component analysis shows key element associations that include those involved with high temperature mineralization (Cu, Co, Bi, Mo, As, MgO, W, LOI, Fe<sub>2</sub>O<sub>3</sub>) and low temperature mineralization (Zn, Pb, Sn, Hg). Short-wave infrared (SWIR) spectroscopic data, particularly AlOH and FeOH absorption hulls, differentiate alteration styles and correlate with lithogeochemical results. The AlOH wavelengths increase in length (>2208 nm) proximal to Zn mineralization, and the wavelength variations correspond with relative abundances for sericite and chlorite. Collectively, these data have been used to reconstruct a three-dimensional alteration model of the Boundary deposit.

## **2.2 Introduction**

Volcanogenic massive sulfide (VMS) deposits are important global resources of base and precious metals and one of the best understood mineral deposit types (e.g., Franklin et al., 1981, 2005; Lydon, 1984, 1988; Large, 1992; Franklin, 1995; Barrie and Hannington, 1999; Galley et al., 2007). They are well represented in the ancient record and our understanding of these deposits has been greatly enhanced by observations on the modern seafloor (e.g., Herzig and Hannington, 1995; de Ronde et al., 2005; Wysoczanski, et al., 2012; Gruen et al., 2014; Petersen et al., 2014; de Ronde et al., 2014). Modern seafloor deposits are associated with venting of hydrothermal fluids in black (and white) smokers, and observed mineralization is predominantly exhalative in nature (Hekinian et al., 1980; Lydon, 1988; Herzig and Hannington, 1995; Hannington and Barrie, 1999, German and Von Damm, 2003; Berkenbosch et al., 2012; de Ronde et al., 2014; Petersen et al., 2014). In the ancient record there are also exhalative deposits, but a sub-class of deposits is interpreted to have formed via replacement of host strata near or slightly below the seafloor (Doyle and Huston, 1999; Doyle and Allen, 2003). Replacement-style VMS deposits are also recognized as some of the largest/highest grade VMS deposits globally (Squires et al., 1991, 2001; Galley et al., 1993, 1995; Doyle and Huston, 1999; Hannington et al., 1999; Doyle and Allen, 2003; Bradshaw et al., 2008). Despite their significance and importance, the processes of subseafloor replacement are incompletely understood. Furthermore, detailed studies of the alteration mineralogy and lithogeochemistry and the associated 2D-3D distribution of these variables in replacement deposits are lacking.

The Boundary volcanogenic massive sulfide (VMS) deposit (~0.5 Mt @ 3.5% Cu, 4% Zn, and 1% Pb, 34.0 g /t Ag) in the Canadian Appalachians is an outstanding example of a replacement-style VMS deposit and is one of the best preserved replacement deposits in the geological record (Piercey et al., 2014). The deposit, while small, is flat lying, contains excellent textural and alteration preservation, and lacks a metamorphic overprint found in many ancient replacement systems, therefore making it an ideal location to study alteration, lithogeochemistry, and mineralogy in VMS deposits. The goal of this paper is to build upon the previous work undertaken on the Boundary deposit by Moore (2003), Squires and Moore (2004), Piercey and Hinchey (2012), and Piercey et al. (2014). To build on our knowledge of the Boundary deposit we have undertaken a detailed study of the alteration distribution, alteration assemblage mineralogy, and lithogeochemistry, in both 2D and 3D. The research includes utilization of immobile elements to understand the tectonic setting of deposit formation and the genesis of the host volcanic facies present at Boundary. Mobile element geochemistry, mass balance calculations, and short wave infrared (SWIR) spectroscopy are utilized to construct an alteration model and provide insight into the key controls and processes

involved with alteration and genesis of the Boundary deposit. These results provide both descriptive and genetic insight into subseafloor replacement alteration processes that have implications for similar deposits both in the Appalachians and globally.

# 2.3 Geological Setting

The Newfoundland Appalachians are divided into four tectonostratigraphic zones (Fig. 2.1): the Humber, Dunnage, Gander and Avalon zones (Williams, 1979; Williams et al., 1988). The Dunnage Zone hosts the Boundary deposit and is comprised of arc, back-arc, and ophiolitic rocks that formed along the margins of Laurentia (Notre Dame Subzone) and Gondwana (Exploits Subzone) from the Cambrian to Ordovician (Swinden et al., 1989; Swinden, 1991; Kean et al., 1995; van Staal and Colman-Sadd, 1997; Evans and Kean, 2002; Rogers and van Staal, 2002; Rogers et al., 2006, 2007; van Staal, 2007). The closure of the Iapetus Ocean resulted in the accretion of the Notre Dame Subzone onto the Laurentian margin during the Taconic orogeny (500-450 Ma), whereas the Exploits Subzone was accreted to the Gondwanan margin during the Penobscot orogeny (486-478 Ma)(van Staal, 2007; Zagorevski et al., 2007a, 2010). The two subzones were juxtaposed against each other during the final stages of the Taconic orogeny and created a suture between the two subzones termed the Red Indian Line (Fig. 2.1) (van Staal, 2007; Zagorevski et al., 2012).

Within the Exploits Subzone, the Boundary deposit is hosted by the Tally Pond group (~513-509 Ma; Dunning et al., 1991; McNicoll et al., 2010), one of six fault bounded packages that make up the Victoria Lake supergroup (Fig. 2.2; Evans and Kean, 2002; Zagorevski et al., 2007b, 2010; McNicoll et al., 2010; Piercey et al, 2014). In

addition to the Boundary deposit, the Tally Pond group hosts the Duck Pond and Lemarchant VMS deposits, as well as several other prospects (Squires et al., 1991, 2001; Wagner, 1993; Evans and Kean, 2002; Squires and Moore, 2004; Piercey et al., 2014). The Tally Pond group is divided further into two informal formations: Lake Ambrose formation and the Bindons Pond formation (Rogers and van Staal, 2002; Rogers et al., 2006; Piercey et al., 2014). The Lake Ambrose formation (~513 Ma; Dunning et al., 1991) is predominantly basalt, consisting of massive and pillow flows, volcaniclastic rocks, and lesser felsic and sedimentary rocks (Kean and Evans, 1986; Evans and Kean, 2002, Rogers and van Staal, 2002; Rogers et al., 2006). The Lake Ambrose formation underlies the younger Bindons Pond formation (509 Ma; McNicoll et al., 2010). The Bindons Pond formation, a felsic-dominated unit that consists of felsic flows, volcaniclastic rocks, and carbonaceous clastic sedimentary rocks, hosts the Duck Pond, Lemarchant, and Boundary deposits (Kean and Evans, 1986; Squires et al., 1991; Evans and Kean, 2002; Rogers and van Staal, 2002; Squires and Moore, 2004, Rogers et al., 2006).

# 2.4 Deposit Geology

The Boundary deposit is located approximately 4 km NE of the Duck Pond deposit. It occurs in two fault-separated lenses, the North zone and the South zone (Fig. 2.3) (Wagner, 1993; Piercey et al., 2014). A third lens, the Southeast zone, occurs as an extension of the South zone but was not a focus in this study. The hanging wall rocks exposed at or near the surface below a few meters of till and consist of a quartz-phyric assemblage of massive and flow-banded rhyolite flows and breccias (Fig. 2.4a-b). The hanging wall is partially preserved in the North zone, whereas in the South zone, it has been mostly eroded, thus exposing massive sulfides at surface. The footwall consists of aphyric lapilli tuff with minor breccia and tuff underlain by aphyric breccias and flows (flow banded to massive) (Fig. 2.4c-f). The top of the footwall is comprised of a thin (<10 cm) tuff that has mostly been replaced by sulfides (G. Squires, pers. comm.). The aphyric lapilli tuff makes up the majority of the footwall and is generally clast-supported with 1-3 cm clasts; however, the unit becomes more matrix supported with depth. Bedded tuffs and aphyric rhyolite breccias are interlayered throughout the lapilli tuff. Deeper within the footwall, aphyric rhyolite flows and breccias underlie the lapilli tuff. Intrusions at Boundary are rare; only one quartz-porphyry intrusion has been recognized just east of the South zone, where it crosscuts all units. The lithofacies and stratigraphic successions are shown in cross section in Figures 2.5a and 2.5b.

The Boundary deposit is essentially structurally intact with two exceptions that involve post-mineralization faulting. The first is the N-S trending Wagner fault, which offsets the North zone from the South zone (Fig. 2.3; Wagner, 1993; G. Squires pers. comm.). This fault is easily recognized in drill holes that occur distal from the deposit as intense, friable sericite zones. The second involves thrust faulting in the western portion of the North zone. Shallow, meter-scale thrust faults have resulted in apparent stacked massive sulfide lenses within the hanging wall; however, detailed core logging reveals that faulting took place at the hanging wall-ore contact and imbricated the ore horizon (Fig. 2.5a).

# **2.5 Mineralization and Alteration**

## 2.5.1 Mineralization

In both zones, mineralization occurs as a series of semi-conformable, discontinuous lenses, most of which occur below the contact between the hanging wall quartz-phyric rhyolite flows and breccias and the footwall aphyric rhyolite lapilli tuff (Fig. 2.4A-C, 2.5A-B; Appendix A; Squires and Moore, 2004; Piercey et al., 2014). At the hanging wall-footwall contact, the lenses (up to 50 meters in diameter, ~1-5 meters thick) consist primarily of massive Cu-Zn sulfide cores surrounded by massive pyrite. The contact between the hanging wall and these lenses is typically sharp; however, locally mineralization consists of pyrite-chalcopyrite and displays relict hanging wall flow textures. The basal portions of these large lenses commonly contain lapilli tuff fragments within a sulfide matrix. Bedded sulfides are also present at the hanging wallfootwall contact along the southern edges of both zones and consist of mm- to cm-scale bands of pyrite, chalcopyrite, and sphalerite. Smaller lenses (~10s of meters, usually 1-2 meters thick) consisting of massive to clast-rich chalcopyrite and pyrite occur less than 10 meters below the hanging wall-footwall contact within the footwall lapilli tuff. Deeper in the footwall, pyrite stringers occur within the inter-fragment space of the flows and breccias. The hanging wall in the South zone has mostly been eroded, with massive sulfides exposed near surface. These near surface sulfides have variable degrees of supergene enrichment and contain surface staining and veinlets of chalcocite, bornite, and covellite.

#### 2.5.2 Alteration

Three main alteration assemblages are present in the Boundary deposit: quartzsericite, chlorite-sericite, and intense chlorite (Figs. 2.6-2.8). The most abundant, least intense quartz-sericite alteration assemblage occurs throughout the deposit, both proximal (within meters) and distal (>100 meters) to mineralization. It extends into the uppermost levels of the remnant hanging wall and deep into the footwall appric rhyolitic flows and breccias (Fig. 2.5a -b). Quartz-sericite alteration may be varying shades of grey, as well as white, purple or black. The black quartz-sericite assemblage sometimes occurs near massive to semi-massive sulfide (Fig. 2.6b). The black color does not appear to be caused by any compositional abnormalities or inclusions within the sericite. Clasts and flows are typically dominated by quartz alteration, whereas the matrix contains variable sericite, quartz, and minor epidote (Fig. 2.6c). Chlorite is rare in the quartz-sericite assemblage, but occurs as small pods in heavily sericite altered, relict feldspar grains, or along the edges of sulfides (Fig. 2.6d). In the more intensely quartz-sericite altered parts of the stratigraphy, chlorite becomes a minor matrix constituent ( $\sim$ 5%). Disseminated pyrite is the dominant sulfide phase with lesser sphalerite and galena. Chalcopyrite is present as chalcopyrite disease in sphalerite (Fig. 2.6e; Large, 1992) or as crosscutting veins.

The chlorite-sericite assemblage forms around the intense chlorite assemblage (see below), usually within a few meters of mineralization (Fig. 2.7). Samples from this assemblage contain quartz (50-70%), sericite (20-30%), and chlorite (20-30%). Texturally, the alteration manifests itself with zoned lapilli tuff clasts that have quartz

altered rims and slightly sericite altered cores (Fig. 2.7a-b), whereas the matrix is dominated by fine-grained chlorite and sericite (Fig. 2.7c-d). The accompanying sulfide mineralogy consists of pyrite, chalcopyrite (± Bi-Te-sulfides), sphalerite, and trace galena.

The intense chlorite assemblage occurs in close association with massive and clast-rich sulfides, commonly as semi-conformable sheets that form thin envelopes (< 2 m) around the sulfides (Fig. 2.8). It also defines discordant alteration pipes that crosscut stratigraphy (Fig. 2.5). Within the intense chlorite assemblage, most if not all, of the original host rock has been replaced by chlorite and sulfides (Fig. 2.8a-b). The intense chlorite assemblage consists predominantly of Mg-chlorite (60-80%), quartz (20-40%), and occasional chaotic carbonate (dolomite; ~5%); the latter of which is a feature more common in the neighboring Duck Pond mine (Squires et al., 2001; Squires and Moore, 2004; Piercey et al., 2014). The sulfide mineralogy in this assemblage includes pyrite, chalcopyrite (+/- Bi-Te-sulfides), and minor sphalerite and Co-rich arsenopyrite.

Accessory minerals such as anatase, apatite, and monazite are found in all three zones, whereas xenotime is present only in the quartz-sericite assemblage. Also common throughout the Boundary deposit are mm-scale Fe-Mg-(+/-Mn) carbonate spots (e. g. Fig 2.6a-b; Fig. 2.7a; and Fig. 2.8b). This type of alteration occurs throughout the Tally Pond group and is interpreted to be the result of regional metamorphism during the late Silurian (van Staal, 2007; Piercey et al., 2014).

# 2.6 Lithogeochemistry

## 2.6.1 Methods

A total of 237 samples from the Boundary deposit were selected from 51 diamond drill holes during the 2013 field season. Samples were selected every ~5 meters or wherever there was a major change in the alteration assemblage. Samples included the hanging wall flows and breccias as well as footwall tuff, lapilli tuff, breccias, and flows from both the North and South zones. Samples that exhibited evidence for faulting such as fractured core or extensive quartz veining were avoided.

All samples were analyzed for major element oxides (SiO<sub>2</sub>, Al<sub>2</sub>O<sub>3</sub>, Fe<sub>2</sub>O<sub>3</sub>, MnO, MgO, CaO, Na<sub>2</sub>O, K<sub>2</sub>O, TiO<sub>2</sub>, P<sub>2</sub>O<sub>5</sub>) and select trace elements (Ba, Sr, Y, Sc, Zr, Be, V, Hg) at Activation Laboratories Ltd. (Actlabs) in Ancaster, ON, Canada. At Actlabs, samples were crushed and pulverized using mild steel before undergoing lithium metaborate/tetraborate fusion followed by HF-HNO<sub>3</sub> dissolution and subsequent analysis by inductively coupled emission-mass spectroscopy (ICP-ES). Separate Hg concentrations were determined via cold vapor flow injection mercury system (Hg-FIMS).

Additional trace elements, including trace metals, low field strength elements (LFSE), rare earth elements (REE), and high field strength elements (HFSE), were analyzed at the Department of Earth Sciences at Memorial University. Base metals (Cu, Zn, Pb), transition metals (Sc, Ti, V, Cr, Mn, Co, Ni), volatile elements (As, Cd, Sn, Sb), and the LFSE (Sr, Ba, Th) were determined by multi-acid dissolution and analyzed by inductively coupled plasma-mass spectrometry (ICP-MS). The whole-rock dissolution

process was a modified version of that described by Jenner (1990) and Longerich et al., (1990) and is described in detail by Lode et al. (in review). The REE and HFSE were determined by means of sodium peroxide (Na<sub>2</sub>O<sub>2</sub>) sinter and analyzed using the same ICP-MS at Memorial University by the following procedure: sintering of a 0.2 g sample aliquot with Na<sub>2</sub>O<sub>2</sub>, dissolution of the sinter cake by addition of deionized water and subsequent REE precipitate formation, separation by centrifuge, and acid dissolution of precipitate prior to ICP-MS analysis. The complete method can be found in Longerich et al., (1990). Precision and accuracy of the various methods have been previously reported in Piercey and Colpron (2009) for Actlabs, Jenner et al. (1990) and Lode et al. (in review) for multi-acid dissolution, and Longerich et al. (1990) for Na<sub>2</sub>O<sub>2</sub> sinter. Representative values are displayed in Table 1 (see Appendix B for complete results).

## 2.6.2 Results

All samples from the Boundary deposit exhibit some degree of hydrothermal alteration, which means most major elements (except Al<sub>2</sub>O<sub>3</sub> and TiO<sub>2</sub>), base metals, volatile metals, and the low field strength elements (LFSE) are considered to have been mobile (Spitz and Darling, 1978; Saeki and Date, 1980; Barrett and MacLean, 1994; Jenner, 1996; Large et al, 2001). The high field strength elements (HFSE) and the REE (except Eu) are often regarded as being immobile, but may become mobile (especially the LREE) during intense hydrothermal alteration (e.g., MacLean, 1988). In extremely altered samples, Y and LREE mobility can be characterized by "fan-shaped" REE patterns (e.g., MacLean, 1988) and anomalous Y/Zr and La/Zr ratios relative to the more robust Nb/Zr and Yb/Zr ratios (Barrett et al., 2008).

Despite the potential for element mobility in intensely altered samples, elements considered immobile are useful in determining primary chemostratigraphic and petrologic attributes such as magmatic affinity or tectonic environments. The mobile elements, in contrast, provide insight into alteration and mineralization processes. Collectively, these data can also be used to undertake quantitative mass balance calculations to understand absolute element mobility.

#### **2.6.3 Affinity Monitors**

Elements such as Zr, Ti, La, Yb, and Nb have been used extensively to determining primary chemostratigraphic and petrologic attributes in altered volcanic terranes due to their immobility during hydrothermal alteration (MacLean and Kranidiotis, 1987; MacLean and Barrett, 1993; Barrett and MacLean, 1994). In a Zr/Ti versus Nb/Y plot (Fig. 2.9a), most samples fall within the rhyolite/dacite field with the exception of two significantly altered lapilli tuff samples. Most samples also exhibit Zr/Y ratios between 2.8 and 4.5, indicating the rocks have transitional affinities (Fig. 2.9b)(Ross and Bedard, 2009). Samples that do not fall within this Zr/Y range have intense chlorite/sericite alteration and plot both above and below the transitional field. This is likely due to the mobilization of Y during intense hydrothermal alteration and is discussed in the Mass Changes section. The Zr/Y ratios and the low Nb/Y ratios (Fig. 2.9c) of most Boundary deposit samples support their formation in an arc environment or from re-melted arc crust (Lentz, 1998; Piercey et al., 2014). Low Zr contents (Zr<200 ppm; Fig. 2.9d) and upper crust–normalized La/Sm ratios of <1 (Fig. 2.9e) are consistent with the formation of felsic rocks from sources less evolved than upper continental crust

(e.g., post-Archean, juvenile crust; Piercey, 2009, 2011). Nearly all samples fall within the FIIIa field of a felsic volcanic discriminant diagram (Fig. 2.9f), indicating that partial melting likely took place at relatively shallow levels in the crust (<10km) (Lesher et al., 1986; Hart et al., 2004).

## **2.6.4 Rare Earth Element Plots**

REE patterns for both the hanging wall and footwall samples are shown in Figure 2.10a-b, and have been normalized to the chondrite values of McDonough and Sun (1995). Both hanging wall and footwall sample patterns display slight LREE-enrichment with flat HREE patterns; however, the hanging wall samples have much tighter patterns with less dispersion in REE abundances. The variation displayed by the footwall rocks (Fig. 2.10b) is due to either mass/volume changes and/or the actual mobility of the REE during hydrothermal alteration. The addition of Si and Fe during alteration results in a net mass gain, which can dilute the overall immobile REE concentration; conversely, a net mass loss (e.g., chlorite or sericite alteration) can result in apparent increase of REE concentrations (Fig. 2.10c)(Barrett et al., 2008). The effects of mass changes should result in a uniform shift; however, the observed LREE contents in highly altered samples have likely been influenced by element mobility, resulting in decreased LREE, without losses in HREE (Fig. 2.10d). The mobility of REE are further investigated in the Mass Change Calculations section below.

## 2.6.5 Mobile Element Lithogeochemistry

Regardless of lithology, all samples from the Boundary deposit have pronounced depletions in Na<sub>2</sub>O (often Na<sub>2</sub>O <0.6%) and high Spitz-Darling index values

(Al<sub>2</sub>O<sub>3</sub>/Na<sub>2</sub>O>40; Fig. 2.11a). High Hashimoto index values

 $[AI=100(MgO+K_2O)/(MgO+K_2O+CaO+Na_2O)]$  (Ishikawa et al., 1976) and chloritecarbonate-pyrite index values  $[CCPI=100(MgO+Fe_2O_3)/(MgO+Fe_2O_3+CaO+Na_2O)]$ (Large et al., 2001) are also characteristic of these samples, with the intense chlorite altered samples plotting near the chlorite-pyrite node and quartz/sericite altered samples plotting near the sericite node (Fig. 2.11b). The three alteration assemblages are most easily differentiated using MgO/Al<sub>2</sub>O<sub>3</sub> and K<sub>2</sub>O/Al<sub>2</sub>O<sub>3</sub> ratios (Fig. 2.11c-d).

Select trace elements also increase towards mineralization (e.g., Large et al., 2001), including Hg/Na<sub>2</sub>O and Ba/Sr which are elevated in all alteration assemblages, with the exception of those that display very intense chlorite and chaotic carbonate alteration (Ba/Sr <10 and Hg/Na<sub>2</sub>O >10) (Fig. 2.11e; Collins, 1989). Increasing Hg/Na<sub>2</sub>O ratios also correlate with increasing Zn contents (Fig. 2.11f), but not Cu contents, and thus likely mirror the edge of high temperature mineralization/alteration.

## 2.6.6 Mass Change Calculations

Although raw lithogeochemical data can illustrate useful alteration trends, the absolute gains and losses of mobile elements yield true alteration dimensions and take into account mass and volume changes during alteration in addition to elemental changes (e.g., Gresens, 1967; Grant, 1986; MacLean, 1990). By plotting pairs of immobile elements against one another, samples from an originally homogeneous protolith will plot along a single line that passes through the origin, whereas those from multiple sources will form additional trend lines (MacLean and Kranidiotis, 1987; Barrett and MacLean, 1991; MacLean and Barrett, 1993). Samples from the Boundary deposit all plot along a

single trend line (Fig. 2.12) indicating that mass changes can be determined by using a single-precursor method (this paper employs the single precursor method of MacLean (1990)). Correlation coefficients calculated for immobile-compatible and immobileincompatible element pairs are reported in Appendix C. The element pair with the highest correlation coefficient between an immobile compatible element (e.g., Al<sub>2</sub>O<sub>3</sub>, TiO<sub>2</sub>) and an immobile incompatible element (e.g., Nb, Zr) was used in determining the mass changes, in this case Al<sub>2</sub>O<sub>3</sub>-Zr. Enrichment factors for each sample were calculated using  $Al_2O_3$  in comparison with the least altered sample. The reconstructed compositions of each sample were then calculated using the enrichment factor, and mass changes for each sample was calculated by subtracting the precursor composition from each of the reconstructed compositions. The least altered precursor composition (Table 2.1-hanging wall sample H501278) was selected based on the sample's low loss on ignition (LOI) and base metal values, minimal Na<sub>2</sub>O depletion, and relatively minor alteration mineral content. The calculated mass changes are shown in Appendix C, plotted in Figure 2.13, and discussed in detail below.

*SiO*<sub>2</sub>: Gains and losses in SiO<sub>2</sub> are the primary causes for mass/volume changes in the Boundary deposit, and they mirror the total mass changes in each sample. Most samples show small to moderate SiO<sub>2</sub> gains, especially those from the footwall lapilli tuff and breccias that exhibit chlorite-sericite alteration. The greatest mass loss is observed in the intense chlorite altered footwall lapilli tuffs (50% loss) and in the chlorite-sericite altered hanging wall flows and breccias (~20% loss)(Fig. 2.13a-b; Appendix C.2). *Na*<sub>2</sub>*O*, *CaO*: The alkali components, Na<sub>2</sub>O and CaO, behave differently from each other. Throughout the Boundary deposit, Na<sub>2</sub>O is nearly completely depleted from all samples (1-2% Na<sub>2</sub>O losses). Only in the least altered samples does Na<sub>2</sub>O appear to be slightly less depleted (0-0.5 Na<sub>2</sub>O% losses). Conversely, no change or only moderate mass gains are observed for CaO within the footwall lapilli tuff and breccias. Calcium gains are present in samples that have undergone carbonate alteration (i.e., contain chaotic carbonate, carbonate spot overprinting) within the chlorite-sericite and intense chlorite assemblages (1-3%; Appendix C.2).

 $K_2O$ : The less altered quartz-sericite assemblage generally displays ~1% K<sub>2</sub>O increase independent of lithology. The chlorite-sericite assemblage displays variable mass changes (most showing either a 1% gain or 1-2% loss) in accordance with the dominant observed matrix mineral (gains=sericite; losses=chlorite). The intense chlorite altered samples all show a 1-3% K<sub>2</sub>O loss (Fig. 2.13a).

*MgO*, *Fe*<sub>2</sub>*O*<sub>3</sub>: MgO gains in the Boundary deposit mirror the presence of chlorite alteration. The quartz-sericite assemblage has very little change ( $\pm$  3%); the chloritesericite assemblage has slight gains in MgO (1-5%), predominantly within the hanging wall breccia and within the footwall lapilli tuff and breccia; and the intense chlorite altered samples exhibit gains of 5-20% of MgO. Because Fe<sub>2</sub>O<sub>3</sub> is not only a major constituent of chlorite but also of the sulfide phases present, Fe<sub>2</sub>O<sub>3</sub> gains range from 5-50%, with the highest gains present in the footwall lapilli tuff (Fig. 13c).

*Transition Metals (Sc, Ti, V, Cr, Mn, Co, Ni):* Overall, all Boundary samples show some degree of transition metal enrichment (Appendix C.2). The greatest

enrichments occur in the footwall lapilli tuff and tuff units. Manganese and Co show significant enrichments within the intense chlorite altered lapilli tuff that underlies the Cu-rich lenses (gains up to 0.8% MnO and ~270 ppm Co). Titanium, V, Cr, and Ni tend to show moderate enrichments (20-50 ppm) in the footwall lapilli tuff, tuff, and breccias within no preferred alteration assemblage. Scandium displays little to no mass change throughout the deposit.

*Base Metals (Cu, Zn, Pb):* Chlorite-sericite and intensely altered chlorite samples within the footwall lapilli tuffs and breccias show significant Cu and Zn gains. Zinc and Pb exhibit gains within the hanging wall flows and breccias as well as the footwall lapilli tuff. Minor depletions in base metal content occur at distances greater than 50 meters from the main ore lenses.

*Other LFSE (Th, Sr, Ba):* No substantial mass changes of Th are observed amongst the Boundary samples. Strontium losses are common throughout the deposit (~10 ppm), with the exception of footwall units below the South ore zones that display no change or slight gains (~10 ppm). Because the least altered sample contains some degree of sericite alteration, Ba contents are higher than would be expected for a completely unaltered sample. The Ba contents of the least altered sample are ~1,400 ppm. Keeping this in mind, all lithologies show significant Ba mass gains and losses ( $\pm$ 1000 ppm). The hanging wall breccias show the highest Ba gains relative to the other lithologies. The footwall flows and breccia exhibit moderate depletions within the chlorite-sericite samples, whereas the quartz-sericite assemblage generally shows moderate to significant enrichments (500-1000 ppm). The intense chlorite samples display significant losses (~1,000 ppm).

*HFSE (Y, Nb, Hf, Ta)*: Amongst the HFSE, Y displays the most significant mass changes throughout the Boundary deposit (Appendix C). Yttrium gains and losses are generally on the order of ~8 ppm in both the hanging wall and footwall flows and breccias. Significant Y gains (~10 ppm) occur proximal to Cu-rich mineralized zones. Hafnium and Nb both exhibit minor mass changes (generally  $\pm 3$  ppm) with slightly greater changes in the more intensely chlorite altered samples (-7 to +10 ppm). Tantalum does not exhibit any significant changes.

*Volatile Elements (As, Cd, Sn, Sb, Hg):* Tin and Sb generally exhibit minor gains and losses (±10 ppm) throughout the Boundary deposit (Appendix C.2). Gains in As (75-100 ppm) and Cd (~20 ppm) are common within the chlorite-sericite samples surrounding Zn ore zones but depletions are present in samples overlying and underlying these Zn-rich zones. Significant Hg gains (>1 ppm) are present in the upper ten meters of the quartz-sericite altered lapilli tuff and in hanging wall units.

*REE (La-Lu):* In general, the REE do not display significant mass changes with the exception of La, Ce, and Nd (Fig. 2.14a), which exhibit exceedingly large gains and losses in a set of 12 samples that are found within footwall lapilli tuff (matrix supported) and tuff samples; however, the gains and losses are not tied to a specific alteration assemblage.

The mobility of REE in VMS deposits has been documented in a number of studies (e.g., MacLean and Kranidiotis, 1987; MacLean, 1988; Barrett and MacLean,

1994; Barrett et al., 2008), and this appears to be the case as well at Boundary, especially in the footwall (Fig. 2.10b, Fig. 2.14a). However, once the effects of volume change are taken into account, the REE values are very similar to those of the least altered sample (Fig. 2.14b-c). All three alteration assemblages show minor LREE depletions and relatively unchanged HREE (Fig. 2.14b). Maclean (1988) attributes sample/precursor ratios between 1-3 to strong REE mobility, whereas the ratios in this study are less than 0.25 for both the LREE and HREE. The intense chlorite assemblage displays the greatest depletions in the LREE, all of which occurs in the footwall lapilli tuff unit. The chloritesericite and quartz-sericite assemblages have similar REE patterns with the exception of Eu, which is liberated and removed as plagioclase breaks down (Sverjensky, 1984). Compared to the footwall (Fig 2.14c), the hanging wall samples show slightly greater REE depletions. However, given that the majority of the hanging wall has been eroded, this is likely a function of proximity to the ore horizon, as all of the hanging wall samples were taken from the bottom ten meters of the unit. The footwall samples display lower depletion levels for two reasons: 1) higher number of distal, less altered samples decrease the effect of the intense chlorite samples; and 2) the LREE removed from the intense chlorite zones were probably redistributed among the less altered zones as evidenced by the precipitation of apatite, monazite, and xenotime, especially within the heavily sericite altered samples (high K<sub>2</sub>O wt.%).

#### 2.6.7 Principal Component Analysis

A principal component analysis (PCA) was carried out on the major and trace element dataset listed in Appendix B. Elements that contained a high percentage of

censored values (i.e., values below or close to detection limit), such as Be and Se, were not included in the analysis due to the false components generated based on censored values. Since the REE and HFSE were deemed essentially immobile, one REE (Lu) and one HFSE (Y) was included in the analysis to represent these groups in order to maximize the correlations between the mobile elements. Lutetium was chosen because it represents the least mobile REE and HFSE elements (from mass change section). Yttruim was chosen because it exhibited the most significant changes amongst the HFSE REE (including the LREE). A subsequent PCA analysis was carried out with the entire dataset, and all REE and HFSE produced similar results to the representative elements choses. Figure 2.15a is an ordered plot of eigenvalues and their relative significance for each component. Determining the number of significant components can be done a number of ways (e.g., "elbow" of the screeplot, components <5%, eigen values<1). In this analysis only the first four components are considered, because any component after four represents <5% of the variance in the dataset and often contains redundant element associations with, only one or two differing elements from one of the four major principal components. Further PCA information is given in Appendix D.

This approach discriminated four components that account for ~55% of the total geochemical variability (Table 2). The first component accounts for 20% of the variation in the Boundary dataset and is reflects the inverse relationship between elements associated with intense chlorite alteration and mineralization (Fe<sub>2</sub>O<sub>3</sub>, LOI, Co, Mo, Cu, W, Bi, MgO, As) and those associated with quartz-sericite alteration (SiO<sub>2</sub>, K<sub>2</sub>O, Ba). The second component accounts for 18% of dataset variance and contains only positive

scores for MgO, Sc, TiO<sub>2</sub>, Al<sub>2</sub>O<sub>3</sub>, Lu, Y, P<sub>2</sub>O<sub>5</sub>, K<sub>2</sub>O, and Ba; these elements represent less intense sericite and chlorite alteration. The lack of scores for Cu, Zn, Pb, Fe<sub>2</sub>O<sub>3</sub>, CaO, MnO, and SiO<sub>2</sub> indicates that, while these elements are indicative of intense alteration/mineralization/silicification, they also occur throughout the deposit in variable amounts. The third component (10% dataset variance), exhibits significant scores for Pb, Zn, Hg, and Sn; which represents that low temperature mineralization is present throughout the deposit regardless of alteration assemblage. The fourth component (7% variance) reflects carbonate alteration (LOI, P<sub>2</sub>O<sub>5</sub>, MnO, CaO, MgO, and Sr) versus quartz-sericite alteration (SiO<sub>2</sub>, K<sub>2</sub>O, and Ba). Figure 2.15b displays components one and two plotted in eigenvector space to illustrate the observed clustering of elemental scores.

# 2.7 Shortwave Infrared (SWIR) Spectroscopy

A total of 237 samples from both the hanging wall flows and breccias and the footwall tuff, lapilli tuff, breccia, and flows were analyzed at Memorial University using a Terraspec<sup>TM</sup> mineral spectrometer with a Hi-Brite Muglight. Minimal sample preparation was needed as SWIR absorption spectra were collected from fresh drill core that had been recently cut and cleaned. Optimization and white references were collected every ~20 analyses or 40 min to avoid significant drifts in data. Analyses were only performed in a naturally sunlit room to avoid any interference from artificial lighting. Reference sample measurements were made after each optimization and white reference to measure both precision and accuracy. Reference sample measurements for pyrophyllite and kaolinite indicate that the Terraspec<sup>TM</sup> was consistently within 1 nm of

the accepted values. Reflectance spectra were collected using the RS<sup>3</sup> Spectral Acquisition software. The resultant spectra were processed using The Spectral Geologist Hotcore v 7.1.55 software in which the hull correction was applied. Hyperspectral scalars were calculated using a fourth order polynomial fitting curve applied over the ranges 2120-2245 nm (focused at 2180-2230 nm) for the AIOH wavelengths and 2230-2270 nm (focused at 2240-2260 nm) for FeOH. Spectral depth filters were applied to remove background noise with filters of 0.015% for AIOH and 0.010% for FeOH. Reported spectra with trough depths less than the filter value were dismissed. The simple mineralogy and absence of weathering products, as determined through petrographic studies and SEM analyses (Appendix E), generally produced diagnostic absorption features and made for clear mineral identification. Due to the variable clast-matrix mineralogy, a minimum of three analyses per sample were made to ensure representative measurements. The complete SWIR absorption data are also reported in Appendix B.

Shortwave infrared (SWIR) spectrometry subjects a sample to a light source and measures the wavelengths of light that are absorbed by certain bonds within minerals (AusSpect International, 2008). Most absorption features are generated by OH, H<sub>2</sub>O, CO<sub>3</sub>, NH<sub>4</sub>, AlOH, FeOH, and MgOH bonds and are represented as minima in the reflectance spectrum over a specified range (AusSpect International, 2008). Two of the most commonly used features in mineral deposit exploration are the AlOH absorption feature between 2190-2225 nm and the FeOH absorption feature between 2,245-2,260 nm which correspond to white mica (sericite) and chlorite compositions at the Boundary deposit, respectively (Fig. 2.16)(AusSpect International, 2008).

Theoretical absorption features for chlorite (FeOH bonds) and white mica (AlOH bonds) are shown in Fig. 2.16. The diagnostic absorption feature for white mica is a deep absorption feature between 2,180-2,228 nm. The exact wavelength at which the absorption feature occurs is directly related to the composition (e.g., Velde, 1978; Besson and Drits, 1997a&b; Guidotti and Sassi, 1998; Herrmann et al., 2001). Shorter wavelengths (2,180-2,195 nm) are characteristic of sodium bearing mica (paragonite), whereas longer wavelengths (2,210-2,228 nm) are typically diagnostic of Fe-Mg mica (phengite) (Herrmann et al., 2001; Yang et al., 2011). Typical potassic mica (muscovite) produces absorption features between 2,200 and 2,204. Intermediate wavelengths are the result of intermediate compositions or multiple white mica phases (Herrmann et al., 2001). Chlorite spectra contain two diagnostic absorption features that represent FeOH (2,235-2,260 nm) and MgOH (2,320-2,360 nm) bonds (Herrmann et al., 2001). Both the FeOH and MgOH bonds wavelengths have been illustrated to increase with iron content (McLeod and Stanton., 1984; Pontual et al, 1997). The MgOH absorption feature occurs over the same range of wavelengths as  $CO_3$  (Fig. 2.16). Due to the varying degree of carbonate spot overprinting in the Boundary samples and overlap with the MgOH range, the measured MgOH wavelengths were not used in this study.

#### 2.7.1 Results

The AlOH wavelengths of white mica within the Boundary deposit range from 2,190-2,213 nm, with the majority of samples falling in the range of 2,200-2,205 (Fig. 2.17a). The FeOH wavelengths range from 2,248-2,257 and display a normal distribution centered at 2,252 nm (Fig. 2.17b). For ease of description, samples that exhibit AlOH

wavelengths  $\leq 2,205$  nm are designated as muscovitic mica, whereas those with wavelengths  $\geq 2,210$  nm are termed phengitic mica. FeOH wavelengths can be used to estimate chlorite compositions (e.g., McLeod and Stanton, 1984; Pontual et al., 1997; Jones et al., 2005), in which longer FeOH wavelengths correlate to Fe-rich chlorite. The FeOH wavelengths from the Boundary deposit rocks do increase with decreasing Mg number (100\*MgO/[MgO+FeO])(Fig. 2.18). Samples that contain higher amounts of Feand Mg-bearing minerals (i.e., pyrite, chalcopyrite, carbonate) plot further from the regression line (Fig. 2.18).

# 2.8 Discussion

#### **2.8.1 Rhyolite Formation and Tectonic Implications**

Rhyolites from the Boundary deposit have FIIIa chemical signatures (Fig. 2.9f). Rocks with FIIIa signatures are interpreted to have formed via partial melting of either continental or oceanic crust due to basaltic underplating during rifting (Lesher et al., 1986; Barrie et al., 1993; Hart et al., 2004; Piercey, 2011). Hart et al. (2004) further argued that FIIIa rhyolites form at shallow depths (<10 km), due to low pressure (<0.5 Gpa) and high temperature (900-1000°C) melting. Melts generated under such conditions typically have negative Ti-Nb anomalies and weakly depleted HREE, Y, and Eu, implying formation at shallow levels and at high temperatures, and these are features present in the rhyolitic rocks of the Boundary deposit (this study and Piercey et al., 2014),

Previous studies have argued that the Tally Pond group formed as the result of continental arc magmatism based on U-Pb geochronology and evolved Nd and Pb isotopic signatures (e.g., Swinden and Thorpe, 1984; Dunning et al., 1991; Pollock, 2004, Rogers et al., 2006; McNicoll et al., 2010). However, rhyolites from the Boundary deposit have signatures that are more akin to those that have formed in post-Archean, juvenile environments (e.g., Figs. 2.9-2.10; Hart et al., 2004; Piercey, 2011; Piercey et al., 2014). These data therefore suggest that the Boundary deposit host rocks could not have formed from solely continental crustal sources and must have had input from sources more juvenile than typical, evolved upper continental crust. Additionally, the Tally Pond group is predominantly bimodal, a characteristic commonly observed in continental rift environments in which felsic rocks are produced by partial melting of overlying crust by basaltic underplating (Barrie et al., 1993; Wright et al., 1996; Smith et al., 2003) as opposed to the full volcanic spectrum (basalt-andesite-dacite-rhyolite) produced by continuous fractionation in continental arc environments (Arculus, 1994; Lentz, 1998; Hart et al., 2004).

Zagorevski et al. (2010) and Piercey et al. (2014) provided tectonic models that incorporate both the juvenile rift and continental arc signatures present within the Tally Pond group, inferring a rifted continental arc on the Iapetus-side of Ganderia. A rift environment would result in crustal thinning and asthenospheric mantle upwelling leading to basaltic underplating of the overlying arc crust (Hart et al., 2004, Zagorevski et al., 2010, Piercey et al., 2014). Shallow (<10 km) partial melting of this crust would result in the production of the bimodal volcanic rocks present at the Boundary deposit (Zagoreveski et al., 2010, Piercey et al., 2014). Furthermore, the partial melting of arc crust would have resulted in rhyolitic melts with hybrid geochemical signatures

containing a juvenile component from the basaltic underplating and a crustal signature from the arc crust.

The rifted arc model also explains the VMS endowment in the Tally Pond group, as arc rifting creates a geodynamic environment favorable to VMS formation (i.e., Franklin et al., 2005; Galley et al., 2007). In particular, as rifting occurs, the thinning of the overlying crust and upwelling of asthenospheric mantle allow for shallow emplacement of heat sources (magma chambers, intrusions, etc.) needed to drive the convection of hydrothermal fluids (e.g., Lesher et al., 1986; Lentz, 1998; Hart et al., 2004; Piercey, 2009, 2011). In addition, rifting increases the fracture permeability and porosity, which creates hydrothermal fluid pathways necessary for constant discharge and recharge of seawater-based hydrothermal fluids (e.g., Franklin et al., 1981, 2005; Lydon, 1984, 1988; Large, 1992; Lentz, 1998; Barrie and Hannington, 1999; Galley et al., 2007; Piercey, 2009, 2011).

## 2.8.2 Characteristics and Controls on Hydrothermal Alteration

Because the Boundary deposit has not undergone any significant metamorphism and has remained mostly structurally intact, most, if not all of the hydrothermal alteration features are extremely well preserved and close to their original distribution. Whereas most VMS deposits only exhibit classic discordant, pipe-like alteration (e.g., Riverin and Hodgson, 1980; Gemmell and Large, 1992; Franklin et al., 2005), Boundary exhibits both discordant alteration and laterally extensive alteration, especially near the ore body. Laterally extensive alteration is interpreted to be a product of upflow rather than semiconformable alteration associated with hydrothermal recharge (e.g., Galley, 1993; Franklin et al., 2005). Its stratigraphic, semi-conformable geometry is a function of unfocused discharge into relatively unconsolidated and porous volcaniclastic rocks, which make up the footwall that were overlain by an impermeable (to semi-permeable), flow-dominated hanging wall (Doyle and Allen, 2003; Gibson, 2005; Franklin et al., 2005). This stratigraphic and hydrothermal architecture was also responsible for the development of the replacement-type mineralization and alteration preserved in the Boundary deposit (Piercey et al., 2014).

The intense depletion of Na<sub>2</sub>O throughout both the hanging wall and footwall (Fig. 2.11a) is due to the destruction of feldspar and zoisite during hydrothermal alteration (e.g., Riverin and Hodgson, 1980; Knuckey et al., 1982; Date et al., 1983; Barrett and MacLean, 1994), which is likely the result of one or more of the following reactions (Riverin and Hodgson, 1980; Barrett and MacLean, 1994):

(1)  $3NaAlSi_3O_8 + 2K^+_{(aq)} + 2H^+_{(aq)} = KAl_3Si_3O_{10}(OH)_2 + 6SiO_{2(aq)} + 3Na^+_{(aq)}$ Albite Muscovite Quartz

(2)  $2Ca_{0.5}Na_{0.5}Al_{1.5}Si_{2.5}O_8 + 4.48Al(OH)_{4^-(aq)} + 1.16K^+(aq) + 2.23H^+(aq) =$ Andesine  $1.16KAl_3Si_3O_{10}(OH)_2 + 1.5SiO_{2(aq)} + Na^+(aq) + Ca^{2+}(aq)$ Muscovite Ouartz

 $\begin{array}{c} \textbf{(3)} Ca_2Al_2Si_3O_{12}(OH) + K^+{}_{(aq)} = KAl_3Si_3O_{10}(OH)_2 + 6SiO_{2(aq)} + 3Na^+{}_{(aq)}\\ Zoisite \qquad Muscovite \qquad Quartz \end{array}$ 

$$\begin{array}{c} \textbf{(4)} 2\text{NaAlSi}_{3}\text{O}_{8} + 5\text{Mg}^{2+}_{(aq)} + 8\text{H}_{2}\text{O} = \text{Mg}_{5}\text{Al}_{2}\text{Si}_{3}(\text{OH})_{8} + 3\text{SiO}_{2(aq)} + 2\text{Na}^{+}_{(aq)} + 2\text{H}^{+}_{(aq)} \\ \text{Albite} & \text{Chlorite} & \text{Quartz} \end{array}$$

These reactions result in the loss of Na (and Ca) and gains in K and Si (and Mg in Eq. 4), which are illustrated by the mass changes for the quartz-sericite zone (Table 3, Fig. 2.13a). This zone does not show significant Mg gains; however, the alteration of albite to chlorite does take place (Fig. 2.6d; Eq. 4). The formation of chlorite in the chlorite-sericite and intense chlorite assemblages, is due to the destruction of (previously formed) white mica (Riverin and Hodgson, 1980; Knuckey et al., 1982):

 $\begin{array}{l} \textbf{(5)} \ 2K(Al_{3}Si_{3}O_{10})(OH)_{2} + 3H_{4}SiO_{4(aq)} + 9Fe^{2+}{}_{(aq)} + 6Mg^{2+}{}_{(aq)} + 18H_{2}O = \\ Muscovite \\ 3Mg_{2}Fe_{3}Al_{2}Si_{3}O_{10}(OH)_{8} + 2K^{+}{}_{(aq)} + 28H^{+}{}_{(aq)} \\ Chlorite \end{array}$ 

(6) 
$$K_2(Fe^{2+}, Mg)Al_3Si_7AlO_{20}(OH)_4 + 9FeCl_{2(aq)} + 16H_2O =$$
  
Phengite  
 $(Fe^{2+}, Mg)_{10}Al_2Si_6Al_2O_{20}(OH)_{16} + H_4SiO_{4(aq)} + 2KCl_{(aq)} + 16HCl_{(aq)}$   
Chlorite

These reactions result in a net gain of Mg and Fe, whereas K ( $\pm$ Si) is lost from the system, which is observed in the mass changes within the chlorite-altered assemblages (Table 3, Fig. 2.13a-b). However, samples within the chlorite-sericite assemblage show varying degrees of K<sub>2</sub>O losses, depending on the abundance of sericite relative to chlorite

in the rocks. Losses of SiO<sub>2</sub> are greatest in the intense chlorite samples, which also exhibit losses of K<sub>2</sub>O (Fig. 2.13a). Despite reactions 5 and 6 accounting for gains of both MgO and Fe<sub>2</sub>O<sub>3</sub>, the gains in these elements are amplified by additional gains of MgO associated with chaotic carbonate alteration, and Fe<sub>2</sub>O<sub>3</sub> associated with pyrite stringers commonly found within the quartz-sericite assemblage (Fig. 2.13c).

A series of 3D models representing the Boundary deposit, including the main lithologic contacts, ore zones, Wagner fault, and key alteration features are shown in Figure 2.19. The mineralized horizon is comprised of the upper  $\sim$ 30m of lapilli tuff and tuff, where the lower footwall contact (Fig. 2.19a) contact marks the first significant (>2m) aphyric flow or flow breccia. Massive pyrite zones are those with >15% assayed Fe and have been cross-checked with drill logs and cross sections. As expected, the CCPI (Fig.2.19b) systematically increases towards zones underlying both the pyrite and Cu-rich zones, reflective of the Mg-Fe-enrichment in these areas. These high Mg-Fe zones are also mirrored by low AlOH and high FeOH SWIR absorption depth values (Fig. 2.19c-d); the higher AlOH and lower FeOH correspond to sericite-rich zones proximal to high Zn-rich zones (Fig. 2.19), a relationship also observed in other VMS deposits (e.g., Jones et al. 2005; Yang et al. 2011). The deepest FeOH absorption features correlate with the highest CCPI values, which underlie the South zone and the Cu-rich areas of the North zone. This not only suggests that these areas have been offset by the Wagner fault, but also that they represent primary, hydrothermal vent locations (i.e., upflow zones)(Fig. 2.19d).
The results above are also reflected in the 3D distribution of mass change results, particularly gains and losses in MgO, K<sub>2</sub>O, and SiO<sub>2</sub>. The interpreted hydrothermal upflow zones contain mass gains in K<sub>2</sub>O and MgO, with >3% MgO gains near the center and ~2% MgO and >0.5% K<sub>2</sub>O gains near the outer edge of the upflow zone; each upflow zone broadly coincides with Cu-enrichment above it and coincident losses in SiO<sub>2</sub> (Fig. 2.19e).

The alteration patterns, and the chemical attributes thereof, as outlined above are controlled by two factors: 1) the host volcanic facies; and 2) the hydrothermal fluid conditions (i.e., temperature, composition, pH). The relatively simple stratigraphy of the Boundary deposit makes documenting the type and intensity of alteration straightforward. Flows and breccias are more crystalline and resistant to hydrothermal alteration, whereas lapilli tuff and tuff units are typically glassy and more susceptible to alteration (e.g., McPhie et al., 1993; Large et al., 2001). This is evident at the Boundary deposit as the flows and breccias are far less altered than the lapilli tuff and tuff units. Furthermore, the porosity and permeability of these units differ due to their fabric. Alteration in the flows and breccias would have been dependent on fracture-controlled porosity and permeability, which include quench fractures or cooling joints (Large et al., 2001). In both footwall and hanging wall breccias (and flow-banded rhyolite flows) only the interfragment regions show alteration, whereas the coherent fragments and massive flows display little to no alteration (Fig. 2.4a-b,e-f), implying fracture controlled fluid flow and alteration. Conversely, the permeability and porosity of the lapilli tuff and tuff was controlled by inter-clast and inter-particle pore space (e.g., McPhie et al., 1993; Large et

al., 2001). The lapilli tuff at the Boundary deposit are generally clast-supported (Fig. 2.4c), which provided an extremely porous matrix consisting of loosely packed ash and/or void space. Consequently, hydrothermal fluids were able to move along numerous paths, rather than just through a single synvolcanic structure, thus resulting in much more widespread and intense fluid-rock interaction and alteration. This relationship has been observed at the Boundary deposit (e.g., Figs. 2.5a-b), and is also recorded as much greater mass changes and the subtle mobility of the LREE of the lapilli tuff relative to the more coherent flows and breccias (Fig. 2.13).

The physicochemical conditions of fluid rock interaction also play a key role in the alteration zoning and chemistry of footwall alteration (e.g., Riverin and Hodgson, 1979; Lydon, 1984, 1988; Gemmell and Large, 1992; Large et al., 2001; Schardt et al., 2001; Franklin et al., 2005; Galley et al., 2007). The well-developed zoning from distal quartz-sericite, to chlorite-sericite, to intense chlorite reflects changing fluid-rock interaction, and fluid temperatures and pH. Schardt et al. (2001) demonstrated that in Cu-Zn systems, the inner, intense chlorite alteration generally forms at higher temperatures (250-350°C) and at moderate pH levels (4.5-5.5). As Mg is fixed within the main upflow zones through the formation of chlorite, the fluids on the periphery of the vent are not only cooler, but become more acidic and form the chlorite-sericite and quartz-sericite alteration assemblages. Increased fluid-rock interactions reduce some of the pH lowering effects of Mg fixation and allow chlorite to form in some areas of the quartz-sericite alteration assemblage (Fig. 2.6d). Additionally, this process explains the absence of chlorite alteration within the hanging wall above the peripheral vent structures. By the time the fluids reached the hanging wall, nearly all the Mg had been fixed within the chloritic pipe leaving only acidic (and cooler) fluids, which created the intense quartz-sericite alteration within the hanging wall.

# 2.8.3 Relationships between Short Wave Infrared Spectroscopy, Alteration Mineralogy and Lithogeochemistry

The SWIR data show clear distinctions among the alteration assemblages (Table 3). Notably, the intense chlorite samples have AlOH absorption wavelengths ~2,196 nm, indicative of sodium bearing white mica (paragonite). The presence of paragonite within the intense alteration pipe is likely the result of a late stage overprint during the waning stages of hydrothermal activity, in which unmodified Na-rich seawater flushed through the hydrothermal system, forming sodic mica (e.g., Date et al., 1983; Green et al., 1983). The chlorite-sericite and quartz-sericite samples exhibit longer wavelengths, the longest of which occur in samples proximal to the Zn-ore zones, indicative of phengite (Fig. 2.20). These samples also contain elevated Zn, Pb, Hg, and Sn (e.g., elements of PCA component 3; Table 2).

The relative proportions of white mica and chlorite is often difficult to determine macroscopically and can be time consuming to determine petrologically. However, in comparison with the whole rock geochemistry, the AlOH depth shows a weak correlation with K<sub>2</sub>O (sericite; Fig. 21a), as does FeOH depth with MgO (chlorite; Fig. 2.21b). Despite the weak correlation, the absorption depth feature (Fig. 21c) does appear to correspond to the relative abundance of sericite and chlorite. The scatter in Figure 2.21,

is attributed to the presence of other K<sub>2</sub>O- and MgO-bearing minerals (e.g., K-feldspar, carbonate), in addition to sericite and chlorite.

Huston et al. (1997) and Herrmann et al. (2001) demonstrated that sericite/chlorite ratios can be estimated semi-quantitatively using the ratio of AlOH to FeOH absorption depths in SWIR spectra (i.e., AlOH/FeOH). The AlOH/FeOH depth ratios in Boundary samples show a distinct correlation with decreasing AlOH/FeOH depths as CCPI increases (Fig. 2.21c). The high CCPI values generally correlate with the intensely altered chlorite pipes that underlie the Cu-rich ore zones, therefore implying that the AlOH/FeOH depth ratio could be used to identify prospective chlorite and Cu-rich alteration zones. In contrast, the best indicator for Zn-rich zones is AlOH wavelength, particularly values greater than 2208 nm (Fig. 2.20). The distribution and variation of the AlOH and FeOH absorption features at the Boundary provide an effective method in tracing fluid pathways and potential vectors to mineralization as summarized in Figure 2.22).

#### 2.8.4 Ore Deposition Mechanism: Exhalative or Replacement

The dominant ore deposition mechanism is difficult to determine at the Boundary deposit. Bedded sulfides along the southern edges of both ore zones provide evidence for exhalative formation, whereas clast-rich sulfides (Fig. 2.7b) and relict hanging wall flow textures indicate replacement processes also took place. The massive sulfides, which can form from either process (e.g., Doyle and Allen, 2003), occur at both the hanging wall-footwall contact and as discrete lenses solely within the lapilli tuff. Although both forms

of mineralization are present, a few key lines of evidence provide insight as to which mechanism was the dominant mineralization process.

- 1) Hanging wall-footwall contact massive sulfides- In non-mineralized zones, the 10 cm, fine-grained tuff layer found at the top of the aphyic lapilli tuff is directly overlain by the hanging wall quartz-phyric flows and breccia. The tuff layer is traceable throughout most of the deposit, including below the massive sulfide lenses at the hanging wall-footwall contact. The hanging wall contact with the ore lenses typically sharp and hanging wall replacement textures absent throughout the lenses except near the top in rare cases. The sulfide lenses found above this tuff layer with sharp hanging wall contacts indicate that they were likely deposited prior to the deposition of the hanging wall. The bedded sulfides that occur directly above the tuff layer, distal to the main alteration pipes also indicate exhalative process and contain drop stones of lapilli tuff.
- 2) Hanging wall alteration/mineralization- As stated previously the hanging wall fabric is not favorable for replacement processes to take place. The limited porosity and permeability inhibited hydrothermal fluid flow and mineralization. However, moderate sericite and minor chlorite alteration, determined petrographically and geochemically (2.11b-d; 2.13a-c), shows that hydrothermal fluids were able to somewhat affect the hanging wall. Furthermore, replacement style mineralization (relict flow textures) is found

within the hanging wall, though it does not extend more than a couple meters and is very uncommon.

3) Footwall alteration/mineralization- The footwall lapilli tuff and tuff unit is far more conducive for hydrothermal fluid flow given the host rock properties. The intensity of hydrothermal alteration, as demonstrated by the geochemistry and mass change calculations, is far greater than in the hanging wall. Although this does not prove replacement mineralization, it does show that the footwall interacted with far more hydrothermal fluids. The main evidence for replacement within the footwall lapilli tuff and tuff unit is the clast-rich massive sulfides, in which the ash tuff matrix has been replaced by sulfides (e.g., Fig. 2.7b). Both the basal portions of the hanging wall-footwall contact massive sulfides and the smaller footwall sulfide lenses are clast-rich, a feature indicative of replacement in the Boundary deposit (Piercey et al., 2014).

The massive sulfides that occur at the hanging wall-footwall and corresponding bedded sulfides were deposited via exhalative processes, whereas the hanging wall sulfides, basal portion of the hanging wall-footwall contact massive sulfides, and footwall lapilli tuff and tuff sulfides were all deposited via replacement processes. Although mineralization by stratigraphic position cannot be calculated, replacement mineralization appears to be the dominant mechanism. The replacement footwall lapilli tuff mineralization accounts for the majority of mineralization at the Boundary deposit

(Appendix A), followed by the exhalative hanging wall-footwall contact sulfides (massive and bedded) and hanging wall replacement mineralization.

#### 2.8.5 The Boundary Replacement Model

Piercey et al. (2014) demonstrated that the Boundary deposit meets the criteria set forth by Doyle and Allen (2003) for replacement-style VMS deposits. However, prior to subseafloor replacement processes, Boundary was initially a seafloor VMS (Fig. 2.23a). The main alteration pipe formed along a single synvolcanic structure within the coherent footwall flows and breccias before entering the highly permeable and porous upper lapilli tuff unit that allowed the formation of multiple vent structures (Fig. 2.23a). Given the size of these vent structures, the primary alteration would have been relatively limited, consisting of a chlorite (reaction 4) or sericite (reaction 1-3) depending on the size of the vent.

As hydrothermal activity progressed, alteration in the main upflow zones broadened (Fig. 2.23b). The main upflow zones became well defined and early sericite alteration was altered to chlorite alteration (e.g., reactions 5 and 6). Smaller offshoots of the main upflow zones would have contained chlorite, but after the precipitation of chlorite and progressive cooling, these smaller zones would have become dominated by sericite. Initially, these fluids were hot (~300°C) and depleted of Mg due to chlorite formation; thus, white mica precipitation would be minimal and muscovitic in composition. As the fluids rose towards the seafloor, they became cooler and Mg-rich through the drawdown of unmodified seawater, which created alteration pipes of increasingly phengitic mica. Precipitation of sulfides occurred primarily at the seafloor

and minor amounts may have formed below as a result of mixing with unmodified seawater (e.g., Doyle and Allen, 2003). Sulfides also precipitated lower in the offshoot upflow zones as these fluids cooled faster than those in the central upflow zones.

Following the deposition of the hanging wall units (~509 Ma; Fig. 2.23c), the internal behavior of the Boundary hydrothermal system would have been greatly altered. The impermeable hanging wall would have forced fluids to move more laterally though the porous lapilli tuff both at the hanging wall contact and within the alteration pipe(s). As the alteration pipes expanded laterally, preexisting sericite was converted to chlorite (reactions 5-6); incomplete conversion of sericite resulted in the formation of the chlorite-sericite zones. Additionally, the lateral fluid flow resulted in the replacement of the lapilli tuff by sulfide mineralization that underlies parts of the massive sulfides (Fig. 2.7b). As Figure 2.11 illustrates, very few hanging wall samples display even moderate chlorite alteration, and those that do occur directly over the main ore lenses. Although hanging wall alteration is evident both visually (i.e., sericite, disseminated pyrite) and geochemically (e.g., SiO<sub>2</sub>, K<sub>2</sub>O mass gains, Na<sub>2</sub>O depletions), the intensity is limited, primarily occurring as quartz-sericite alteration, likely due to the limited permeability of the hanging wall.

The deposition of the hanging wall likely marked the decline of hydrothermal activity at Boundary; however, its life may have been temporarily prolonged during the emplacement of the quartz-porphyry intrusion (Figs. 2.4g and 2.23c). Although only two samples were taken from the intrusion, they display similar immobile element and alteration patterns (e.g., Figs. 2.9a-f, 2.10a, and 2.11a-f) to both the hanging wall and

footwall. It is uncertain as to whether or not the intrusion contributed heat and fluids/metals, especially near the South zone. Further work is required to decipher the relationship between this intrusion and the Boundary alteration and mineralization. As the hydrothermal activity came to an end, Na-bearing seawater was circulated though the main upflow zones, which resulted in the formation of the minute amounts of paragonite identified by SWIR spectroscopy within the intense chlorite and chlorite-sericite samples. (Table 3, Fig. 2.22).

### **2.9 Conclusions**

The main conclusion of this manuscript include:

- The Boundary deposit formed on the leading edge of Ganderia in rhyolitic rocks within a rifted continental arc. This is supported by immobile element geochemistry and is consistent with previous studies within the Tally Pond group.
- 2. The Boundary deposit contains three dominant alteration assemblages: quartz-sericite, chlorite-sericite, and intense chlorite, each which have distinct geochemical and hyperspectral signatures. These alteration assemblages are predominantly controlled by the host rock permeability and porosity in addition to hydrothermal fluid conditions.
- Alteration indices, such as CCPI, and element gains (e.g., Cu, MgO, MnO, Fe<sub>2</sub>O<sub>3</sub>) provide useful vectors for Cu-mineralization; whereas, Hg/Na<sub>2</sub>O and Hg, Zn, Pb, Cd, and Ba increase in proximity to Zn mineralization.

- 4. Mass changes among the major oxides (e.g., SiO<sub>2</sub>, MgO, Fe<sub>2</sub>O<sub>3</sub>, K<sub>2</sub>O), base metals, transition metals (e.g., Ba, Sr) vary with each alteration assemblage.
  Other elements, such as the REEs and high field strength elements, are largely immobile throughout the deposit.
- 5. Hyperspectral data reveal distinct trends that correlate with alteration assemblages and geochemistry: A) AlOH absorption hull depths loosely correspond with the K<sub>2</sub>O wt% and the wavelengths increase proximal to Znrich mineralization; B) FeOH wavelengths correlate with whole rock Mg number and indicate a high abundance of Mg-rich chlorite in the intense chlorite assemblage; and C) the absorption hull depths for both AlOH and FeOH are useful in differentiating between assemblages and the ratio of the two (AlOH depth/FeOH depth) correlates strongly with the CCPI.
- 6. The Boundary deposit formed as a result of both exhalative and replacement mechanisms. The majority of the mineralization, including the clast-rich basal portions of the massive sulfides, footwall mineralization, and the relict flow band mineralization, was formed through replacement processes. Only the massive and bedded mineralization at the hanging wall-footwall contact, which accounts for a lesser portion of mineralization at the Boundary deposit, formed through exhalative processes.

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Sample ID	H501278	H500680	H501213	H501317	H501222	H501217	H500663
Hole ID	BD10-11	BD10-87	BD10-40	BD10-6	BD10-50	BD10-98	BD10-102
Depth (m)	4	17.9	9.5	7.3	9	13.7	25.1
HW or FW	HW	FW	HW	FW	HW	FW	FW
Lithology	Flow	Lapill Tuff	Breccia	Lapilli Tuff	Flow	Lapilli Tuff	Lapilli Tuff
Alteration	Least altered	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl-Ser	Int Chl	Int Chl
SiO <sub>2</sub> (wt %)	73.94	70.79	65.39	65.02	54.83	21.26	52.08
Al <sub>2</sub> O <sub>3</sub>	11.79	11.14	12.23	11.35	13.1	18.1	6.05
Fe <sub>2</sub> O <sub>2</sub> (T)	2.34	8.06	7.94	8.49	13.69	24.73	21.35
MnO	0.095	0.045	0.185	0.164	0.07	0.233	0.073
MgQ	2.46	3	2 41	3 11	4 75	16.23	5.47
(ng0	0.16	0.05	0.07	0.04	0.05	0.09	0.06
No O	1.97	0.05	0.29	0.09	0.05	0.61	0.02
Na <sub>2</sub> 0	1.57	2.06	2.01	0.08	0.22	0.01	0.03
K <sub>2</sub> U	2.29	2.00	5.01	2.50	2.41	0.71	0.12
	0.144	0.169	0.156	0.14	0.162	0.242	0.095
P <sub>2</sub> O <sub>5</sub>	0.01	< 0.01	0.02	0.01	0.01	0.01	0.01
	2.93	5.42	6.85	6.14	9.44	17.88	11.66
Iotal	98.15	100.9	98.54	97.1	98.73	100.1	97
Hg (ppb)	12	11	9	1580	333	68	100
Ba (ppm)	1408	811	1367	1776	1441	391	46
Sr	22	11	12	5	12	34	4
Y	39	43	45	36	46	70	23
Sc	10	10	11	11	13	17	7
Zr	145	126	153	120	158	250	86
Be	1	< 1	1	1	< 1	< 1	< 1
V	< 5	8	7	22	< 5	11	6
Cr	<8.9	<8.9	32.25	12.73	<8.9	<8.9	14.01
Co	1.11	15.34	3.11	18.74	6.04	16.42	133.23
Ni	<10	<10	<10	<10	<10	<10	<10
Cu	83.36	<19	55.96	2252.77	108.98	198.95	16898.55
Zn	150.64	58.08	287.07	5509.07	151.30	1109.40	167.78
As	2.12	1.96	9.96	183.87	17.89	79.21	31.47
Se	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	61.22
Br	176.10	247.70	257.37	267.29	142.76	240.93	162.60
Mo	<1	18.77	8.96	18.90	7.57	3.10	27.15
Ag	<1	<1	<1	9.78	<1	<1	2.13
Cd	<0.9	<0.9	<0.9	15.47	<0.9	3.14	<0.9
Sn	5.61	4.11	3.92	13.47	4.47	3.46	4.25
Sb	0.82	0.64	1.39	25.57	7.16	1.29	2.33
Te	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	45.98
1	<37	<37	<37	<37	<37	<37	<37
W	1.32	6.89	3.41	3.32	1.71	6.01	15.59
Pb	<25.3	<25.3	<25.3	1331.20	<25.3	<25.3	<25.3
Bi	0.30	2.09	1.34	9.38	5.33	4.02	44.03
Nb	6.07	6.03	7.58	5.04	8.10	15.55	3.00
La	20.65	33.51	21.14	16.75	19.53	27.67	9.32
Ce	43.88	71.84	43.82	34.47	40.85	55.42	19.28
Pr	5.49	9.16	5.50	4.35	5.35	6.86	2.41
Nd	23.07	40.38	23.26	18.36	22.78	29.77	10.36
Sm	5.55	9.22	5.97	4.48	5.62	7.36	2.36
Eu	1.19	1.10	1.15	1.41	1.12	1.19	0.25
Tb	1.04	1.29	1.22	0.87	1.20	1.73	0.55
Dy	6.72	8.20	8.12	5.80	8.00	11.70	3.75
Ho	1.48	1.69	1.69	1.25	1.61	2.45	0.83
Er	4.53	4.87	5.20	3.92	5.05	7.50	2.53
Tm	0.68	0.70	0.75	0.58	0.74	1.13	0.38
Yb	4.80	4.49	5.29	4.04	4.91	7.87	2.65
Lu	0.74	0.65	0.80	0.62	0.79	1.20	0.40
Hf	4.70	3.04	5.15	3.77	6.26	7.09	2.03
Та	0.25	0.22	0.32	0.21	0.31	0.58	0.13
Th	5.59	4.62	5.87	4.18	6.37	12.23	2.68
AlOH WL (nm)	2,213	2,198	2,202	2,209	2,196	2,190	NA
AlOH Depth (%)	14.85%	13.28%	7.52%	10.41%	2.16%	3.08%	NA
FeOH WL (nm)	2,252	2,254	NA	2,253	2,252	2,252	2,251
FeOH Depth (%)	2.42%	6.02%	NA	3.39%	2.94%	4.24%	2.45%
AIOH depth/FeOH depth	6.14	2.21	NA	3.07	0.73	0.73	NA

Table 2.1 Representative Whole-rock Analyses for Boundary Samples

Abbreviations: HW=Hanging wall; FW=Footwal; Qtz-Ser= Quartz-sericite; ChI-Ser=ChIorite-sericite; Int ChI= Intense chIorite; WL=wavelength

Component	Variance	Positive Scores	Negative Scores
1	20%	Fe <sub>2</sub> O <sub>3</sub> , LOI, Co, Mo, Cu, W, Bi, MgO, As	SiO <sub>2</sub> , K <sub>2</sub> O, Ba
2	18%	MgO, Sc, TiO <sub>2</sub> , Al <sub>2</sub> O <sub>3</sub> , Lu, Y, P <sub>2</sub> O <sub>5</sub> , MnO, CaO, Sr, V, K <sub>2</sub> O, Ba	
3	10%	Pb, Zn, Hg, Sn	
4	7%	LOI, P <sub>2</sub> O <sub>5</sub> , MnO, CaO, MgO, Sr	SiO <sub>2</sub> , K <sub>2</sub> O, Ba

 Table 2.2 Principal Component Analysis Components

Average Values	Quartz-Sericite	Chlorite-Sericite	Intense Chlorite
SiO <sub>2</sub> (wt. %)	72.11	63.1	40.11
$\Delta SiO_2$ (wt. %)	3.56	2.39	-31.13
K <sub>2</sub> O(wt. %)	2.96	1.98	0.36
Δ K <sub>2</sub> O (wt. %)	0.77	-0.08	-1.99
MgO (wt. %)	1.95	3.98	13.01
∆ MgO (wt. %)	-0.47	1.99	9.82
Ba/Sr	157	140	32
Hg/Na <sub>2</sub> O	1248	2152	5624
ССРІ	66	86	98
AI	94	95	97
AlOH Wavelength (nm)	2204	2202	2196
AlOH Depth	0.12	0.07	0.04
FeOH Wavelength (nm)	2253	2252	2250
FeOH Depth	0.02	0.04	0.06
AlOH Depth/FeOH Depth	6.83	2.73	0.59

**Table 2.3 Alteration Characteristics** 



Figure 2.1 Geological map of the Newfoundland Appalachians with tectonostratigraphic zones, accretionary tracts, and associated belts. Tectonostratigraphic divisions modified from van Stall (2007), and van Staal and Barr (2012). Abbreviations: BBL=Baie Verte Brompton Line; DF=Dover Fault; DBL=Dog Bay Line; GBF=Green Bay fault; LCF=Lobster Cove fault; LRF=Lloyds River fault; RIL=Red Indian Line.



Figure 2.2 Geological setting of the Victoria Lake supergroup and the Tally Pond group, displaying the Boundary deposit and the neighboring Duck Pond and Lemarchant deposits. Modified from McNicoll et al. (2010) and Piercey et al. (2014).



Figure 2.3 Boundary deposit surface geology plan map with distribution of sulfide mineralization and locations of sections in Figure 5a-b. Modified from Hennessey (2013) (Unpublished).



Figure 2.4 Lithologies of the Boundary deposit. A) Hanging wall flow-banded rhyolite. B) Hanging wall hyaloclastite breccia. C) Footwall lapilli tuff. D) Footwall tuff. E) Footwall jig-saw breccia. F) Footwall flow-banded rhyolite. G) Quartz-porphyry intrusion.







Figure 2.6 Photographs and photomicrographs of quartz-sericite alteration. A) Gray, relatively unaltered lapilli tuff with tan carbonate spots. B) Black, quartz-sericite altered lapilli tuff with dark red carbonate spots. C) Photomicrograph of quartz altered clasts with sericite replaced matrix (cross-polarized light). D) Photomicrograph of relict feldspar grain altered to fine grained sericite and chlorite pockets (cross-polarized light). E) Photomicrograph of chalcopyrite disease in sphalerite (reflected light). F) Scanning electron microscope back scatter electron (SEM-BSE) image of anatase and minor monazite in quartz-sericite matrix. Ser = sericite; Qtz = quartz; Chl = chlorite; Kspar = potassium feldspar; Py = pyrite, Sph = sphalerite; Ccp = chalcopyrite; Ana = anatase (rutile); Mon = monazite.



Figure 2.7 Photographs and photomicrographs of chlorite-sericite alteration. A) Gray, chlorite-sericite altered lapilli tuff. B) Zoned lapilli clasts in chlorite-sericite-chalcopyrite matrix. C) Photomicrograph of chlorite-sericite matrix (cross-polarized light). D) Photomicrograph of chlorite-sericite matrix with Fe-Mg-Mn carbonate spot (cross-polarized light). E) Photomicrograph of fractured pyrite with chalcopyrite infill (reflected light). F) SEM-BSE image of Mn-Mg-Fe zoned carbonate spot. Carb = carbonate.



Figure 2.8 Photographs and photomicrographs of intense chlorite alteration. A) Intense chlorite alteration with chaotic carbonate alteration. B) Intense chlorite alteration with carbonate spots. C) Photomicrograph of chaotic carbonate in chlorite-quartz matrix (cross-polarized light). D) Photomicrograph of chlorite replaced lapilli clasts in coarse grained chlorite matrix (cross-polarized light). E) Photomicrograph of fractured pyrite with chalcopyrite infill (reflected light). F) SEM-BSE image of chalcopyrite and Bi-Te sulfide in chlorite matrix.



Figure 2.9 Immobile element plots for rhyolitic rocks from the Boundary VMS deposit. A) Modified Winchester and Floyd (1977) Zr/TiO<sub>2</sub>-Nb/Y discrimination diagram for rock classification (from Pearce, 1996). B) Zr-Y diagram for discriminating magma affinity (from Ross and Bedard, 2009). C) Nb-Y tectonic discrimination diagram (from Pearce et al., 1984). D) Zr-Nb diagram useful for discriminating juvenile environments from evolved environments (modified from Piercey, 2009). E) Upper-crust normalized (UCN) La-Sm diagram (upper-crust values from Taylor and McLennan, 1985; from Piercey and Colpron, 2009). F) La/Ybcn-Ybcn FI-FIV rhyolite discrimination diagram (chondrite normalized values (CN) McDonough and Sun, 1995; diagrams from Lesher et al., 1986; and Hart et al., 2004).



Figure 2.10 REE plots of the Boundary deposit (chondrite-normalized (CN) to the values of McDonough and Sun, 1995). A) Hanging wall. B) Footwall. C) Mass change effects on REE signature. D) REE mobility effects on signature.


Figure 2.11 Mobile element plots for rhyolitic rocks from the Boundary VMS deposit. A) Spitz-Darling (Spitz and Darling, 1978) index versus Na<sub>2</sub>O (diagram after Ruks et al., 2006). B) Alteration box plot with the Hashimoto alteration index (AI; Ishikawa et al., 1976) plotted against the chlorite-carbonate-pyrite index (CCPI) (CCPI and diagram from Large et al. 2001). C) Diagram of MgO plotted against  $Al_2O_3$ . D) Diagram of K<sub>2</sub>O plotted against  $Al_2O_3$ . E) Hg/Na<sub>2</sub>O-Ba/Sr plot of the "Duck Pond index" (modified from Collins, 1989). F) Hg/Na<sub>2</sub>O plotted against Zn.



Figure 2.12 Linear immobile trend line for the Boundary deposit. Mass loss and gain of mobile elements have shifted the relative amounts of  $Al_2O_3$  and Zr in the shown directions.



Figure 2.13 Mass change plots showing gains and losses of key alteration elements. A) Mass change of  $K_2O$  versus SiO<sub>2</sub> showing the formation of quartz and sericite. B) Mass change plot of Fe<sub>2</sub>O<sub>3</sub>+MgO versus SiO<sub>2</sub> showing the formation of quartz, chlorite, and pyrite. C) Mass change plot of MgO vs Fe<sub>2</sub>O<sub>3</sub> differentiating between the formation of pyrite, chlorite, and carbonate. Chl=chlorite, Py=pyrite, Qtz=quartz, Carb=carbonate



Figure 2.14 REE mass balance changes. Concentrations are compared to that of a least altered sample (H501278; Table 2.1). A) Absolute mass gains and losses in ppm. B) Plot of average reconstructed REE values/precursor values by alteration assemblage. C) Plot of average reconstructed REE values/precursor values by hanging wall and footwall.



Figure 2.15 Principle component analysis results. A) Screeplot showing the significance of each component. B) Component 1 plotted against Component 2.



Figure 2.16 Typical SWIR absorption features of white mica and chlorite highlighted in blue with the ranges of other key hydroxyl groups shaded in gray. Figure modified from Jones et al. (2005) and Herrmann et al. (2001).



Figure 2.17 SWIR results. A) Histogram of the AlOH wavelengths present at Boundary. B) Histogram of FeOH wavelengths.



Figure 2.18 Plot of whole-rock Mg number (100\*MgO/(MgO+Fe<sub>2</sub>O<sub>3</sub>) versus FeOH wavelength.



Figure 2.19 Series of 3D models of the Boundary deposit. A) Model showing key lithological contacts, major sulfide zones, and drill holes used to determine alteration extents. Ore zones are defined by assays from additional drill holes not shown. B) Model showing CCPI values in relationship to major ore zones. C) Model showing AlOH wavelengths. Note that wavelengths are significantly longer in the South zone. Long wavelengths in the North zone only occur on the western end of the deposit. D) Model showing FeOH absorption depth. The higher values occur within the same samples of the high CCPI samples from Figure 19B. E) Alteration model of the Boundary deposit based on mass balance changes. The major chlorite gains correlate with the high CCPI values (2.19B) and FeOH absorption values (2.19D).



Figure 2.20 Plot of sample depth vs. AlOH wavelength of samples from the chlorite-sericite and quartz-sericite zones.



Figure 2.21 SWIR and geochemistry. A) Plot of  $K_2O$  versus AlOH absorption depth. B) Plot of MgO versus FeOH absorption depth. C) Plot of the AlOH absorption depth/FeOH absorption depth versus CCPI.



formation (MgO, Fe<sub>2</sub>O<sub>3</sub>) and mineralization (Cu, Zn, Hg,Fe<sub>2</sub>O<sub>3</sub>) and hyperspectral signatures indicating abundant Mg-chlorite content (shorter FeOH wavelength and increasing absorption depth). The offshoot of the main upflow zone exhibits gains in low temperature minrealization (Zn, Pb, Hg), minor chlorite (MgO, Fe<sub>2</sub>O<sub>3</sub>), and increasing sericite (K<sub>2</sub>O,Ba). The hyperspectral signatures indicate an alteration distribution is primarily controlled by the host rock permeability and porosity, the SWIR variation is heavily dependent on the addition of unmodified seawater, temperature, and pH. The central upflow zone exhibits gains in elements related to chlorite increase in phengitic white mica (longer AlOH wavelengths and increasing absorption depth) near the top of the lapilli tuff unit. Figure 2.22 Schematic diagram showing the SWIR data variation and mass changes with geologic explanations. Although the



Figure 2.23 Schematic diagram showing the evolution of the Boundary deposit (see text for details; not to scale). Light green=moderate chlorite alteration; Dark green=intense chlorite alteration; Light blue=sericite alteration, Dark blue=intense sericite alteration. Orange=Cu mineralization; Purple=Zn mineralization; Blue arrows=hydrothermal fluids (size relative to fluid abundance); black arrows=unmodified seawater.

### **Chapter 3: Summary and Future Research**

#### **3.1 Summary**

The bimodal felsic Zn-Cu-Pb Boundary VMS deposit is hosted within the Cambrian Tally Pond group Newfoundland, Canada, and is an example of a replacementstyle VMS deposit. The lack of metamorphic overprint and structural integrity make this deposit an ideal location to study the distribution of mineralization and alteration in subseafloor replacement-type VMS deposits. Furthermore, geochemical and hyperspectral data provide insight into the genesis of the Boundary deposit and its hydrothermal alteration system. The major conclusions from this study are as follows:

- The Boundary deposit formed on the leading edge of Ganderia in rhyolitic host rocks within a rifted continental arc. This is supported by immobile element geochemistry and is consistent with previous studies within the Tally Pond group.
- 2. The Boundary deposit contains three alteration assemblages: quartz-sericite, chlorite-sericite, and intense chlorite, each which have distinct geochemical and hyperspectral signatures. These alteration assemblages are predominantly controlled by the host rock permeability and porosity in addition to hydrothermal fluid conditions.
- Alteration indices, such as CCPI, and element gains (e.g., Cu, MgO, MnO, Fe<sub>2</sub>O<sub>3</sub>) provide useful vectors for Cu-mineralization; whereas, Hg/Na<sub>2</sub>O and Hg, Zn, Pb, Cd, and Ba increase in proximity to Zn mineralization.

- Mass changes among the major oxides (e.g., SiO<sub>2</sub>, MgO, Fe<sub>2</sub>O<sub>3</sub>, K<sub>2</sub>O), base metals, transition metals (e.g., Ba, Sr) vary with each alteration assemblage. Other elements, such as the REEs and high field strength elements, are predominantly immobile throughout the deposit.
- 5. Hyperspectral data reveal distinct trends that correlate with alteration assemblages and geochemistry: A) AlOH absorption hull depths correspond with the K<sub>2</sub>O wt% and the wavelengths increase proximal to Zn-rich mineralization; B) the FeOH wavelengths correlate with whole rock Mg number and indicate a high abundance of Mg-rich chlorite in the intense chlorite assemblage; and C) the absorption hull depths for both AlOH and FeOH are useful in differentiating between assemblages and the ratio of the two (AlOH depth/FeOH depth) correlates strongly with the CCPI.
- 6. The evolution of the Boundary deposit led to the unique replacement features present in both the hanging wall and footwall. The formation of Boundary took place in three main stages: 1) initial faulting: formation of the main upflow zones for hydrothermal fluids to travel through. Lower footwall lapilli tuff and flow units were more resilient than the upper lapilli tuff, which was more heavily fractured creating numerous fluid pathways; 2) initiation of hydrothermal activity: deposition of typical VMS mineralization (i.e., massive Py-Ccp-Sph, bedded sulfides). Some of the smaller upflow zones cooled faster are dominated by sericite alteration and typically underlay Zn-mineralization; and 3) deposition of the hanging wall: fluids were forced to flow laterally

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beneath the hanging wall causing replacement of tuff matrix with sulfide mineralization. This also led to moderate hanging wall alteration (generally chlorite-sericite or quartz-sericite) and some restricted hanging wall mineralization.

### **3.2 Future Research**

Although the Boundary and Duck Pond mine site is now entering the reclamation phase, there are a number of unanswered questions that would greatly benefit the understanding of the Boundary deposit. Potential areas of interest for future research include: 1) the carbonate distribution, both within the Boundary deposit and throughout Tally Pond group, to test if the carbonate compositions vary in proximity to known mineralization; 2) a detailed study of the mineralization that includes the use of sulfur and lead isotopes to determine the main sources of metals and fluids in the deposit; 3) detailed mineralogy, mineral chemistry, stable isotope, and physicochemical modeling of the sulfides and associated hydrothermal fluids to determine fluid and metal origins, temperatures, and conditions of formation; 4) an electron probe microanalysis (EPMA) study of the sericite and chlorite at Boundary to compare compositions with Terraspec<sup>TM</sup> measurements and determine more precisely how wavelengths are related to composition.

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### **Appendix A: Graphic Logs**

### **A.1-Graphic Logs**

Preliminary fieldwork at the Boundary deposit consisted of detailed logging and sampling of the hanging wall, ore horizon, and upper 100 m of the footwall in diamond drill core. Drill core logs focused predominantly on lithology, alteration assemblages, and general mineralization at the Boundary deposit. Fieldwork took place during August-September 2013 and June 2014 at the Duck Pond mine site just outside of Millertown, Newfoundland. A total of 62 drill holes were logged and 237 samples (from 51 drill holes) were taken. A total of 45 thin sections were made from these samples.

Below is the Abbreviation Key and Legend (Appendix A.2) for the 62 drill hole logs (Appendix A.3). Drill holes are labeled using the following nomenclature: BDXX-YYY, where BD stands for Boundary Deposit, XX stands for the last two numbers in the year the hole was drilled, and YYY represents the hole number drilled in that year's drilling program (i.e. BD10-45 was drilled at the Boundary deposit in 2010 and is the 45<sup>th</sup> hole drilled that year). As most holes are relatively short and vertical, these logs represent the true depth with the exception of a few longer holes and those drilled at dips other than 90°.

The logs contain the field interpretation of alteration. In some locations, this is not the true alteration determined through geochemistry and petrography, in which case a note has been made in the description section. Because carbonate spot alteration is present throughout the deposit, only non-carbonate spot alteration (e.g., veins, chaotic carbonate) is represented by the carbonate alteration field. Spot features are sometimes noted in the description if needed. Sulfide occurrences have their own lithology and symbols. Only disseminated sulfide amounts are represented in the alteration field.

### A.2-Abbreviation Key and Legend for Graphic Logs

General			
Е	Easting		
m	Meter		
Ν	Northing		
UTM	Universal Transverse Mercator		
Alteration Types			
Carb	Carbonate		
Chl	Chlorite		
Qtz	Quartz		
Ser	Sericite		
Sulf	Sulfide		

#### **Table A.2.1-Abbreviation Key for Graphic Logs**

Host Rock Facies				
Arg	Argillite			
BX	Breccia			
Flow	Volcanic flow			
Int	Intrusion			
Lap	Lapilli tuff			
MS	Massive sulfide			
Tuff	Tuff			
Descriptions				
alt.	Alteration			
CC	Chaotic carbonate			
Сср	Chalcopyrite			
Dia.	Diameter			
Dis.	Disseminated			
Fe	Iron			
frag(s)	Fragment(s)			
FW	Footwall			
HW	Hanging wall			
Hyalo	Hyaloclastite			
int. chl	Intense chlorite			
mod.	Moderate			
Pheno(s)	Phenocryst(s)			
Ро	Pyrrhotite			
Ру	Pyrite			
Sph	Sphalerite			
Zn	Zinc			





## **Mineralization**

Stringer Sulfides **Replacement Sulfides** 

**Massive Sulfides** 

## Alteration

 Weak
 Moderate
 Intense



## **Additional Symbols**

Sample ~~~~ Fault

Projec	t: Boundary		UTM	Azimuth: 339
Diamond Drill Hole: BD07-199		540816.00E	Dip: -90	
Date: 8-15-2013 5389		5389053.00N	Length: 30.1 m	
	ALTERATION	FACIES	>	
DEPTH (m)	Ser Otz Ser Carb	- Ary - Lap	- Ho - MS	DESCRIPTION
5 <u> </u>	10.6r			
_	i 12.2 r			Dark gray, qtz-phyric hyaloclastite with slight chl alteration near contact.
15—	14.1r 14.9r 14.9r			LT and tuff layers are relatively unaltered. Except the 17.4-18.0 m interval where tuff is intensely chl altered. LT units are all similar light-medium gray color. Clasts are more so matrix supported in the upper two LT units but become clasts runnerted in the third.
20—				unit. Clasts also increase in size from ~1 cm in diameter in the top two and become bimodal in the third (2-4 cm and <0.5 cm). May be potential fault/shear ~17 m. Dis. Py is only sulfide in hole.
25 —				
30—	   . 30.1 r			
35—				
40				
45—				
50—				
55				
60 —				
65 —				

# A.3-Compilation of Graphic Logs






















Project: Bou	ndary		UTM	Azimuth: 360
Diamond Drill Hole: BD10-028			540558.80E	Dip: -90
Date: 8-16-2	2013		5389497.86N	Length: 35
A	LTERATION	FACIES	\$	
DEPTH (m) ₿   3	I arb	— Arry — Arry — BX	- Int MS	DESCRIPTION
DEPTH       N       N         5       1         5       1         10       1         10       1         10       1         20       1         20       1         30       1         30       1         35       1         40       1         50       1         50       1         55       1	LTERATION R         S         R         K <t< td=""><td>FACIES X8 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0</td><td></td><td>DESCRIPTION Sample: H501247 Typical LT footwall all the way through with minor graded tuff beds (&lt;100m). LT clasts are bimodal in size (1cm and 3cm) and varying shades of gray. Near top of hole, LT is very much clast supported but become matrix supported further down in hole. Over the last five meters of the hole, the matrix contains black chl/ser, dis. Py, and minor Fe carbonate veins. Clasts over this stretch also have white rims with grey cores. Sample: H501248 Sample: H501249</td></t<>	FACIES X8 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0		DESCRIPTION Sample: H501247 Typical LT footwall all the way through with minor graded tuff beds (<100m). LT clasts are bimodal in size (1cm and 3cm) and varying shades of gray. Near top of hole, LT is very much clast supported but become matrix supported further down in hole. Over the last five meters of the hole, the matrix contains black chl/ser, dis. Py, and minor Fe carbonate veins. Clasts over this stretch also have white rims with grey cores. Sample: H501248 Sample: H501249
60				
65				























Projec	t: Boundary	UTM	Azimuth: 360	
Diamo	ond Drill Hole: BD10-0	61	540630.90E	Dip: -90
Date: 8	3-14-2013		5569572.741	Length: 50 m
	ALTERATION	FACIES	3	
DEPTH (m)	Qtz Ser Chl Sulf	- Ar - Lap - BX		DESCRIPTION
-	4.5 m			
5 —				LT is clast supported with most clasts ~1 cm. Clasts are light gray, with slightly darker cores. Disseminated Py occurs throughout
-				within matrix.
10—				Sample: H500698
-				
15 —				
-				LT is clasts are darker (dark gray-black). Matrix consists of black chlorite and ser? LT is
20—				slightly more matrix supported than upper portion.
_				
25				Sample: H500699
25				
30-				Lower five meters of LT consist of dark
-				gray-black clasts >1cm. The LT is clast supported (little-no matrix visible: chl altered). Dis. Py found throughout.
35—				
-	38.5 m	0000000000		Sample: H500700
40—	41.0 m			Sample: H501201
-				
45—				Sample: H501202 The BX from 38.5-41m is aphyric, dark gray, with boavily cilicifod fragmonts
_		YYYYY		Interfragment space is typically chl-ser altered and contains minor Ccp-Py. Some
50—	50.0 m	YYYYYY		BX is lighter gray with mm-cm scale fractures commonly filled with milky carbonate
				(usually dolomite?).
55				
55				
-				
60 —				
-				
65 —				
-				



Project: Boundary					UTM		Azimuth: 360		
Diamond Drill Hole: BD-063			540	540606.93E		Dip: -90			
Date: 8-13-2013					538	5389608.09N		Length: 77 m	
	ALTERATION			FAC	IES				
DEPTH	법 . ㅎ. ㄷ. 옆. 놀.	- Ara	-Tuff	-Lap	-BX	-Flow	-Int	-MS	
(m)			~1.7	5					DESCRIPTION
-	77.0 m		XX	XI	K				Sample: H500697
80									
_									
-									
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Projec	t: Boundary		UTM	Azimuth: 360
Diamond Drill Hole: BD10-076			540695.19E	Dip: -90
Date. 8-13-2015			2209002.2211	Length: 50 m
	ALTERATION	FACIES		
DEPTH (m)	Qtz Ser Chl Carb Sulf			DESCRIPTION
5 — 10 — 10 — 15 — 20 — 25 — 30 — 35 — 40 — 40 — 45 — 50 — 55 — 60 — 65 — 61 —	5.8 m			Sample: H500690 LT is extremely clast supported. Subrounded clasts are varying shades of gray and all are ~1 cm in diameter. Most clasts are fairly hard (grat ally whereas darker ones are somewhat (grat ally whereas darker ones are somewhat softer (ser). Deeper in unit (i. e. near 20 m), clasts become larger (up to 6 cm) and contain some flow banded frags. Matrix becomes more prevalent and is chl replaced. Clasts (1-2 cm) are nearly completely altered by dark sericite or chlorite. Stringer Py occurs intermittently throughout over 20-35m stretch. This stretch is also more so matrix supported and contains seemingly high amount of carbonate spotting. Sample: H500689 After stringer section, unit is much like upper portion (all ~1cm in diameter and little to no matrix). Dis. Py scattered throughout. Lowest section in hole doesn't look all that ch altered, but later geochem suggests it is actually significantly altered (CCPI=91). Sample: H500687



Projec	t: Boundary		UTM	Azimuth: 360
Diamond Drill Hole: BD10-078			540667.95E	Dip: -90
Date: 0-13-2014			5389640.13N	Length: 92 m
	ALTERATION	FACIES	3	
DEPTH (m)	Qtz Ser Carb Sulf	— Arr — Lap — BX	- Int MS	DESCRIPTION
-	2.8 m			
5 —				
-				Black-gray hyaloclastite. Fragments space
10		Richard		filled with black sericite. Very minor, minor chlorite even visible. Dis. Py throughout but not much.
-				
15 —				
-	18.3 m			Thin ~20 cm interval of massive Py. Very crumbled core. Appears to be faulted at
20—				Upper ~1 m is matrix supported, dark
-				gray-black clasts (~2 cm). Very little dis. Py within matrix space.
25—				LT becomes more clast supported deeper in the hole. After 35 m, no visible matrix.
-				Clasts are all ~1 cm.
30—		000000000000000000000000000000000000000		
-				
35—				
_				
40				
_				
45				
_	46.6 m	YYYYY		
50				Jigsaw fit BX is dark gray in color. Very minor sulfide (Py). Small patches of chl
_				found within fractures. Some frags are flow banded.
55-		YYYYY		
	57.0 m			
60-				IT very similar to IT above. Unit is close
				supported and light to medium gray. Matrix is dark gray tuff-arg in some areas.
65-				major.













Project: Boundary			UTM	Azimuth: 360
Diamond Drill Hole: BD10-104			540787.00E	Dip: -90
Date: 8-11-2013			5389648.00N	Length: 26.4
	ALTERATION	FACIES	≥	
DEPTH (m)	Qtz Ser Cchl Sulf	— Ar — Lap — BX	- Int MS	DESCRIPTION
DEPTH 5 — 10 — 15 — 20 — 25 — 30 — 35 — 40 — 45 —	ALTERATION			DESCRIPTION UT is gray/green with zoned clasts. Clasts are bimodal in size (<1cm and ~2 cm). Generally clasts supported and clasts grade into tuff near 18 m. Minor dis. Py. Sample: H500642 Description: This section contains dats that have a much wider range of clasts from <0.5 up to 4 cm with no size preference.
50—				
_				
55				
-				
60 —				
-				
65 —				
-				



Project: Boundary					UTM		Azimuth: 360
Diamond Drill Hole: BD10-106					540760.00E		Dip: -90
Date:9-5-2013					538967	8.00N	Length: 90.4 m
	ALTERATION	_	FA	CIES	>		
DEPTH (m)	traine in the state of the stat	– Arç	-Tuff -Lap	i BX	Int	MS	DESCRIPTION
(,							
-							
75—							
-							
80—							
_							
85 —							
_							
90	     00.4m						
50	90.4 m				-		
95 —							
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-							
-							
-							
_							
_							


Projec	t: Boundary		UTM	Azimuth: 360
Diamo	ond Drill Hole: BD10-1	08	540827.00E	Dip: -90
Date: 8	3-17-2013		5589600.00M	Length: 60 m
	ALTERATION	FACIES	2	
DEPTH (m)	Qtz   Chi   Sulf			DESCRIPTION
5 — 10 —	6.4 m			MS is mostly Py with short intervals of increased Ccp (~7-8m and ~15-18 m). Small intervals of LT clasts within sulfides. Clasts content increases near base of massive sulfide.
15 — 20 — 25 — 30 —	20.2 m			Sample: H500621 Matrix sulfides are mostly Py with minor blebs of Ccp. LT clasts are matrix support (Py and black chl), dark gray, subangular, and most are <0.8 cm in diameter. Minor carbonate spotting. Sulfide matrix decreases after 25 m and is dominated by chl. Some clasts have white alteration rims. Sample: H500622
35—	32.8 m		I	Sample: H500623 Flow and BX are both dark gray with dark ser/ch1 filled fractures. Minor dis. Ccp. Sample: H500624 UT is clast supported with dark gray clasts (~2-3 cm) in a Ccp-ch1 matrix. LT below sulfides are highly silicified.
40	40.9 m			BX very similar to BX/flow above.
-	42.8 m			Sample: H500625
45	50.0 m		1	Sample: H500626 Sample: H500627 Sample: H500628
55 — 60 —	51.1 m		1	Sample: H500628 LT (43.6-46.8 m): LT is matrix supported (chl). Clasts are generally medium to dark gray, sub angular, and between 1-3 cm in diameter. High volume of carbonate spotting/beige chaotic carbonate. LT (46.8-50.2 m): LT becomes completely black with no recognizeable clasts. Py
				content increases and carbonate content decreases. White chaotic carbonate also present. Flow (50.2-51.5 m): Flow is medium-dark gray and exercises means are used of future.
65 —				and contains mionr amount of beige carbonate spots. LT (51.5-60.0 m): LT similar to those from (43.6-46.8 m) though more clast supported and contain some Ccp in upper portion of unit.

Projec	t: Boundary		UTM	Azimuth: 360
Diamo	ond Drill Hole: BD10-1	12	540848.00E	Dip: -90
Date: 8	3-13-2013		5389014.00IN	Length: 60.6 m
	ALTERATION	FACIES	8	
DEPTH (m)	Qtz Ser Sulf Sulf			DESCRIPTION
DEPTH (m) 5  10  15  20  25  30  30  35  40  40  45  50 	ALIERATION			Sample: HS00606         Breccia appears to be most flow breccias with some smaller massive flows within. Unit is dark grey with inter flow fractures filled with black chlorite. Minor dis. Py throughout and small localized zones of intense chlorite. Carbonate appears to be fracture filler-Not chaotic carbonate aspociated with the black chlorite alteration.         "Massive sulfide" is a thin lens consisting dominantly of Py and minor Ccp. Sulfide lens form sharp contacts with black chlorite. Some sharp contacts with black chlorite both above and below.         Sample: HS00607         To wort the next 10 meters is fairly consistent in that most clasts are around 1 cm and contains minor tuff layers. Clast supported with very minor visible matrix (black chl/ser). Some minor dis. Py but not much.         Sample: HS00607         To were the casts also show evidence of flow banding. Minor Ccp in a few localities.         Sample: HS00609         Taileration appears to be somewhat lesser than previous five meeters but still contains shore from ends to tails cast are about the same size but appear lighter in color.
55 	     60.0 m			Sample: H500610 Could make case for last 3 meters to be hyaloclastite breccia. Fragmentsare large (~5 -7cm) and light-dark gray in color. Matrix area contains black chlorite and Py stringer along with minor Ccp. Likely marks the depth of transition to the lower footwall BX and flows.
_				









Project: Boundary		UTM	Azimuth: 360	
Diamo	Diamond Drill Hole: BD11-117		540855.42E	Dip: -90
Date: 8	3-19-2013		5389601.35N	Length: 71 m
	ALTERATION	FACIES	>	
DEPTH	, if, is, is, is, is, is, is, is, is, is, is	- Arg - Tuff -Lap -BX	-Int -MS	
(11)				DESCRIPTION
5 —				
_				Otz-phyric flow is both massive and flow
10	, 9.4 m			banded. Upper portion is gray-green but becomes more gray-black closer to
10		0		mineralization. Contains small amounts of dis. Py and Sph. Sharp lower contact.
				Sumple. IISOUT
15-		<u>(</u> )		Sample: H500612
_				
20—	19.0 m 19.6 m			Massive pyrite. Over the LT interval from 19.6-36.5 m, clasts
_				are dark gray, matrix supported, and mostly around 2-3 cm in diameter. Matrix consists of dark ser/chl and commonly a mix of Py Cro
25—		50 50 50 9		and lesser Sph. Sample: H500613
20				
30-				
_		2000 0 90 00 P		
35—				
_		000000000		From ~37-52 m, the clasts become matrix supported and most are 1 cm in diameter.
40—				gray-yellow alteration rims. Little to no sulfides present within matrix.
_				
45				Sample: H500614
50				
-				
55—				
_		000000000		Sample: H500615 From 52-62 m, alteration intensifies and
60 —				clasts become darker and bimodal in size (~3 cm and >0.5 cm). Clasts are matrix supported and separated by mm scale
	62.0 m			veinlets of sulfides (mostly Py some Ccp).
~ ~ ~		Sing : Sing : Si		The lower hyaloclastite BX is highly silicified and takes on a light gray, "curdled"
65 -		B: SAB: SAB		appearance (may be chaotic carbonate?). Dis. Py is present in upper portion.
-		SUB: SUB: SU		
70 —	71.0 m	BORNE: SNE		Sample: H500616



Project: Boundary					UTM		Azimuth: 360	
Diamond Drill Hole: BD11-120			5	540702.53E		Dip: -90		
Date: 8	3-11-2013				5	5389673.12N		Length: 269 m
	ALTERATION		FA	CIES	\$			
DEPTH (m)	Ser Strand	- Arç	—Lap	—BX	-Flo	Int	MS	DESCRIPTION
			0.			I	I	
-			s. C					
75—			$\mathcal{O}$					
			)0.					
80—			00					
			0. S					
85 —			S.0					
_			<u>)</u> 					
90—								
_								
95 —			Sol					
,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,								
100			50					
100-			0.					LT clasts are mostly ~1-2 cm by 100 m. Clasts are varying shades of gray within a dark gray/black tuff-sericite matrix. Minor dis. Pv
			<u>s</u>					throughout.
105—								
-			50					
110—								
-			S.C					
115—								
		0000000000	)0.					
120		مل وقري ا						
								(chl/ser alt.) and slightly darker. Clasts are softer subangular, 1-2 cm in diameter, and clast
125-		and a sol						supported.
125			Sol					
120								
130-			0.					
			500					
135-								
		0000000	50					
			$o \subset$					

Project: Boundary						UTM		Azimuth: 360
Diamo	ond Drill Hole: BD11-1	20			540	540702.53E		Dip: -90
Date: 8	3-11-2013	5389673.12N			Length: 269 m			
	ALTERATION		FAC	CIES	2			
DEPTH (m)		– Arg	-Lap	-BX	-Flo	Int	MS	DESCRIPTION
- 145			0,0 0,0 0,0 0,0	·		•		LT clasts slowly increasing in size from ~135-155 m (from ~1 cm up to ~3-4 cm). Over this stretch ch alteration increases.
150								Clasts become darker and dis. Py content increases.
- 155			50					
			000000000000000000000000000000000000000					
			0 5 9					From 167-169 m, interval is marked by several thin layers of strong stringer Py and lesser Ccp (near 50-70%). Percentages are near massive, but sulfides appear to be
								directional or wormlike (near vertical). Matrix is consists dominantly of black chl.
			200 000					
180			0000					Few other small intervals of strong stringer Py marked by int. chl alteration, though mostly the rest of the unit is only qtz-ser
185—			200 000					altered. Clasts appear to be light-medium gray throughout and remain clast supported.
190—			0000					
195—			200					
200-			000					
205—			200					Slight increase in dis. Ccp from 200-210 m.

Project: Boundary		UTM	Azimuth: 360
Diamond Drill Hole: BD11-	120	540702.53E	Dip: -90
Date: 8-11-2013		5389673.121	Length: 269 m
ALTERATION DEPTH (m) + الجارية : حارية الجارية : (m)	FACIES but the formed by the f	—Flow — Int — MS	DESCRIPTION
215 220 225 230 230 235 240 245 255 260 265 260 265 269.0			Hyaloclastite is fairly consistent thoughout except for the final 10 m. The unit consists of medium gray fragments in darker gray (qt2-sef) matrix. About half the fragments are flow banded. Multiple meter scale intervals of massive flows throughout. Chlorite alteration and chaotic carbonate alteration increases ~250 m and becomes intense from 260 until the end of the hole.







Project: Boundary UTM				Azimuth: 360
Diamond Drill Hole: BD11-129 54078		540789.64E	Dip: -90	
Date: 8	3-12-13		5389563.49N	Length: 62 m
	ALTERATION	FACIES	2	
DEPTH (m)	Strarb Luff Carb	– Arç – Tufi – BX	- MS	DESCRIPTION
(,				
5 —				
-	8.1 m			
10				
_				Sample: H500665
15 —				
				Top of LT consists of distinct bimodal clast sizes (<0.5 cm and ~2 cm). Clasts are light to medium gray, angular, and matrix supported.
20				Matrix is black ser/chl. Clasts even out in size (~1cm) and become more clast supported after ~15 m. Near the flow banded contact
20-				As content seemingly increases. Minor dis. Py throughout. Localized zone of Ccpy and moderate interace she alteration
-				Sample: H500666
25—	26.1 m			
-		<u>s</u>		Sample: H501361
30 —				bands may be slighly chl altered but seem to dominantly be ser altered. Moderate dis. Py
-	32.0 m	FIF FILL		(lesser Ccp) located around margins and contact with underlying jigsaw BX. Jigsaw fit BX is dark gray with Py-Ccp filled fractures.
35—		S		Sample: H500668
-	37.6 m	S S S S S S S S S S S S S S S S S S S		Sample: H500667
40—				Sample: H500669
_				cm), black, angular clasts with Ccp-Py sulfides within matrix. Rest of LT is lighter gray in
45				color and clasts are all ~1cm. Lesser sulfides within matrix.
	45.3 m	CIDE CIDE CI		Hyaloclastite from 45.3-51.1 m consists of light to medium gray fragments that range in size from 4-8 cm with minor tuff within
		E. C.		matrix, though most matrix is black sericite. Sample: H500670
50	51.1 m	ZAB : ZAB : ZA		Flow banded flow is intensely chl altered with lighter bands of quartz present. No sulfides
-		<u>s</u>		just altered. Sample: H500671
55—	55.5 m	S S C O O O		Sample: H500672
-				Lower LT unit is moderately chl altered but no sulfides. Unit is matrix (ser) supported with ~1 cm. grey-white, rounded clasts.
60 —				
-	1 1 1			
65 —				
_				





Project: Boundary		UTM	Azimuth: 320
Diamond Drill Hole: BD11-134		540860.27E	Dip: -57
Date: 8-21-134		5389377.07N	Length: 47 m
ALTERATION	FACIES	2	
DEPTH (m) ដូរគួរគួរគ្ន	— Arg — Tufi — BX	- Int - MS	DESCRIPTION
	$\circ$		
5 — I I I I I I I I I I I I I I I I I I			Massive Py.
	000000		Light pink-gray, matrix supported LT. Clasts are bimodal (~1 cm and <0.5 cm). Matrix is light gray to black (ser?). Minor dis. Py.
			Sample: H501298
-			
			Near ~15 m, LT clasts are typical medium to dark gray with black matrix. Clasts are clast
	50 000		(~2cm). Orange chaotic carbonate present. Interval of intense chl from 20-24 m.
			Sample: H501299
	00000		
			From 30-41 3 m clasts are larger (range from
			1-4 cm), pink-gray, and clast supported. Minor dis. Py throughout. Clasts are also softer (Qtz <ser).< td=""></ser).<>
			Sample: H501300
35-			
	20000000 00000000000000000000000000000		
40			
	LLL		Sample: H501301 Jigsaw breccia contains flow banded frags.
45—   47.0 m	L L L L		Unit is light gray/blue? with fractures that cut across flow bands and are filled with Py-Ccp or dark ser/chl.
50			
55			
60—			
65—			







Projec	t: Boundary		UTM	Azimuth: 360
Diamo	ond Drill Hole: BD11-	54	54089.22E	Dip: -90
Date: 8	3-21-2013	1	5389415.151	Length: 32 m
	ALTERATION	FACIES	>	
DEPTH (m)	Qtz Ser Sulf	- Ar - Lap - BX	- Ho MS	DESCRIPTION
-				
5 —				
	7.5 n			
10				Sample: H501302
-				Most of the LT unit is matrix supported and consists of large (>2 cm), angular clasts.
15 —				gray-yellow, the middle is gray-white, and lower portion is white and pink/purple. The
-		0000000		hole. The replacement sulfides are predominantly Py with minor Ccp. The clasts
20—				have white rims and dark cores. The lower sulfide layer (30.8 m) is a thin layer of massive Py bound by sharp contacts both above and
-				below.
25 —				Sample: H501303
_				
30—	 			Sampla: HE01204
_	31.0 r 32.0 r			Sample. H501504
35				
40				
40				
45				
-				
50—				
-				
55—				
-				
60 —				
-				
65 —				
_				



Projec	t: Boundary		UTM	Azimuth: 360
Diamo	ond Drill Hole: BD11-1	83	540579.00E	Dip: -90
Date: 9	9-6-2013		5389712.00N	Length: 125 m
	ALTERATION	FACIES	>	
DEPTH (m)	arb I arb	- Arg - Tuff - Lap - BX		DESCRIPTION
-				
5 —				
-	7.4 m			Sample: H501370
10				Light gray, massive, qtz-phyric, massive flow. Minor fractures-dark sericite. Very
_				minor dis. Py.
15 —	13.2 m	Richer Christer	1	
_				BX contains qtz-pyric, flow banded frags. Matrix is black (ser). Little to no sulfides or
20		Bigge is the		carbonate spots. Both contacts are gradational.
20		COR SUR S		Sample: H501371
25—	27.1 m	Q B C Q C A C		
-				Qtz-phyric flow is dark gray to black
30 —				(qtz-ser altered only). Upper ~20 meters are flow banded.
-				
35—		9		Sample: H501372
-				
40—				
45				
45				
50—		0		Sample: H501373
-		Ŭ		
55—				Lower portion is more so massive and gets darker (ser). No chlorite or sulfides. Only minor carbonate spots. Hole goes to 125
-				m but was unable to log remaining portion of hole. Teck drill logs indicate this flow continues to 94 m before
60 —				encountering a 6 m zone of black chlorite and minor Py. From 100-125 m log indicates flow but does not mostion at
_				phenos. This likely represents the FW LT mislabeled (loggers consistantly call most
65 —				tootwall flows regardless of clast size matrix).
	70.0~	9		Sample: H501374
	70.011		-	







Projec	t: Boundary		UTM	Azimuth: 90
Diamo	ond Drill Hole: BF14-0	03	540646.41E	Dip: -38
Date: 6	5-17-2014		5389648.16N	Length: 110 m
	ALTERATION	FACIES	>	
DEPTH	let ar prest	– Arg – Tuff –Lap	-Flov - MS	DESCRIPTION
(11)				Deschir Hort
-				c
5 —				
_				
10—	9.8 m		0000	
_		Richer		
15				
15-				
-				Qtz-phyric hyaloclastite is dark gray with
20—				black chl/ser in intrafragment space. Some frags are flow banded but very few. Dis. Py throughout but sulfider increase poor
-		SAOS:SAOS:SA		contact (Py+Ccp).
25 —				
-		SVE:SVE:SV		
30—				
_				
35-				
_	36.0 m			Moderate-intense chlorite alteration. Part of matrix is Ccp and Py. Clasts (qtz only, no
10				supported.
40				
-				
45—				
-				Clasts become more clast supported pear MS
50—	51.0 m	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~		(20 cm Py+Ccp). Clasts are ~1cm and matrix consists of chl alteration and minor arg.
-	.   52.4 m	0000000		
55—				LT between tuff and hyalo appears generally unaltered, clast supported, and has clasts
_	57.5 m	0000000		sizes ~1cm. Hvaloclastite consists of black fragments with
60-	, 57.9 m	K:ONK:ONK:		white quartz infill in the martix (zebra appearance). Couple of framents are flow
		0000000		banded.
				LT Py contents greatly increases below the hyaloclastite. From here to end of hole clasts are light gravito white clast supported (no.
65 —				matrix except where noted), and bimodal in size (~2 and <0.5 cm).
-		00000000		

Project: Boundary						ГМ	Azimuth: 90
Diamond Drill H	ole: BF14-00	03			540646	.41E 9.16N	Dip: -38
Date:6-17-2014	5-17-2014 5-389648.161					Length: 110 m	
ALTE DEPTH (m) 분 평 등		Arg	FACIE de	S X8			DESCRIPTION
	110.0 m		10, 0, 0, 0, 0, 0, 0, 0, 0, 0, 0, 0, 0, 0				Over the interval of 87-100 m, matrix volume slightly increases as do the amount of sulfides present (Py).



Projec	t: Boundary		UTM	Azimuth: 360
Diamo	ond Drill Hole: BG10-0	07	540902.14E	Dip: -90
Date: 9	9-6-2013		5389507.66N	Length: 70 m
	ALTERATION	FACIES	2	
DEPTH (m)	lar Sulf		- Int MS	DESCRIPTION
	3.4 m			
5 —		6		Sample: H501364
-		Ŭ		HW is massive, dark gray with lighter gray
10				fractures, and qtz phyric. Minor carbonate spotting. No sulfides or significant visible alteration except minor chl near contact.
				Sharp contact with LT below.
15 —				
_	16.5 m		1	
20	20.1 m	De Co Co Co Do		Sample: H501365 Upper LT consists of dark gray clasts
20	21.0 m			(~1cm and less) in tuff matrix. Matrix supported. Clasts are subangular. BX contains some flow banded frags ~4-5 cm
25				in diameter. No visible faulting. LT from 21.0-52.5 m is fairly consistent.
25-				Clasts are light gray-white within tuff-arg matrix. Clasts have minor shear orientation to it. Angle is about 30
-				degrees to core axis. Clasts are all ~1cm or less. Upper portion is matrix supported in a tuff and arg.
30 —				Sample: H501366
-				
35—				Below ~32 m clasts are clast supported and very little tuff is present. Arg is still
-				present.
40—				
_				Sample: H501367
45—				
50		000000000000000000000000000000000000000		
50-				
-	52.5 m	DIS AUDISAUD		Medium gray, hyaloclastite, flow banded fragments with black sericite in margins. Minor tuff and small clasts mixed in as
55—	56.8 m			well. Sample: H501368
-		0000000		
60 —				
-				Sample: H501369
65 —				rounded than upper portion. Clasts are white-light gray and have some alteration
_				rims. Does not have sheer like appearance as upper LT.
70—	//////////////////////////////////////			

Appendix D	. I a	DIC D1.1	whole	-IUCK O	rock Geochennistry and Terraspec				
Sample ID			H500606	H500607	H500608	H500609	H500610	H500611	H500612
Hole ID			BD10-112	BD10-112	BD10-112	BD10-112	BD10-112	BD11-117	BD11-117
Depth (m)			20.1	28.8	39.8	50.9	58.2	12.6	15.2
Lithology			HBX	FLT	FLT	FLT	FLT	HFL	HFL
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	76	71.25	68.23	67.25	63.33	72.98	70.63
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	9.68	11.19	10.84	8.11	9.74	12.62	11.89
$Fe_{\tau}O_{\tau}(T)$	0/0	FUS-ICP	4 79	5.63	10.11	12.64	10.91	4.68	5.12
MpO	0/2	FUS ICP	0.210	0.041	0.010	0.000	0.061	0.080	0.122
MaQ	/0	FUS-ICF	2.219	0.041	0.019	0.009	4.82	0.089	2.97
MgO CoO	/0 0/.	FUS ICP	0.27	2.10	0.03	0.37	4.82	2.40	0.08
Na O	/0 0/	FUS-ICP	0.27	0.1	0.09	0.05	0.07	0.07	0.08
Na <sub>2</sub> O	70	FUS-ICP	0.14	0.18	0.21	0.15	0.1	0.2	0.15
K <sub>2</sub> O	%	FUS-ICP	2.01	2.65	3	2.27	1.46	2.84	2.37
TiO <sub>2</sub>	%	FUS-ICP	0.12	0.161	0.154	0.116	0.142	0.156	0.144
$P_2O_5$	%	FUS-ICP	0.01	< 0.01	0.01	< 0.01	0.02	0.01	0.01
LOI	%	FUS-ICP	4.49	4.73	6.79	7.94	7.59	4.21	4.7
Total	%	FUS-ICP	100.9	98.1	100.1	98.9	98.24	100.3	99.08
Ba	ppm	FUS-ICP	1279	1653	1571	1139	726	2516	1761
Sr	ppm	FUS-ICP	7	9	10	6	6	9	7
Y	ppm	FUS-ICP	29	33	38	23	38	41	40
Sc	ppm	FUS-ICP	9	13	13	8	10	11	10
Zr	ppm	FUS-ICP	132	151	157	113	136	159	145
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	1	< 1
V	ppm	FUS-ICP	< 5	12	< 5	< 5	< 5	< 5	5
Hg	ppb	FIMS	< 5	6	46	51	10	< 5	6
Cr	ppm	HF/HNO3	9.59	13.17	<8.9	<8.9	9.28	<8.9	<8.9
Co	ppm	HF/HNO3	1.19	1.59	<0.9	24.48	29.46	<0.9	2.59
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	344.01	78.89	2271.38	635.39	2428.34	639.85	95.41
Zn	ppm	HF/HNO3	187.11	143.24	50.92	74.64	124.52	165.12	202.06
As	ppm	HF/HNO3	<1.6	6.58	35.62	61.22	17.62	1.90	4.99
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	145.73	157.96	171.52	167.48	157.66	177.37	170.50
Mo	ppm	HF/HNO3	1.98	23.53	9.41	22.00	13.17	2.73	2.39
Ag	ppm	HF/HNO3	<1	<1	1.30	4.87	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	1.70	<0.9
Sn	ppm	HF/HNO3	6.24	6.21	7.81	6.09	5.55	8.60	6.90
Sb	ppm	HF/HNO3	1.06	3.37	5.98	4.99	1.89	1.22	7.28
le	ppm	HF/HNO3	<4.3	<4.3	<4.3	10.93	5.26	<4.3	<4.3
l W	ppm	HF/HNO3	<3/	<3/	<3/	<3/	<3/	<3/	<3/
W	ppm	HF/HNO3	1.70	3.09	4.10	4.07	2.13	1.50	1.72
PU Di	ppm	HE/INO2	~23.5	~23.5	20.30	52.07	~23.3	~23.5	~23.5
BI	ppm	No O Sinton	0.23	0.97	0.23	19.30	7.10	0.44	0.33
Nb	ppm	$Na_2O_2$ Sinter	5.14	6.33	6.01	4.53	5.96	6.75	6.21
La	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	15.51	19.76	17.87	11.81	18.76	19.25	17.82
Ce	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	32.87	42.76	37.71	25.75	38.97	41.75	38.08
Pr	ppm	Na2O2 Sinter	4.13	5.46	4.74	3.24	4.99	5.29	4.81
Nd	ppm	Na2O2 Sinter	17.70	23.28	19.86	13.63	21.28	22.80	20.89
Sm	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 39	5.61	4 99	3 33	5.09	5 54	5.07
En	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.78	1 34	1.27	0.91	1 18	0.83	1.15
Th	ppin	Na.O. Sinter	0.70	0.05	0.07	0.62	1.00	1.04	0.06
10	ppm	No O Sinter	0.79	0.93	0.97	0.62	1.00	1.04	0.96
Dy	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	5.24	6.27	6.42	4.18	6.48	6.93	6.34
Но	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.14	1.34	1.38	0.92	1.36	1.47	1.37
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	3.62	4.10	4.11	2.81	4.17	4.57	4.25
Tm	ppm	Na2O2 Sinter	0.55	0.61	0.62	0.43	0.62	0.70	0.66
Yb	ppm	Na2O2 Sinter	3.76	4.08	4.15	2.95	4.15	4.92	4.64
Lu	ppm	Na2O2 Sinter	0.59	0.62	0.63	0.45	0.65	0.77	0.72
Hf	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	3 34	4 11	3.83	3.07	3 43	4 55	4 16
Та	ppin	Na.O. Sinter	0.25	0.20	0.25	0.21	0.24	0.29	0.26
1 a Th	ppm	Na O Sintor	0.23	0.29	0.23	0.21	0.24	0.28	0.20
	ppm	Tama and	4.33	5.00	4.61	3.66	4.36	5.57	4.99
AIOH WL AIOU Donth	nm 9/	Terraspec	2,204	2,202	2,205	2,204	2,199	2,202	2,202
AIOH Depth	%	1 erraspec	5.99%	0.26%	3.89%	10.76%	1.66%	9.90%	5.83%
FeOH Dopth	nm 0/	Torrace	2,203	2,233	#DIV/0!	#DIV/0! #DIV/0!	2,232	2,233	2,235
AlOH Jepth	70	Terraspec	2.79%	1.95%	#DIV/0!	#DIV/0!	3.00%	2.80%	1.91%
AIOH deptn/reOH depth		1 erraspec	2.15	3.24	#DIV/0!	#D1V/0!	0.54	3.34	3.06
Abbreviations:	Units	Hanain 11 d		FT= Footwall t	uff	Alteration	ta anioit-	Other	oth
	ПГL= ЦDV.	<ul> <li>nanging wall flo</li> <li>Hanging wall be</li> </ul>	occia	FEI = Footwall	flow	Chl-Ser= Chlor	iz-sericite	wL- waveler	igui
	TIDA:	- manging wall Dr	uu a	INT Interest	now	Chi-Chi-chior	ne-seriene		
	FLI=	rootwall lapilli ti	111	11NI = Intrusion		Uni=Chiorite			

		ТМ	
Annondiv D. Table D1 1 V	Whole wool Coochamistr	u and Tannaanaa <sup>111</sup> Dat	-
ADDENUIX D. LADIE DI.I V	и поле-госк стеоспенныг	vanu rerraspec Dat	

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Аррениіх	<b>D</b> , 1				I ULUL	nemisti	y anu l	i ci i asp		/ata
Sample ID			H500613	H500614	H500615	H500616	H500617	H500618	H500619	H500621
Hole ID			BD11-117	BD11-117	BD11-117	BD11-117	BD10-107	BD10-107	BD10-107	BD10-108
Depth (m)			25.5	45.1	57.7	70.4	10.8	18.3	31.2	22.5
Lithology			FLT	FLT	FLT	FBX	FBX	FLT	FLT	FLT
Alteration			Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	60.72	58.05	63.78	73.95	72.99	71	46.74	55.03
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	9.29	13.79	8.84	11.57	12.62	11.66	18.87	8.85
$Fe_2O_3(T)$	%	FUS-ICP	14.59	10.85	13.65	4.08	3.99	4.6	14.28	18.7
MnO	%	FUS-ICP	0.049	0.066	0.015	0.035	0.047	0.092	0.088	0.189
MgO	%	FUS-ICP	2.33	2.87	0.56	1.97	2.21	4.2	8.39	2.71
CaO	%	FUS-ICP	0.07	0.06	0.09	0.11	0.24	0.14	0.26	0.21
Na <sub>2</sub> O	%	FUS-ICP	0.13	0.22	0.16	0.18	0.21	0.2	0.2	0.15
K <sub>2</sub> O	%	FUS-ICP	2.08	34	2 43	2.87	3.07	2 33	2 33	1.87
TiO.	0/0	FUS-ICP	0.122	0.19	0.123	0.158	0.157	0.18	0.263	0.128
RO	0/	FUS ICD	< 0.01	0.02	< 0.01	< 0.01	0.02	0.10	0.205	0.120
1 <sub>2</sub> 0 <sub>5</sub>	70 0/.	FUS-ICF	0.67	8.62	< 0.01	2.05	0.05	4.01	8.11	11.27
LOI Total	70 0/.	FUS-ICP	9.07	08.02	08.56	3.93	4 00 57	4.91	00.6	00.22
Po	/0	FUS-ICF	1410	1927	1106	1676	2011	1241	1207	99.22
Da Sr	ppm	FUS-ICP	7	10	7	0	10	10	1207	7
V	ppm	FUS-ICP	25	10	31	42	10	40	60	30
Sc	ppm	FUS-ICP	25	14	8	12	10	11	16	11
Zr	nnm	FUS-ICP	116	174	111	147	146	144	209	105
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	1	< 1	< 1	< 1
V	nnm	FUS-ICP	5	< 5	< 5	< 5	< 5	15	< 5	21
Но	nnh	FIMS	458	263	37	< 5	5	< 5	7	10
Cr	ppo	HF/HNO3	10.80	<8.9	<8.9	15 21	< 8 9	<8.9	<89	21.34
Co	ppm	HF/HNO3	1.89	1.71	35.76	0.93	<0.9	2.22	6.21	4.11
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	14.61	<10	<10
Cu	ppm	HF/HNO3	117.56	816.42	512.75	940.60	646.06	<19	312.02	25.39
Zn	ppm	HF/HNO3	123.86	113.78	36.54	113.27	70.34	112.58	187.70	53.61
As	ppm	HF/HNO3	78.03	47.76	28.56	3.89	5.24	2.01	<1.6	135.59
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	164.68	170.70	164.06	173.52	198.44	182.29	220.37	236.73
Mo	ppm	HF/HNO3	19.39	21.94	20.53	7.51	1.76	5.00	19.28	11.54
Ag	ppm	HF/HNO3	1.37	<1	<1	<1	<1	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	< 0.9	<0.9	<0.9	<0.9	<0.9	< 0.9	<0.9
Sn	ppm	HF/HNO3	6.76	7.88	5.98	6.78	4.96	3.46	5.47	5.44
Sb	ppm	HF/HNO3	25.53	18.63	2.18	1.78	1.27	0.71	1.33	1.21
Те	ppm	HF/HNO3	<4.3	<4.3	10.95	<4.3	<4.3	<4.3	20.10	10.70
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.73	2.45	1.58	2.10	2.29	3.64	6.10	7.49
Pb	ppm	HF/HNO3	96.62	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HF/HNO3	4.33	6.24	11.82	1.45	0.36	0.52	19.97	10.08
Nb	ppm	Na2O2 Sinter	4.44	7.22	4.52	6.35	8.27	8.78	12.81	5.40
La	ppm	Na2O2 Sinter	15.94	18.28	17.24	18.28	21.45	22.52	32.20	16.61
Ce	ppm	Na2O2 Sinter	33.59	39.49	36.10	38.66	45.20	47.62	69.18	33.98
Pr	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 24	5.05	4 50	4.85	5.76	6.01	8 77	4.15
NJ	ppin	Na.O. Sinter	18.07	21.22	10.24	21.01	22.05	25.76	28.20	17.09
Nu	ppm	Na O Sinter	18.07	21.22	19.34	21.01	23.93	23.70	38.20	17.08
Sm	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.28	5.24	4.56	5.04	5.75	6.23	9.04	4.11
Eu	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.14	1.28	0.94	1.02	0.81	1.48	2.18	1.20
Tb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.64	1.03	0.83	1.06	1.16	1.15	1.71	0.84
Dy	ppm	Na2O2 Sinter	4.14	6.97	5.41	7.19	7.98	7.73	11.62	5.42
Но	ppm	Na2O2 Sinter	0.88	1.49	1.12	1.53	1.71	1.60	2.43	1.14
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	2.66	4 51	3 38	4 74	5 21	4 93	7 48	3 48
Tm	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.39	0.70	0.50	0.72	0.80	0.77	1 13	0.52
vh	Phil	Na <sub>2</sub> O <sub>2</sub> Sinter	2.57	1 71	3 20	4.02	5 26	5.11	7 50	3 77
10	ppm	No O Sinter	2.72	4.71	5.39	4.92	3.30	3.11	7.30	5.72
Lu	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.42	0.74	0.52	0.76	0.85	0.79	1.19	0.60
Hf	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	2.89	4.78	2.97	4.02	5.89	5.95	9.12	4.05
Та	ppm	Na2O2 Sinter	0.16	0.27	0.19	0.27	0.34	0.36	0.49	0.20
Th	ppm	Na2O2 Sinter	3.63	5.40	3.72	5.09	7.35	7.42	10.68	4.19
AIOH WL	nm	Terraspec	2,204	2,203	2,204	2,205	2,205	2,201	2,195	2,197
AlOH Depth	%	Terraspec	3.09%	6.64%	6.14%	11.38%	3.88%	8.73%	2.83%	4.14%
FeOH WL	nm	Terraspec	2,250	2,252	#DIV/0!	2,253	2,253	2,252	2,252	2,253
FeOH Depth	%	Terraspec	1.21%	1.11%	#DIV/0!	1.20%	1.21%	1.98%	5.56%	2.88%
AlOH depth/FeOH depth		Terraspec	2.57	5.98	#DIV/0!	9.51	3.21	4.42	0.51	1.44
										_
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	<ul> <li>Hanging wall f</li> </ul>	low	FBX= Footwa	all breccia	Qtz-Ser= Qua	rtz-sericite		WL= Waveler	ngth
	HBX	= Hanging wall b	reccia	FFL= Footwa	ll flow	Chl-Ser= Chlo	orite-sericite			
	FLT=	Footwall lapilli	tuff	INT= Intrusio	n	Chl=Chlorite				

Annendix B• '	Table R1 1	Whole-rock	Geochemistry	and Terrasn	ec <sup>TM</sup> Data
				7 HUL I VI I AND	

Appendix D.	. I a	DIE DI.I	w noie	-IUCK O	reochei	msu y a		raspec	Data
Sample ID			H500622	H500623	H500624	H500625	H500626	H500627	H500628
Hole ID			BD10-108	BD10-108	BD10-108	BD10-108	BD10-108	BD10-108	BD10-108
Depth (m)			28.3	34	36.6	43.5	46.4	47.4	50.6
Lithology			FLT	FBX	FLT	FBX	FLT	FLT	FFI
Alteration			Chl-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Chl	Chl	Otz-Ser
Alteration	0./	FUG LOD	55.07	Qiz-Sei	55.04	Q12-361	51.07	20.00	Q12-361
SIO <sub>2</sub>	%0	FUS-ICP	55.07	04.4	55.94	80.91	51.87	39.09	07.90
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	14.42	12.38	9.29	7.01	10.68	14.09	14.33
$Fe_2O_3(T)$	%	FUS-ICP	12.8	10.12	19.03	4.74	18.37	21.68	5.49
MnO	%	FUS-ICP	0.065	0.021	0.016	0.043	0.172	0.108	0.025
MgO	%	FUS-ICP	4 12	0.79	0.73	1 13	8 27	9.05	2 37
CoO	0/.	FUS ICP	0.16	0.05	0.06	0.07	0.11	0.1	0.04
CaO	/0	FUS-ICF	0.10	0.03	0.00	0.07	0.11	0.1	0.04
Na <sub>2</sub> O	%	FUS-ICP	0.24	0.22	0.18	0.13	0.06	0.11	0.25
K <sub>2</sub> O	%	FUS-ICP	3.2	3.32	2.38	1.66	0.39	1.24	3.23
TiO <sub>2</sub>	%	FUS-ICP	0.417	0.21	0.129	0.103	0.231	0.199	0.197
P.O.	%	FUS-ICP	0.06	0.02	< 0.01	< 0.01	0.04	0.04	0.02
1,205	0/	FUS ICP	0.34	7 22	10.01	2.2	0.84	12.46	4.28
LOI	70	FUS-ICP	9.54	7.52	10.91	3.3	9.64	12.40	4.28
Total	%0	FUS-ICP	99.89	98.80	98.68	99.1	100	98.17	98.19
Ba	ppm	FUS-ICP	1498	2181	1123	998	165	622	1560
Sr	ppm	FUS-ICP	13	11	7	6	3	6	10
Y	ppm	FUS-ICP	37	46	28	27	44	50	41
Sc	ppm	FUS-ICP	21	17	9	8	13	14	16
Zr	ppm	FUS-ICP	123	142	118	94	130	189	186
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	98	23	6	6	26	5	8
Нσ	nnh	FIMS	44	171	93	24	14	16	7
Cr	nnm	HE/HNO3	45 51	20.65	19.25	<8.9	12 59	10.43	<8.9
Co	ppm	UE/UNO2	17.02	8 70	12.22	6.26	20.02	20.07	2.96
Ni	ppm	III/IINO2	22.55	11.20	<10	<10	<10	27.07	5.00 <10
INI Cu	ppm	HF/HNO3	25.55	11.59	2522.04	<10	1120.22	<10 451.65	1205.10
Cu	ppm	HF/HNO3	112.95	458.08	2522.04	/2/.91	1129.33	451.65	1205.19
Zn	ppm	HF/HNO3	103.93	526.75	49.30	93.12	132.26	176.51	38.46
As	ppm	HF/HNO3	25.84	41.76	218.58	7.88	16.35	144.47	8.01
Se	ppm	HF/HNO3	<26.7	<26.7	40.56	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	254.58	262.97	269.21	238.24	240.28	253.41	211.83
Mo	ppm	HF/HNO3	3.42	6.84	6.20	8.16	12.85	10.77	6.44
Ag	ppm	HF/HNO3	1.30	1.19	1.03	<1	<1	<1	<1
Cd	ppm	HF/HNO3	< 0.9	1.56	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9
Sn	ppm	HF/HNO3	4.71	5.32	5.70	4.07	3.43	4.14	4.15
Sb	ppm	HF/HNO3	5 40	6.84	9 76	1.51	1 12	1.82	1 39
Te	nnm	HE/HNO3	<4.3	<4.3	<43	5 39	11.04	8 22	5.05
I	ppm	UE/UNO2	<1.5	<1.5	<1.5	~27	<27	<27	<27
1	ppm	HF/HNO3	14.21	< <u>3</u> /	< <u>3</u> /	2.22	< <u>3</u> /	2.79	< 99
W	ppm	HF/HNO3	14.21	9.32	4.11	3.23	0.49	2.78	0.00
Pb	ppm	HF/HNO3	<25.3	31.65	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HF/HNO3	4.34	1.73	13.26	4.07	10.56	9.84	7.26
Nb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	7.46	7.79	6.46	5.31	8.46	10.72	10.90
La	ppm	Na2O2 Sinter	15.75	19.26	16.14	14.54	16.38	25.16	25.23
Ce	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	33.26	40.06	34 33	30.37	34.62	53 21	53.28
2	ppm	No O Sinter	55.20	40.00	54.55	50.57	54.02	55.21	55.20
Pr	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.22	5.08	4.35	3.73	4.35	6.69	6.59
Nd	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	18.51	21.29	18.25	15.82	18.64	28.38	27.23
Sm	ppm	Na2O2 Sinter	4.75	5.24	4.43	3.73	4.92	6.78	6.46
Fu	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	1 31	2 19	0.93	0.85	1.07	1.46	1 37
Eu .	ppm	No O Sintor	0.00	2.17	0.75	0.05	1.10	1.40	1.57
16	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.98	1.16	0.79	0.70	1.19	1.33	1.12
Dy	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	6.79	7.82	5.29	4.46	8.17	9.05	7.47
Но	ppm	Na2O2 Sinter	1.46	1.69	1.11	0.98	1.70	1.97	1.58
Fr	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 4 3	5.03	3 31	2 97	5.16	5.93	4 90
Tm	ppin	Na.O. Sinter	0.67	0.76	0.52	0.46	0.81	0.02	0.74
Im	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.67	0.76	0.52	0.46	0.81	0.92	0.74
Yb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.71	5.30	3.73	3.06	5.27	6.28	5.14
Lu	ppm	Na2O2 Sinter	0.69	0.77	0.55	0.46	0.78	0.95	0.78
Hf	ppm	Na2O2 Sinter	4.79	4.77	4.10	3.00	4.33	6.86	5.48
То	F.F	Na.O. Sinter	0.25	0.25	0.24	0.21	0.20	0.27	0.40
1a	ppm	N <sub>2</sub> O <sub>2</sub> Sinter	0.25	0.25	0.24	0.21	0.50	0.57	0.40
Th	ppm	$1Na_2O_2$ Sinter	4.65	5.40	4.60	3.77	4.94	7.30	7.30
AIOH WL	nm	Terraspec	2,201	2,203	2,203	2,200	2,195	2,196	2,200
AlOH Depth	%	Terraspec	7.31%	3.51%	4.06%	8.28%	1.82%	5.73%	13.68%
FeOH WL	nm	Terraspec	2,252	#DIV/0!	#DIV/0!	2,253	2,253	2,252	2,254
FeOH Depth	%	Terraspec	1.89%	#DIV/0!	#DIV/0!	5.23%	5.24%	7.46%	3.47%
AlOH depth/FeOH depth		Terraspec	3.87	#DIV/0!	#DIV/0!	1.58	0.35	0.77	3.95
Abbreviations:	Units HFL= HBX:	<ul> <li>Hanging wall flo</li> <li>Hanging wall broken and the second s</li></ul>	ow eccia	FT= Footwall FBX= Footwal FFL= Footwal	tuff Il breccia I flow	Alteration Qtz-Ser= Quar Chl-Ser= Chlor	tz-sericite	Other WL= Waveleng	gth
	FLT=	Footwall lapilli tu	ıff	INT= Intrusion	ı	Chl=Chlorite			

Appendix B: Table B1.1 Whole-rock Geochemistry and Terraspec<sup>TM</sup> Data
Appendi	х в:	I able I	51.1 WI	lole-roc	K Geoc	nemistr	'y and i	errasp	ec Da	ลเล
Sample ID			H500629	H500642	H500643	H500644	H500645	H500649	H500650	H500661
Hole ID			BD10-108	BD10-104	BD10-104	BD11-120	BD11-120	BD10-88	BD10-88	BD10-102
Depth (m)			53.2	14.3	21.6	22.8	63.3	8.2	22.1	6.2
Lithology			FLT	FLT	FLT	HFL	FLT	FLT	FLT	FLT
Alteration			Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	55.23	73.5	73.56	78.78	75.98	73.06	73.2	55.23
Al <sub>2</sub> O <sub>2</sub>	%	FUS-ICP	9.43	11.72	9.18	9.21	6.94	9.41	9.95	8.63
$Fe_{2}O_{2}(T)$	0/0	FUS-ICP	16.51	3 54	5 79	3 54	8 29	6 79	6.72	20.71
MpO	0/.	FUS ICP	0.173	0.067	0.041	0.102	0.023	0.042	0.02	0.015
MaQ	70 0/.	FUS ICP	6.91	1.7	2.02	1.40	2.45	2.92	4.15	0.015
NgO	70	FUS-ICP	0.01	1.7	5.02	1.49	2.43	5.65	4.15	0.55
CaO	%	FUS-ICP	0.04	0.13	0.1	0.32	0.06	0.05	0.05	0.08
Na <sub>2</sub> O	%	FUS-ICP	0.04	0.24	0.16	0.15	0.13	0.12	0.15	0.17
K <sub>2</sub> O	%	FUS-ICP	0.22	3	1.82	2.35	1.2	1.51	1.42	2.31
TiO <sub>2</sub>	%	FUS-ICP	0.14	0.164	0.148	0.115	0.137	0.156	0.18	0.157
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	0.02	0.02	0.02	< 0.01	< 0.01	< 0.01	< 0.01
LOI	%	FUS-ICP	8.41	3.97	4.36	3.71	5.41	4.84	4.62	11.27
Total	%	FUS-ICP	97	98.06	98.2	99 79	100.6	99.81	100.5	98.93
Ba	nnm	FUS-ICP	106	1380	764	1082	405	778	579	1258
Sr.	ppm	FUS-ICP	< 2	13	8	9	7	6	9	1230
V	ppm	FUS-ICP	32	30	30	26	24	27	31	30
Sc	ppm	FUS-ICP	10	12	0	20	24 7	13	12	0
50 7r	ppm	FUS-ICP	10	12	9	110	112	15	15	9 104
	ppm	FUS-ICP	121	132	95	110	112	- 1	11/	104
DC	ppm	FUS-ICP	< 1	1	< 1	< 1	< 1	< 1	< 1	< 1
V H	ppm	FUS-ICP	5	1	16	< 5	11	18	21	12
нg	ppb	FIMS	34	< 5	< 5	7	< 5	30	< 5	19
Cr	ppm	HF/HNO3	<8.9	<8.9	9.08	<8.9	12.69	13.18	19.99	12.77
Co	ppm	HF/HNO3	19.90	1.77	2.65	<0.9	17.84	8.04	12.53	25.47
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	16782.95	<19	290.37	93.11	<19	197.34	71.07	8312.24
Zn	ppm	HF/HNO3	200.77	176.25	96.23	66.14	26.31	112.58	167.27	322.51
As	ppm	HF/HNO3	15.47	<1.6	7.12	6.95	5.11	<1.6	12.85	23.15
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	242.58	232.87	180.67	174.04	256.87	152.15	158.47	174.28
Mo	ppm	HF/HNO3	19.41	41.77	7.75	3.79	13.99	3.72	2.90	32.17
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	1.60
Cd	ppm	HF/HNO3	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9
Sn	ppm	HF/HNO3	4.33	6.79	5.26	5.90	3.43	5.54	4.79	8.55
Sb	ppm	HF/HNO3	2.43	1.04	0.98	1.56	0.76	0.39	0.74	1.66
Те	ppm	HF/HNO3	6.12	<4 3	4 66	<4 3	<4 3	<4 3	<4 3	<4 3
I	nnm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HE/HNO3	6.40	5.00	4 62	1 97	5 71	6.75	5.83	3 80
Ph	ppm	HE/HNO3	28.91	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HE/HNO3	7.86	1 17	3.08	2.80	1.49	0.58	0.50	4 39
	ppm	No O Sintor	7.80	1.17	5.76	2.00	1.49	0.56	0.50	4.57
Nb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	/.50	7.02	5.36	4.11	4.16	4.16	4.41	4.41
La	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	15.12	2.97	12.67	14.00	8.75	15.47	14.92	16.22
Ce	ppm	Na2O2 Sinter	31.48	6.35	26.79	29.76	17.39	28.10	30.33	33.24
Pr	ppm	Na2O2 Sinter	3.97	0.83	3.33	3.71	2.15	3.30	3.77	4.08
Nd	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	16.85	3.84	14.36	16.04	8 56	14.03	15.85	17.40
	ppm	Na O Sintor	10.05	1.22	14.50	10.04	0.50	14.05	13.65	1/.40
500	ppm	Na <sub>2</sub> O <sub>2</sub> Sintel	4.22	1.33	3.48	5.62	2.12	5.21	5./4	4.09
Eu	ppm	$1Na_2O_2$ Sinter	0.79	0.35	0.62	0.67	0.41	0.65	0.58	0.86
Tb	ppm	Na2O2 Sinter	0.94	0.58	0.73	0.64	0.49	0.63	0.75	0.76
Dy	ppm	Na2O2 Sinter	6.12	4.44	5.06	4.21	3.40	3.96	5.11	5.08
Но	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1 32	1.07	1 18	0.89	0.81	0.88	1.08	1.05
 E+	PPm	Na <sub>2</sub> O <sub>2</sub> Sinter	2.04	2 20	2 5 4	2 70	2 51	2.00	2 4 2	2 22
E1	ppm	Na O Sinte	3.94	5.39	3.34	2.70	2.31	2.84	5.45	3.22
Im	ppm	$1Na_2O_2$ Sinter	0.60	0.55	0.55	0.42	0.37	0.43	0.52	0.47
Yb	ppm	Na2O2 Sinter	4.07	3.97	3.76	2.98	2.72	3.03	3.56	3.24
Lu	ppm	Na2O2 Sinter	0.61	0.61	0.56	0.47	0.42	0.48	0.56	0.51
Hf	ppm	Na2O2 Sinter	4.74	4.94	4.05	2.88	2.29	2.50	2.80	2.89
Та	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.24	0.31	0.23	0.19	0.16	0.18	0.20	0.13
т.,	ppm	Na O Sintor	0.24	5.00	0.25	0.17	0.10	0.10	0.20	0.15
In	ppm	ma <sub>2</sub> O <sub>2</sub> Sinter	4.66	5.98	4.92	3.80	3.14	3.59	3.99	3.47
AIOH WL	nm	Terraspec	2,197	2,202	2,201	2,205	2,198	2,200	2,198	2,203
AIOH Depth	%	Terraspec	4.55%	13.45%	9.71%	5.95%	10.41%	13.60%	9.75%	10.87%
FeOH WL	nm	Terraspec	2,253	2,254	2,252	#DIV/0!	2,253	2,253	2,254	#DIV/0!
FeOH Depth	%	Terraspec	6.50%	1.42%	3.18%	#DIV/0!	2.96%	7.78%	5.15%	#DIV/0!
AlOH depth/FeOH depth		Terraspec	0.70	9.51	3.06	#DIV/0!	3.52	1.75	1.89	#DIV/0!
Abbreviations:	Units HFL=	Hanging wall	flow	FT= Footwall FBX= Footwa	tuff all breccia	Alteration Qtz-Ser= Qua	rtz-sericite		Other WL= Waveler	igth

		· · · TM p	
Appendix B: Table BL	Whole-rock Geochem	istry and Terraspec Dat	ta

HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow INT= Intrusion

Chl-Ser= Chlorite-sericite Chl=Chlorite

	- 21	I dole E	11200((2	1010 100	11 00000	11200//2	J 4612 64 .	usp	11200/(0	11500//0
Sample ID			H500662	H500663	H500664	H500665	H500666	H500667	H500668	H500669
Hole ID			BD10-102	BD10-102	BD10-102	BD11-129	BD11-129	BD11-129	BDI1-129	BD11-129
Depth (m)			16.6	25.1	43.7	12.8	24.8	38.1	35.4	38.4
Lithology			FLT	FLT	FLT	FLT	FLT	FLT	FFL	FLT
Alteration			Chl	Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	19.65	52.08	74.39	70.17	64.64	60.05	77.19	62.75
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	15.81	6.05	9.66	13.78	8.64	6.94	8.48	8.47
Fe <sub>2</sub> O <sub>2</sub> (T)	%	FUS-ICP	27.58	21.35	6.89	3.37	14.33	13.72	5.76	11.67
MnO	0/0	FUS-ICP	0.142	0.073	0.058	0.061	0.007	0.025	0.008	0.01
MgO	0/2	FUS-ICP	16.1	5.47	2.08	4.45	0.3	0.45	0.38	0.34
CoO	70 0/.	FUS ICP	0.07	0.06	2.96	4.45	0.04	0.45	0.03	0.06
No	/0	FUS-ICF	0.07	0.00	0.00	0.07	0.04	0.08	0.03	0.00
Na <sub>2</sub> O	70	FUS-ICF	0.02	0.03	0.16	0.24	0.2	0.16	0.18	0.2
K <sub>2</sub> O	%	FUS-ICP	0.05	0.12	1.65	3.02	2.32	1.87	2.27	2.32
TiO <sub>2</sub>	%	FUS-ICP	0.251	0.095	0.183	0.186	0.137	0.107	0.116	0.124
$P_2O_5$	%	FUS-ICP	0.02	0.01	0.02	0.03	< 0.01	< 0.01	< 0.01	< 0.01
LOI	%	FUS-ICP	15.65	11.66	4.18	4.77	8.81	9.87	4.04	9.2
Total	%	FUS-ICP	95.33	97	100.2	100.1	99.43	93.28	98.47	95.13
Ва	ppm	FUS-ICP	22	46	641	2096	1154	951	1098	1099
Sr	ppm	FUS-ICP	3	4	9	14	8	8	6	8
Y	ppm	FUS-ICP	59	23	31	38	30	22	28	33
Sc	ppm	FUS-ICP	16	7	11	14	9	7	8	8
Zr	ppm	FUS-ICP	219	86	118	178	104	95	107	108
Be	ppm	FUS-ICP	< 1	< 1	< 1	1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	17	6	25	22	16	14	7	16
Hg	ppb	FIMS	108	100	11	37	62	237	49	134
Cr	ppm	HF/HNO3	<8.9	14.01	15.31	<8.9	<8.9	<8.9	17.99	<8.9
Со	ppm	HF/HNO3	107.92	133.23	4.84	< 0.9	13.44	44.93	11.01	9.61
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	27636.08	16898.55	271.56	138.77	215.86	14253.16	993.30	<19
Zn	ppm	HF/HNO3	246.54	167.78	73.02	168.01	47.86	62.94	38.88	74.06
As	ppm	HF/HNO3	34.05	31.47	3.75	3.80	47.26	37.57	9.16	23.86
Se	ppm	HF/HNO3	34.65	61.22	<26.7	<26.7	<26.7	38.76	<26.7	<26.7
Br	ppm	HF/HNO3	180.37	162.60	151.22	238.10	219.01	225.14	226.02	257.14
Mo	ppm	HF/HNO3	61 23	27.15	2 30	6 36	6.86	16.68	4 77	3 69
Ag	ppm	HF/HNO3	3.01	2 13	<1	<1	1.02	3.03	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	7 49	4 25	5 71	3 92	4 10	4 90	3 61	3.52
Sh	nnm	HF/HNO3	3 40	2 33	0.65	8.05	6.53	8.00	1.56	1.67
Te	nnm	HF/HNO3	59.31	45.98	<4 3	<4 3	7.05	16 30	<4 3	<4 3
I	nnm	HE/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	nnm	HF/HNO3	26.54	15 59	9 54	4 05	3.67	2.62	3.85	3.95
Ph	nnm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	33.81	<25.3	<25.3
Bi	nnm	HF/HNO3	83 73	44.03	1.76	0.32	24 32	27.08	5.10	1.89
Nb	ppin	Na.O. Sinter	8 17	3.00	5.69	10.77	5.96	5 47	5.70	6.07
IND .	ppm	No O Sintor	0.17	3.00	5.08	10.77	5.80	3.47	3.79	0.07
La	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	33.76	9.32	10.41	25.38	15.66	13.02	15.54	15.94
Ce	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	65.11	19.28	22.45	53.38	32.56	26.88	32.42	33.18
Pr	ppm	Na2O2 Sinter	8.13	2.41	2.93	6.86	4.13	3.38	4.15	4.13
Nd	ppm	Na2O2 Sinter	34.95	10.36	12.70	29.59	18.18	14.08	17.38	17.58
Sm	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	8 76	2 36	3 28	6 87	4 44	3 47	4 01	4 24
En	PP	Na.O. Sinter	1.02	0.25	0.61	1.66	1.00	0.82	0.71	0.04
Eu Ti	ppm	No O Sintor	1.05	0.25	0.01	1.00	1.00	0.62	0.71	0.94
10	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.59	0.55	0.75	1.14	0.84	0.63	0.76	0.85
Dy	ppm	$Na_2O_2$ Sinter	10.19	3.75	5.28	7.09	5.52	4.06	5.14	5.44
Но	ppm	Na2O2 Sinter	2.09	0.83	1.18	1.55	1.22	0.87	1.08	1.20
Er	ppm	Na2O2 Sinter	6.26	2.53	3.63	4.67	3.91	2.69	3.42	3.68
Tm	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.96	0.38	0.56	0.74	0.58	0.40	0.54	0.56
Vh	PP	Na.O. Sinter	6.75	2.65	2.95	4.84	3 79	2.00	2.60	2.86
10	ppm	Na O Sinter	0.75	2.03	5.85	4.64	5.78	2.90	5.00	5.80
Lu	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.01	0.40	0.59	0.76	0.59	0.45	0.55	0.60
Hf	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	5.25	2.03	3.25	6.15	3.86	3.21	3.17	3.65
Та	ppm	Na2O2 Sinter	0.34	0.13	0.21	0.37	0.22	0.17	0.21	0.18
Th	ppm	Na2O2 Sinter	7.59	2.68	4.44	6.98	4.38	3.57	4.28	4.11
AlOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	2,199	2,202	2,202	2,208	2,202	2,202
AlOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	2.81%	11.62%	4.19%	1.74%	4.64%	6.71%
FeOH WL	nm	Terraspec	2,251	2,251	2,253	2,252	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!
FeOH Depth	%	Terraspec	2.28%	2.45%	2.02%	1.60%	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	1.39	7.25	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!
Abbreviations:	Units			FT= Footwall	tuff	Alteration		Other		
	HFL=	Hanging wall fl	ow	FBX= Footwa	ll breccia	Qtz-Ser= Quar	tz-sericite	WL= Waveleng	gth	
	HBX	= Hanging wall be	reccia	FFL= Footwal	l flow	Chl-Ser= Chlor	rite-sericite			

FLT= Footwall lapilli tuff

INT= Intrusion

Chl=Chlorite

		TM
Annondiv R. Tahla R1 1	Whole real Conchamistr	w and Tarragnaa <sup>1111</sup> Data
Appendix D. Table D1.1	w noie-rock Geochennsu	vallu i tiraspet Data

Sample ID			H500670	H500671	H500672	H500673	H500674	H500675	H500676	H500677
Hole ID			BD11-129	BD11-129	BD11-129	BD11-123	BD11-123	BD11-123	BD11-123	BD11-12
ithology			FBX	FFL	FLT	HFL	FLT	FFL	FLT	22.5 FLT
Alteration			Qtz-Ser	Chl-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	70.86	66.7	63.16	75.93	66.95	73.3	68.59	63.93
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	11.31	11.2	14.32	11.19	13.78	11.38	11.63	15.22
$e_2O_3(T)$	%	FUS-ICP	6.97	9.1	7.06	3.5	5.82	4.31	5.7	6.9
4nO	%	FUS-ICP	0.024	0.061	0.089	0.108	0.137	0.236	0.195	0.118
ЛgO	%	FUS-ICP	1.93	5.96	5.64	2.41	3.16	2.9	3.67	2.41
CaO	%	FUS-ICP	0.03	0.03	0.07	0.07	0.12	0.07	0.07	0.1
la <sub>2</sub> O	%	FUS-ICP	0.21	0.11	0.2	0.28	0.31	0.2	0.21	0.33
L <sub>2</sub> O	%	FUS-ICP	2.68	1.2	2.42	2.71	3.37	2.89	2.77	3.98
iO <sub>2</sub>	%	FUS-ICP	0.155	0.151	0.208	0.14	0.193	0.153	0.164	0.224
2O5	%	FUS-ICP	< 0.01	< 0.01	0.03	0.02	0.03	< 0.01	0.03	0.03
OI	%	FUS-ICP	5.15	5.58	5.87	4.38	6.2	5.2	6.31	6.89
otal	%	FUS-ICP	99.33	100.1	99.07	100.7	100.1	100.6	99.35	100.1
a	ppm	FUS-ICP	1310	579	1206	1843	2082	1595	1394	3118
r	ppm	FUS-ICP	8	4	8	12	13	8	9	16
	ppm	FUS-ICP	40	46	61	38	47	37	45	55
c	ppm	FUS-ICP	13	11	16	10	14	13	12	16
.r	ppm	FUS-ICP	147	146	183	170	191	152	162	194
e	ppm	FUS-ICP	< 1	< 1	< 1	1	1	1	< 1	1
ia.	ppin	FIMS	25	12	-5	30	37	16	17	61
r r	ppo	HF/HNO3	<8.9	<8.9	<8.9	9.12	10 11	<8.9	<8.9	9 3 3
0	ppm	HF/HNO3	26.51	15.11	9.73	2.67	3.93	<0.9	<0.9	2.13
i	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	10.61
u	ppm	HF/HNO3	40.63	<19	<19	43.20	881.23	991.96	1033.22	32.03
n	ppm	HF/HNO3	29.14	47.49	111.19	148.80	130.67	105.96	130.59	222.72
S	ppm	HF/HNO3	78.27	8.05	2.59	5.78	3.43	3.49	2.15	22.39
9	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
r	ppm	HF/HNO3	240.36	246.27	239.64	258.72	272.84	279.42	263.40	284.94
lo	ppm	HF/HNO3	15.10	11.26	7.83	22.08	13.67	7.12	5.16	8.12
g	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
n	ppm	HF/HNO3	<0.9 4.52	<0.9 3 20	3.87	< 0.9	1.90	< 0.9	< 0.9	<0.9 4 85
h	ppm	HE/HNO3	1.80	0.68	1.20	1.30	1.85	1.55	1.17	10.33
e	ppm	HF/HNO3	12.11	<4 3	<4 3	<4 3	<4 3	<4 3	<4 3	<4 3
•	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
/	ppm	HF/HNO3	3.48	3.39	5.58	1.84	3.67	10.96	3.11	5.41
b	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	28.66	<25.3	<25.3	<25.3
i	ppm	HF/HNO3	14.68	2.37	7.01	0.36	1.61	0.30	0.34	2.36
b	ppm	Na2O2 Sinter	8.50	8.11	11.00	6.20	7.74	6.83	7.02	9.18
a	ppm	Na2O2 Sinter	18.16	21.70	23.82	17.33	21.24	17.82	20.00	27.70
e	ppm	Na2O2 Sinter	37.74	45.79	50.50	36.61	45.24	37.40	41.65	58.40
r	ppm	Na2O2 Sinter	4.76	5.70	6.41	4.61	5.56	4.72	5.31	7.35
d	ppm	Na2O2 Sinter	20.54	24.05	27.26	19.78	23.84	19.90	22.24	31.16
m	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 93	6 36	6.97	4 94	6.00	4 92	5.75	7.91
	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.63	1.26	1.48	1.10	1.42	0.82	1 11	2.76
u h	PDm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.05	1.20	1.40	0.04	1.12	0.02	1 11	1 30
	ppm	Na.O. Sinter	7.07	9 56	1.40	6 20	7.70	6.67	7.64	0.20
y	ppm	Na.O. Sinter	1.52	0.00	10.07	0.58	1.79	0.0/	/.04	9.39
0	ppm	No O Sinton	1.53	1.68	2.06	1.46	1.71	1.43	1.67	2.07
r	ppm	$Na_2O_2$ Sinter	4.69	5.10	6.72	4.48	5.33	4.45	5.11	6.29
m	ppm	$1Na_2O_2$ Sinter	0.70	0.74	1.00	0.70	0.80	0.66	0.73	0.95
b	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.88	5.25	7.08	4.67	5.39	4.48	5.07	6.65
1	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.74	0.81	1.12	0.70	0.81	0.72	0.80	1.00
f	ppm	Na2O2 Sinter	5.55	5.27	5.38	4.05	4.27	3.75	3.59	4.98
1	ppm	Na2O2 Sinter	0.32	0.31	0.38	0.24	0.30	0.26	0.26	0.35
h	ppm	Na2O2 Sinter	5.81	5.62	6.94	5.15	6.41	5.19	5.51	7.17
IOH WL	nm	Terraspec	2,200	2,199	2,200	2,201	2,200	2,205	2,203	2,204
IOH Depth	%	Terraspec	9.23%	9.50%	10.98%	9.85%	6.90%	16.13%	12.83%	11.10%
eOH WL	nm	Terraspec	2,254	2,253	2,253	2,251	2,253	2,252	2,252	#DIV/0
eOH Depth	%	Terraspec	2.43%	6.15%	4.05%	1.19%	1.16%	2.87%	2.33%	#DIV/0
IOH denth/FeOH denth		Terraspec	3.80	1.54	2.71	8.25	5.93	5.61	5.51	#DIV/0!

Annendix <b>B</b>	• Tabl	e R1 1	Whole-rock	Geochemistry	and Terra	snec <sup>1M</sup> Data
	• I avr				anu itiia	isince Dala

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

Qtz-Ser= Quartz-sericite Chl-Ser= Chlorite-sericite Chl=Chlorite

WL= Wavelength

	л D.	Table	D1.1 11			mennisei	y and	rerrasp		uta
Sample ID			H500678	H500679	H500680	H500681	H500682	H500683	H500684	H500685
Hole ID			BD10-87	BD10-87	BD10-87	BD10-87	BD10-77	BD10-77	BD10-77	BD10-77
Depth (m)			6.3	11	17.9	30.5	12.4	22.3	26	29.5
Lithology			FLT	FLT	FT	FLT	FLT	FBX	FLT	FLT
Alteration			Chl	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	46.6	62.04	70.79	72.73	71.37	50.47	70.96	48.35
ALO.	0/	EUS ICP	12.42	8 33	11.14	11.09	10.73	18.12	11.8	8 97
F: O (T)	/0	FUS-ICF	12.42	12.07	8.00	(15	( 21	12.00	11.0	18.20
$Fe_2O_3(1)$	70	FUS-ICP	19.04	13.07	8.06	0.15	0.51	12.09	4.35	18.30
MnO	%	FUS-ICP	0.217	0.103	0.045	0.032	0.091	0.051	0.056	0.123
MgO	%	FUS-ICP	9.14	6.21	3	1.78	2.78	1.67	3.61	5.92
CaO	%	FUS-ICP	0.06	0.04	0.05	0.04	0.14	0.14	0.09	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.08	0.05	0.2	0.22	0.19	0.33	0.2	0.1
K <sub>2</sub> O	%	FUS-ICP	0.69	0.38	2.06	2.62	2.77	5.11	2.67	1.18
TiO <sub>2</sub>	%	FUS-ICP	0.218	0.149	0.169	0.163	0.169	0.296	0.169	0.118
P <sub>2</sub> O <sub>2</sub>	%	FUS-ICP	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	0.04	0.03	0.02
101	0/2	FUS-ICP	11.71	8 12	5.42	4.69	6.18	9.42	5.07	11.47
Total	0/2	FUS-ICP	100.2	98.49	100.9	90.51	100.7	97.75	90	94.67
Ba	70	FUS-ICP	280	163	811	1054	1602	2785	1329	445
Sr.	ppin	FUS ICP	20)	105	11	11	1052	14	11	7
SI V	ppm	FUS-ICF	24	4	11	11	10	60	25	18
1 So	ppin	FUS ICP	24	*** \$	43	42	40	17	12	7
50 7r	ppm	FUS-ICP	175	122	10	120	154	220	127	116
Be	ppm	FUS-ICP	- 1	123	/ 1	130	< 1	∠30 1	- 1	< 1
V	ppm	FUS-ICP	~ 1	~ 1	~ 1	~ 1	~ 1	1	~ 1	~ 1
v Ug	ppm	FUS-ICP	16	0	ð 11	9	9	0	5	~ >
ng Cr	ppo	FINIS	-8.0	0.22	-8.0	< 3	13	/0	< 3	-8.0
Ci Ca	ppm	HF/HNO3	<0.9 0.11	9.22	<0.9 15.24	<0.9 45 49	<0.9 1.07	<0.9 1.99	< 0.9	<0.9 11.97
C0	ppm	HF/HNO3	9.11	14.76	13.34	43.46	1.07	1.00	<0.9	11.6/
NI Gu	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu Z	ppm	HF/HNO3	37.80	<19	<19	132.74	549.00	52.78	<19	31841.13
Zh	ppm	HF/HNO3	159.94	130.42	58.08	52.98	123.00	104.45	110.70	456.28
AS	ppm	HF/HNO3	15.80	3.11	1.90	19.60	12.72	24.24	<1.0	8.90
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	35.91
Br	ppm	HF/HNO3	2/8.84	254.22	247.70	1/1.0/	220.64	192.71	196.51	184.84
Mo	ppm	HF/HNO3	11.78	11.48	18.77	2.65	11.39	39.96	3.65	8.01
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	2.55
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	4.12	3.31	4.11	7.79	5.65	6.57	3.90	8.00
Sb	ppm	HF/HNO3	1.74	0.70	0.64	0.92	2.24	8.43	0.98	2.68
Te	ppm	HF/HNO3	4.98	<4.3	<4.3	<4.3	<4.3	6.88	<4.3	<4.3
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	13.30	7.71	6.89	5.63	5.23	9.37	7.16	3.14
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	31.60	<25.3	<25.3
Bi	ppm	HF/HNO3	5.14	5.23	2.09	3.59	7.01	10.73	1.59	4.52
Nb	ppm	Na2O2 Sinter	8.77	5.88	6.03	5.42	7.92	13.40	7.03	5.59
La	ppm	Na2O2 Sinter	4.38	22.54	33.51	12.97	21.44	28.92	12.54	6.28
Ce	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	9 11	48 50	71.84	28.10	44 59	62 15	26.56	12 71
Br.	ppin	Na <sub>2</sub> O <sub>2</sub> Sinter	1.15	6.21	0.16	3 50	5.67	7.76	2 26	1.64
	ppm	No O Sinter	1.15	0.21	9.10	3.39	5.07	7.70	5.50	1.04
Nd	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.86	25.54	40.38	15.91	24.49	32.85	14.65	7.10
Sm	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	1.33	6.30	9.22	4.06	5.83	8.12	3.72	1.83
Eu	ppm	Na2O2 Sinter	0.32	0.82	1.10	0.47	1.32	2.09	0.91	0.34
Tb	ppm	Na2O2 Sinter	0.43	1.08	1.29	0.88	1.11	1.46	0.75	0.44
Dv	ppm	Na2O2 Sinter	3 55	7.50	8 20	6 14	7 25	9 99	5.33	3.18
Ho	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.00	1.61	1.60	1 3/	1.60	2.26	1 25	0.74
E.	Phu	Na O Sinto	0.70	1.01	1.07	1.34	1.07	2.20	1.20	0.74
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	3.22	4.73	4.87	4.12	5.02	6.95	3.88	2.50
Tm	ppm	INa2O2 Sinter	0.56	0.69	0.70	0.62	0.79	1.07	0.59	0.42
Yb	ppm	Na2O2 Sinter	4.22	4.61	4.49	4.18	5.41	7.30	4.28	3.15
Lu	ppm	Na2O2 Sinter	0.68	0.68	0.65	0.64	0.79	1.14	0.67	0.48
Hf	ppm	Na2O2 Sinter	4.89	3.53	3.04	3.48	4.87	7.14	4.08	3.85
Та	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.31	0.22	0.22	0.22	0.30	0.46	0.28	0.21
ть.	Phil	Na.O. Sintor	0.51	4.26	1.0	4.29	6.50	0.70	5.20	4.64
	ppm	Tamara	0.09	4.50	4.62	4.38	0.84	9.86	5.95	4.64
AIOH WL	nm	1 erraspec	2,196	#DIV/0!	2,198	2,200	2,204	2,204	2,201	2,198
AIOH Depth	%	1 erraspec	2.26%	#DIV/0!	13.28%	15.10%	8.93%	11.10%	0./5%	2.89%
FEOH WL	nm	1 erraspec	2,252	2,255	2,254	2,254	#DIV/0!	#DIV/0!	2,252	2,252
reOH Depth	%	1 erraspec	/.94%	9.73%	6.02%	1.95%	#DIV/0!	#DIV/0!	2.00%	2.82%
AIOH depth/FeOH depth		1 erraspec	0.28	#DIV/0!	2.21	7.76	#DIV/0!	#DIV/0!	3.38	1.02
Abbreviations:	Units HFL= HBX:	= Hanging wa	ll flow	FT= Footwall FBX= Footwal FEI = Footwal	tuff ll breccia l flow	Alteration Qtz-Ser= Qua	rtz-sericite		Other WL= Wavelen	gth
	FLT=	Footwall lapi	lli tuff	INT= Intrusion	1	Chl=Chlorite				

	л D.	Labic	D1.1 W1			nemisti	y anu i	l ci i asp		ata
Sample ID			H500686	H500687	H500688	H500689	H500690	H500691	H500692	H500693
Hole ID			BD10-77	BD10-76	BD10-76	BD10-76	BD10-76	BD10-63	BD10-63	BD10-63
Depth (m)			49.1	48.2	36	27.2	77	11.4	21.3	37.5
Lithology			FLT	FLT	FLT	FI T	FLT	HBX	HFI	FLT
Alteration			Ota Sar	ChilSon	Ota Sar	Chl	Ota San	Ota Son	Ota Sar	Ota San
Alteration	0 <i>i</i>		Qiz-Sei	CIII-Sei	Qiz-Sei	CIII	Qtz-Sei	Qtz-Sei	Qiz-Sei	Qiz-Sei
SiO <sub>2</sub>	%	FUS-ICP	80.12	69.63	77.39	35.35	76.08	75.18	69.35	67.18
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	10.49	8.46	9.65	12.86	7.96	9.55	9.13	13.06
$Fe_2O_2(T)$	%	FUS-ICP	2 75	10.3	4 29	22.93	7 56	2.5	7 65	5 71
MnO	0/.	EUS ICP	0.012	0.071	0.022	0.154	0.025	0.225	0.042	0.158
MIIO	70	FUS-ICF	0.012	0.071	0.025	0.134	0.023	0.225	0.043	0.138
MgO	%	FUS-ICP	0.88	3.91	0.77	12.64	1.09	2.48	0.59	2.92
CaO	%	FUS-ICP	0.06	0.05	0.07	0.06	0.04	0.16	0.05	0.08
Na <sub>2</sub> O	%	FUS-ICP	0.2	0.13	0.22	0.04	0.15	0.12	0.12	0.18
K-0	%	FUS-ICP	2 75	1.18	2 57	0.38	2.07	27	2 71	3 37
7.0		FUG ICD	0.150	0.100	0.145	0.170	2.07	0.107	2.71	0.010
1102	%	FUS-ICP	0.152	0.128	0.145	0.178	0.112	0.127	0.105	0.212
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.02	0.01	0.02	< 0.01	< 0.01	0.02	< 0.01	0.03
LOI	%	FUS-ICP	2.79	6.26	3.66	14.75	5.29	4.22	5.51	5.69
Total	%	FUS-ICP	100.2	100.1	98.82	99.35	100.4	97.31	95.25	98.59
Ba	nnm	FUS-ICP	949	431	945	156	918	1611	1999	2138
S.	ppin	EUS ICP	0	6	10	2	7	7	6	2150
31	ppm	FUS-ICF	25	0	10	3	20	25	0	/
Ŷ	ppm	FUS-ICP	35	33	35	4/	28	35	29	40
Sc	ppm	FUS-ICP	11	11	11	16	8	9	7	13
Zr	ppm	FUS-ICP	124	110	118	150	106	110	128	142
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	1	< 1	< 1
V	ppm	FUS-ICP	5	< 5	< 5	< 5	7	5	6	10
Hg	ppb	FIMS	< 5	< 5	< 5	6	6	393	3170	102
Cr	ppp	HF/HNO3	< 8.9	<8.9	< 8.9	< 8 9	<89	12 24	<8.9	< 8.9
с. Со	Ppm	HE/HNO2	10.42	52 74	-0.7	-0.7 82 10	14 74	<0.0	<0.0	1 27
C0	ppm	HF/HNO3	19.45	32.74	10.20	05.10	14.74	<0.9	<0.9	1.57
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	301.78	73.83	52.16	35.83	576.22	140.39	2416.34	788.50
Zn	ppm	HF/HNO3	33.87	39.48	27.71	201.16	99.93	4684.56	29897.05	453.44
As	ppm	HF/HNO3	2.22	15.03	3.11	13.47	8.30	4.80	126.17	15.04
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	35.33	<26.7	<26.7	<26.7	<26.7
Br	nnm	HE/HNO3	200.28	206.82	187.45	205 52	253.93	181 51	177.63	164.87
Di Ma	ppin	HE/INO2	12.04	200.82	10 42	50.27	7 46	6.07	5 20	4 20
NIO	ppm	HF/HNO5	12.04	2.16	10.45	50.57	7.40	0.07	3.39	4.50
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	4.18	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	14.33	95.10	1.35
Sn	ppm	HF/HNO3	3.47	3.67	4.23	3.51	3.77	3.24	8.26	3.58
Sb	ppm	HF/HNO3	0.94	1.80	0.80	0.91	0.99	3.96	12.11	6.39
Те	ppm	HF/HNO3	<4.3	6.93	4.74	8.18	<4.3	<4.3	<4.3	<4.3
I	nnm	HE/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppin	HE/UNO2	2 79	5.60	3 70	867	2 29	3.06	6.17	3 36
W DI	ppm		5.76	5.09	5.70	6.02	5.56	501.85	2404 (2	5.50
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	501.85	3494.63	59.91
Bi	ppm	HF/HNO3	2.29	8.37	7.69	12.87	1.86	<0.1	4.34	1.65
Nb	ppm	Na2O2 Sinter	5.23	5.29	5.41	8.86	4.59	5.75	5.12	7.61
Ia	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	21.45	16.63	16.81	20.37	12.04	14 30	1/1.18	20.02
La	ppin	No O Sintor	21.45	10.05	10.01	20.57	12.04	14.50	14.10	20.02
Ce	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	46.55	35.48	34.37	42.72	26.05	31.03	30.79	42.45
Pr	ppm	Na2O2 Sinter	6.05	4.58	4.35	5.32	3.31	3.92	3.90	5.35
Nd	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	25.42	19.76	18 76	23.09	13.66	16.97	16.48	22.98
- ····	Phu	Na O Sinter	23.72	17.70	4.20	20.07	2.00	10.77	10.40	22.70
Sm	ppm	ina <sub>2</sub> O <sub>2</sub> Sinter	5.77	4.74	4.29	5.83	5.61	4.09	4.08	5.41
Eu	ppm	Na2O2 Sinter	0.90	0.55	0.70	1.13	0.39	0.78	1.23	1.12
Tb	ppm	Na2O2 Sinter	0.92	0.83	0.83	1.20	0.67	0.88	0.79	1.03
Dv		Na.O. Sinter	5 07	5 50	5 16	Q 40	1 15	5.00	5 41	6.01
Dy	ppm		3.87	3.38	3.40	0.48	4.40	5.90	5.41	0.91
Но	ppm	$Na_2O_2$ Sinter	1.28	1.21	1.19	1.84	0.87	1.31	1.16	1.50
Er	ppm	Na2O2 Sinter	3.86	3.69	3.62	5.49	2.74	4.00	3.48	4.53
Tm	pom	Na <sub>2</sub> O <sub>2</sub> Sinter	0.59	0.55	0.58	0.94	0.29	0.62	0.52	0.60
1111	phu	Na O Sint	0.58	0.55	0.38	0.64	0.38	0.02	0.33	0.09
Yb	ppm	$1Na_2O_2$ Sinter	4.17	3.64	3.88	5.41	2.67	4.17	3.86	4.74
Lu	ppm	Na2O2 Sinter	0.62	0.58	0.58	0.86	0.45	0.65	0.59	0.73
Hf	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 14	3 64	4 07	4 70	3 14	3.67	3 91	4 59
····	Phu	No O Sint-	7.17	0.07	4.07	4.70	0.17	5.07	0.01	4.57
1a	ppm	$1Na_2O_2$ Sinter	0.23	0.22	0.22	0.31	0.19	0.21	0.21	0.27
Th	ppm	Na2O2 Sinter	5.35	4.57	5.01	6.27	3.49	4.89	4.84	5.78
AlOH WL	nm	Terraspec	2,202	2,199	2,203	#DIV/0!	2,202	2,208	2,207	2,205
AlOH Denth	%	Terraspec	12 75%	7 86%	10 58%	#DIV/01	5 73%	13 20%	6 97%	9.01%
FeOH WI	70	Terracipoo	2.1570	2 252	#DIV/01	2 251	#DIV/01	2 2 5 2	#DIV/01	2.0170
EaOH Donth	0/	Tomaspee	2,230	2,233	#DIV/0!	6.240/	#D11/0	1,232	#DIV/01	1,204
reOH Depth	70	1 erraspec	1.04%	/.08%	#DIV/0!	0.24%	#DIV/0!	1.31%	#DIV/0!	1.00%
AIOH depth/FeOH depth		Terraspec	12.26	1.11	#DIV/0!	#DIV/0!	#DfV/0!	10.11	#DIV/0!	5.44
Abbreviations:	Units	;		FT= Footwall t	uff	Alteration			Other	
	HFL=	<ul> <li>Hanging wal</li> </ul>	l flow	FBX= Footwal	l breccia	Qtz-Ser= Quar	tz-sericite		WL= Wavelen	gth
	HBX	= Hanging wal	l breccia	FFL= Footwall	flow	Chl-Ser= Chlo	rite-sericite			
	FI T=	Footwall land	li tuff	INT= Intrusion		Chl=Chlorite				
	1.1.1 -	1 OOtwall lapli		11 T - HILL USION		Cin-Cinorite				

	A D.	I abic I	<b>71.1 711</b>		Geoc	nemistry	anu i	crraspe		ia
Sample ID			H500694	H500695	H500696	H500697	H500698	H500699	H500700	H501201
Hole ID			BD10-63	BD10-63	BD10-63	BD10-63	BD10-61	BD10-61	BD10-61	BD10-61
Depth (m)			57.3	59.1	64.1	75.4	11.1	24.9	37.9	39.5
Lithology			FLT	FBX	FFL	FBX	FLT	FLT	FLT	FBX
Alteration			Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl	Chl-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	48.39	75.63	76.1	74.08	66.13	55.32	67.31	70.52
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	6.36	10.83	10.88	10.17	7.18	12.26	8.27	8.18
$Fe_2O_3(T)$	%	FUS-ICP	23	4.32	3.78	5.89	14.65	7.5	11.4	9.1
MnO	%	FUS-ICP	0.099	0.056	0.048	0.044	0.012	0.204	0.068	0.061
MgO	%	FUS-ICP	4.35	2.41	2.53	1.95	0.29	12.97	5.37	4.94
CaO	%	FUS-ICP	0.05	0.13	0.08	0.08	0.04	0.88	0.04	0.05
Na <sub>2</sub> O	%	FUS-ICP	0.06	0.18	0.18	0.17	0.13	0.06	0.06	0.07
K <sub>2</sub> O	%	FUS-ICP	0.5	2.7	2.73	2.51	2.05	0.52	0.48	0.68
TiO <sub>2</sub>	%	FUS-ICP	0.084	0.177	0.173	0.166	0.107	0.387	0.167	0.141
P <sub>2</sub> O <sub>2</sub>	%	FUS-ICP	< 0.01	0.03	0.03	0.02	0.02	0.08	< 0.01	0.02
LOI	%	FUS-ICP	12 54	3.99	4 17	4 94	8 38	8.92	6.01	5
Total	%	FUS-ICP	95.44	100.4	100.7	100	98.98	99.11	99.18	98 75
Ba	ppm	FUS-ICP	215	1156	1103	1059	1592	312	208	308
Sr	ppm	FUS-ICP	4	8	7	8	4	12	2	3
Y	ppm	FUS-ICP	16	36	40	29	24	32	31	30
Sc	ppm	FUS-ICP	5	9	11	10	7	16	9	8
Zr	ppm	FUS-ICP	79	134	133	126	89	119	102	105
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	< 5	8	< 5	< 5	< 5	77	11	5
Hg	ppb	FIMS	42	10	< 5	< 5	52	13	11	12
Cr	ppm	HF/HNO3	<8.9	<8.9	<8.9	<8.9	28.66	58.80	10.50	<8.9
Co	ppm	HF/HNO3	30.32	1.09	<0.9	1.73	9.53	5.01	43.73	27.87
Ni	ppm	HF/HNO3	<10	<10	<10	<10	20.12	18.11	<10	<10
Cu	ppm	HF/HNO3	26390.83	504.50	145.31	62.86	1270.79	25.55	529.14	402.61
Zn	ppm	HF/HNO3	222.84	86.13	71.48	202.89	290.41	365.36	81.92	72.94
As	ppm	HF/HNO3	30.49	3.09	5.05	13.04	72.59	<1.6	21.25	3.76
Se	ppm	HF/HNO3	43.17	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	173.88	167.92	185.64	189.40	162.49	197.07	185.95	217.70
Mo	ppm	HF/HNO3	10.47	15.41	2.47	4.36	12.37	1.99	14.91	8.62
Ag	ppm	HF/HNO3	2.80	<1	<1	<1	1.02	<1	<1	<1
Ca	ppm	HF/HNO3	< 0.9	< 0.9	< 0.9	< 0.9	<0.9	< 0.9	< 0.9	<0.9
Sn	ppm	HF/HNO3	3.11	4.15	3.27	3.30	3.57	3.39	2.79	2.83
SD To	ppm	HF/HNO3	2.60	1./4	1.22	1.85	5.54	0.72	0.62	0.84
1e	ppm	HF/HNO3	10.94	<4.5	<4.5	<4.5	0.58	<4.5	0.73	<4.5
1 W	ppin	HE/HNO3	4.97	1.51	1 00	1.56	2.44	2 53	867	2.26
Ph	ppm	HE/HNO3	27.75	<25.3	<25.3	<25.3	132.00	67.27	<25.3	<25.3
Bi	nnm	HF/HNO3	17.94	1 73	0.68	1.36	6.06	0.78	6.90	3 13
Nb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.22	7.25	6.45	5.48	4.14	6.90	5.08	4 79
INU L-	ppm	Na O Sinter	4.22	10.09	16.50	16.11	4.14	0.90	5.08	4.79
La	ppm	Na O Sinter	8.95	19.08	16.50	10.11	12.25	12.62	10.67	11.13
Ce	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	17.98	41.02	34.55	33.92	25.58	28.17	22.20	22.95
Pr	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	2.26	5.05	4.35	4.24	3.15	3.58	2.75	2.95
Nd	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	9.52	21.09	18.33	18.03	13.52	14.83	11.71	12.20
Sm	ppm	Na2O2 Sinter	2.35	5.07	4.51	4.32	3.46	3.81	3.11	3.06
Eu	ppm	Na2O2 Sinter	0.50	0.98	0.91	0.75	0.87	1.05	0.52	0.55
Tb	ppm	Na2O2 Sinter	0.49	0.90	0.92	0.79	0.64	0.78	0.75	0.72
Dv	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	3 51	5.91	6 35	5.18	4 35	5 54	5.12	4 75
Но	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.77	1.28	1 30	1.06	0.93	1.20	1.02	0.98
Fr.	ppin	Na.O. Sinter	2 27	2.97	4.04	2.16	2.74	2.60	2 20	2.80
EI Ter	ppm	Na O Sinter	2.27	5.67	4.04	5.10	2.74	5.09	5.20	2.89
Im	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.36	0.59	0.60	0.48	0.42	0.58	0.48	0.45
Yb	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	2.47	4.00	4.09	3.32	2.83	3.87	3.32	2.96
Lu	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	0.39	0.61	0.65	0.51	0.45	0.61	0.51	0.45
Hf	ppm	Na2O2 Sinter	2.35	4.07	4.29	3.60	2.95	4.18	2.92	2.85
Та	ppm	Na2O2 Sinter	0.15	0.26	0.26	0.22	0.17	0.25	0.19	0.19
Th	ppm	Na2O2 Sinter	3.19	5.60	5.19	4.85	3.65	4.53	3.81	3.92
AIOH WL	nm	Terraspec	2,197	2,205	2,204	2,202	2,206	#DIV/0!	2,198	2,202
AlOH Depth	%	Terraspec	1.68%	9.43%	17.00%	13.63%	6.12%	#DIV/0!	2.74%	8.61%
FeOH WL	nm	Terraspec	2,251	2,251	2,252	2,255	#DIV/0!	2,250	2,253	2,253
FeOH Depth	%	Terraspec	4.00%	1.12%	2.64%	1.16%	#DIV/0!	3.44%	6.41%	7.46%
AlOH depth/FeOH depth		Terraspec	0.42	8.44	6.45	11.75	#DIV/0!	#DIV/0!	0.43	1.15
Abbreviations:	Units HFL= HBX:	<ul> <li>Hanging wall f</li> <li>Hanging wall b</li> </ul>	low reccia	FT= Footwall tu FBX= Footwall FFL= Footwall	ff breccia flow	Alteration Qtz-Ser= Quart Chl-Ser= Chlori	z-sericite ite-sericite		Other WL= Wavelen	gth
	FLT=	rootwall lapilli	turf	IN I = Intrusion		Chi=Chlorite				

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Appendix B: Table B1.1	Whole-rock	Geochemistry	and Terraspec	Data

Appendix	<b>D</b> :	i able D1.	1 VV IIU	IE-IOCK	Geoci	remistr	y anu i	errasp	ec D	ala
Sample ID			H501202	H501203	H501204	H501205	H501206	H501207	H501208	H501212
Hole ID			BD10-61	BD10-58	BD10-58	BD10-58	BD10-58	BD10-58	BD10-58	BD10-40
Depth (m)			44.4	8.2	15	18.2	29.5	32.3	34.9	4
Lithology			FBX	FLT	FLT	FLT	FLT	FBX	FLT	HFL
Alteration			Chl-Ser	Chl-Ser	Chl	Chl-Ser	Chl-Ser	Chl-Ser	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	68.61	59.01	15.62	65.17	47.86	57.12	67.85	74.58
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	8.9	6.46	13.37	7.92	7.41	4.37	7.63	10.69
$Fe_2O_3(T)$	%	FUS-ICP	6.54	18.33	41.47	15.57	23.44	21.52	11.82	5.99
MnO	%	FUS-ICP	0.261	0.046	0.176	0.035	0.096	0.037	0.049	0.078
MgO	%	FUS-ICP	4.44	3.31	9.62	0.79	3.12	0.8	1.36	1.29
CaO	%	FUS-ICP	1.73	0.05	0.05	0.06	0.05	0.04	0.05	0.05
Na <sub>2</sub> O	%	FUS-ICP	0.11	0.08	0.05	0.19	0.1	0.08	0.15	0.22
K <sub>2</sub> O	%	FUS-ICP	1.46	0.64	0.33	2.02	0.8	0.86	1.43	2.8
TiO <sub>2</sub>	%	FUS-ICP	0.148	0.107	0.208	0.118	0.081	0.052	0.092	0.134
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.01	< 0.01	0.01	< 0.01	< 0.01	< 0.01	< 0.01	0.02
LOI	%	FUS-ICP	5.99	9.87	19.31	8.4	14.26	11.16	7.3	4.75
Total	%	FUS-ICP	98.19	97.9	100.2	100.3	97.2	96.04	97.73	100.6
Ba	ppm	FUS-ICP	638	375	167	966	372	380	625	1320
Sr	ppm	FUS-ICP	28	3	3	10	5	5	7	8
Y	ppm	FUS-ICP	31	15	27	19	17	10	25	55
Sc	ppm	FUS-ICP	10	7	12	9	7	4	10	10
Zr	ppm	FUS-ICP	105	69	169	103	76	54	88	140
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	< 5	12	27	16	10	< 5	6	7
Hg	ppb	FIMS	6	143	115	111	61	344	33	16
Cr	ppm	HF/HNO3	25.64	15.55	<8.9	9.25	9.49	15.08	<8.9	<8.9
C0 Ni	ppm	HF/HNO3	12.77	83.27	45.54	20.24	85.57	5.38	14.42	59.97
NI Cu	ppm	HE/HNO3	22.32	1254 50	6775 10	7002.28	16384.00	16023 74	2028 48	2118 22
Zn Zn	ppm	HE/HNO3	65.25	66.97	205 22	79.35	130.89	182 75	69.21	10.22
As	nnm	HF/HNO3	3 37	37.57	303.63	40.52	80.96	236.23	16.71	30.08
Se	ppm	HF/HNO3	<26.7	67.46	73.96	43.53	75.58	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	175.61	187.03	150.81	165.52	154.26	166.20	179.81	255.44
Mo	ppm	HF/HNO3	12.16	49.06	59.84	26.72	56.11	13.99	14.84	11.06
Ag	ppm	HF/HNO3	<1	<1	1.31	<1	2.20	3.65	<1	<1
Cd	ppm	HF/HNO3	<0.9	< 0.9	<0.9	<0.9	<0.9	<0.9	<0.9	< 0.9
Sn	ppm	HF/HNO3	2.83	3.14	4.51	4.55	4.18	5.36	5.05	4.13
Sb	ppm	HF/HNO3	0.91	1.33	4.39	2.55	4.94	36.48	1.18	4.75
Te	ppm	HF/HNO3	<4.3	9.85	45.12	11.85	248.50	10.17	6.24	11.08
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.07	8.44	15.84	4.40	6.50	3.46	2.49	3.57
Pb Di	ppm	HF/HNO3	<25.3	<25.3	50.63	<25.3	20.37	50.08	<25.5	<25.3
BI	ppm	HF/HNU5 No2O2 Sinton	2.10	9.34	05.44	19.66	3/4.70	19.50	9.30	15.66
Nb	ppm	Na2O2 Sinter	6.12	3.03	8.03	4.09	3.37	2.72	4.63	7.65
La	ppm	Na2O2 Sinter	18.58	8.39	21.00	10.45	9.14	6.49	8.81	20.30
Ce	ppm	Na2O2 Sinter	38.35	16.83	42.87	22.48	19.15	13.76	18.65	42.90
Pr	ppm	Na2O2 Sinter	4.86	2.08	5.26	2.87	2.44	1.74	2.39	5.41
Nd	ppm	Na2O2 Sinter	20.43	8.53	22.15	12.41	10.28	7.25	9.94	22.38
Sm	ppm	Na2O2 Sinter	4.83	1.84	4.96	3.01	2.70	1.77	2.63	5.87
Eu	ppm	Na2O2 Sinter	0.61	0.27	0.89	0.77	0.40	0.53	0.76	0.74
Tb	ppm	Na2O2 Sinter	0.86	0.36	0.82	0.52	0.52	0.28	0.66	1.26
Dv	ppm	Na2O2 Sinter	5.70	2.31	5.81	3.54	3.57	2.05	4.38	8.67
Но	nnm	Na2O2 Sinter	1 23	0.52	1 27	0.78	0.77	0.42	1.02	1.84
Fr	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	3 50	1.64	4.14	2.36	2.45	1.33	3.22	5.81
Tm	ppin	Na2O2 Sinter	0.54	0.26	0.65	0.28	0.27	0.20	0.51	0.89
1 III Vl	ppm	Na2O2 Sinter	2.91	1.82	4.70	0.58	0.37	0.20	0.51	0.88
YD	ppm	Na2O2 Sinter	3.81	1.83	4.72	2.64	2.56	1.49	3.52	6.02
Lu	ppm	Na2O2 Sinter	0.55	0.30	0.73	0.39	0.42	0.25	0.58	0.96
Hf	ppm	Na2O2 Sinter	3.88	2.45	4.83	3.34	3.36	2.02	3.51	4.49
Та	ppm	Na2O2 Sinter	0.23	0.10	0.29	0.16	0.14	0.11	0.20	0.28
Th	ppm	Na2O2 Sinter	4.88	2.43	6.59	3.64	3.25	2.23	3.74	5.04
AlOH WL	nm	Terraspec	2,202	2,202	#DIV/0!	2,201	2,204	2,202	2,200	2,203
AlOH Depth	%	Terraspec	7.13%	1.56%	#DIV/0!	3.87%	1.91%	3.22%	2.24%	13.80%
FeOH WL	nm	Terraspec	2,253	2,253	2,253	#DIV/0!	2,255	#DIV/0!	2,257	#DIV/0!
FeOH Depth	%	Terraspec	4.50%	2.26%	2.06%	#DIV/0!	1.37%	#DIV/0!	1.57%	#DIV/0!
AIOH depth/FeOH depth		Terraspec	1.58	0.69	#DIV/0!	#DIV/0!	1.39	#DIV/0!	1.43	#DIV/0!
Abbrariations	17-14			ET= E - · · · · ·	t., ff	Altonetter			Othen	
AUDIEVIAUDIIS.	HFI =	= Hanging wall flow	w	FBX= Footwall	ull breccia	Otz-Ser= Oug	rtz-sericite		WI = Waveley	noth
	HRX	= Hanging wall bre	 ccia	FFL= Footwal	ll flow	Chl-Ser= Chlo	rite-sericite		wavele	.5
	FLT=	Footwall lanilli tut	ff	INT= Intrusio	n	Chl=Chlorite				

		~	TN	1
Annondiv R. Tahla R11	Whole reals (	'aachamistry c	nd Torrognoot"	* Nata
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Appendix	D: 1	able D1.1	whole-rock Geochemistry and Terrasp						Jata	
Sample ID			H501213	H501214	H501215	H501216	H501217	H501218	H501219	H501220
Hole ID			BD10-40	BD10-40	BD10-40	BD10-40	BD10-98	BD10-98	BD10-98	BD10-98
Depth (m)			9.5	30.5	43.7	16.4	13.7	23	40.4	42.8
Lithology			HBX	HBX	FLT	HBX	FLT	FLT	FLT	FFL
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl-Ser	Chl	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	65.39	67.62	69.79	76.73	21.26	60.94	62.32	72.7
AlaOa	%	FUS-ICP	12 23	13 34	10.26	11.11	18.1	14 73	10.43	13
Fe O (T)	0/2	FUS ICP	7.0/	1 88	7 21	2 79	24.73	6.29	9.78	4.15
MrO	0/	FUS-ICF	0.195	9.00	0.071	0.122	0.222	0.102	0.129	4.15
MinO	%0 0/	FUS-ICP	0.185	0.145	1.07	0.132	0.233	0.102	0.128	0.065
MgO	%0 0/	FUS-ICP	2.41	2.99	1.97	1.79	10.23	/.9/	9.63	2.44
CaO	%o	FUS-ICP	0.07	0.14	0.06	0.28	0.09	0.31	0.06	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.28	0.47	0.39	0.24	0.61	0.27	0.06	0.23
K <sub>2</sub> O	%	FUS-ICP	3.01	3.44	2.63	2.97	0.71	2.19	0.5	3.28
TiO <sub>2</sub>	%	FUS-ICP	0.156	0.168	0.149	0.14	0.242	0.243	0.167	0.182
$P_2O_5$	%	FUS-ICP	0.02	0.02	0.01	0.03	0.01	0.03	0.02	0.03
LOI	%	FUS-ICP	6.85	6.76	6.91	3.86	17.88	6.96	7.63	4.41
Total	%	FUS-ICP	98.54	99.97	99.47	100.1	100.1	100	100.7	100.5
Ba	ppm	FUS-ICP	1367	1377	832	1295	391	2759	202	1444
Sr	ppm	FUS-ICP	12	25	17	12	34	27	5	9
Y	ppm	FUS-ICP	45	52	35	41	70	46	38	50
Sc	ppm	FUS-ICP	11	12	9	11	17	20	10	14
Zr	ppm	FUS-ICP	153	180	132	152	250	168	145	175
Be	ppm	FUS-ICP	1	1	< 1	1	< 1	< 1	< 1	1
V	ppm	FUS-ICP	7	6	10	7	11	42	13	7
Hg	ppb	FIMS	9	19	23	9	68	20	6	8
Cr	ppm	HF/HNO3	32.25	<8.9	<8.9	13.63	<8.9	<8.9	9.09	<8.9
Co	ppm	HF/HNO3	3.11	2.47	9.56	<0.9	16.42	0.94	8.00	3.11
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	55.96	50.21	275.50	21.99	198.95	<19	38.83	233.80
Zn	ppm	HF/HNO3	287.07	231.26	128.79	130.56	1109.40	238.01	202.13	65.67
As	ppm	HF/HNO3	9.96	6.56	18.05	12.47	79.21	<1.6	2.69	10.90
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	257.37	244.01	235.41	246.97	240.93	218.08	252.34	236.83
Mo	ppm	HF/HNO3	8.96	3.11	6.62	5.93	3.10	4.80	4.88	3.98
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	3.14	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	3.92	5.47	5.35	3.22	3.46	5.79	3.21	4.62
Sb	ppm	HF/HNO3	1.39	1.79	1.37	1.14	1.29	0.69	0.93	1.68
Te	ppm	HF/HNO3	<4.3	<4.3	5.93	<4.3	<4.3	<4.3	<4.3	<4.3
l	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	3.41	4.74	9.59	5.55	6.01	6.54	1.91	4.00
Pb	ppm	HF/HNO3	<25.3	<25.3	33.28	<25.3	<25.3	<25.3	<25.3	28.18
Bi	ppm	HF/HNO3	1.34	0.62	8.32	0.59	4.02	0.47	1.58	1.01
Nb	ppm	Na2O2 Sinter	7.58	9.54	8.43	9.83	15.55	10.72	9.17	10.96
La	ppm	Na2O2 Sinter	21.14	19.82	15.73	19.78	27.67	19.41	18.78	22.56
Ce	ppm	Na2O2 Sinter	43.82	41.90	32.77	41.52	55.42	40.30	39.77	48.20
Pr	ppm	Na2O2 Sinter	5.50	5.32	4.06	5.22	6.86	5.12	5.05	6.08
Nd	nnm	Na2O2 Sinter	23.26	22.23	17.80	22.24	29.77	21.79	22.04	25 50
Sm	ppm	Na2O2 Sinter	5.07	5.51	4 16	5 41	7 26	5.40	5.49	6.15
Sin	ppm	No2O2 Sinter	3.97	5.51	4.10	5.41	7.50	3.40	5.48	0.15
Eu	ppm	Na2O2 Sinter	1.15	0.96	0.57	0.94	1.19	1.98	0.83	1.08
ТЬ	ppm	Na2O2 Sinter	1.22	1.19	0.83	0.99	1.73	1.13	1.00	1.28
Dy	ppm	Na2O2 Sinter	8.12	8.21	5.77	6.70	11.70	7.74	6.53	8.51
Но	ppm	Na2O2 Sinter	1.69	1.68	1.19	1.43	2.45	1.60	1.42	1.78
Er	ppm	Na2O2 Sinter	5.20	5.17	3.67	4.33	7.50	5.10	4.18	5.52
Tm	nnm	Na2O2 Sinter	0.75	0.79	0.57	0.68	1 13	0.77	0.64	0.81
Vh	ppm	Na2O2 Sinter	5 20	5 50	2.02	4.60	7 97	5.10	4.44	5.65
10	ppm	No2O2 Sinter	5.29	5.50	5.52	4.09	1.00	5.10	4.44	5.05
Lu	ppm	Na2O2 Sinter	0.80	0.87	0.57	0.70	1.20	0.77	0.67	0.87
Hf	ppm	Na2O2 Sinter	5.15	4.82	4.50	6.64	7.09	6.41	4.75	6.15
Та	ppm	Na2O2 Sinter	0.32	0.35	0.30	0.36	0.58	0.40	0.32	0.40
Th	ppm	Na2O2 Sinter	5.87	6.16	7.88	7.77	12.23	8.16	7.64	8.09
AlOH WL	nm	Terraspec	2,202	2,203	2,201	2,204	2,190	2,200	2,195	2,203
AlOH Depth	%	Terraspec	7.52%	9.53%	17.75%	13.10%	3.08%	2.70%	3.02%	12.55%
FeOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	2,254	2,252	2,251	2,251	2,253
FeOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	1.49%	4.24%	2.09%	9.66%	2.45%
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	8.81	0.73	1.29	0.31	5.12
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	<ul> <li>Hanging wall flow</li> </ul>		FBX= Footwa	all breccia	Qtz-Ser= Quar	tz-sericite		WL= Wavele	ngth
	HBX	= Hanging wall breed	cia	FFL= Footwa	ll flow	Chl-Ser= Chlo	rite-sericite			

Chl=Chlorite

FLT= Footwall lapilli tuff

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Abbendix B:	I able BLI	wnoie-rock	<b>U</b> teocnemistry	and Terras	Dec Data

Appendix	<b>B:</b> ]	<b>Fable B1</b>	.1 Who	le-rock	Geocl	hemistr	y and T	<b>Ferraspec<sup>TM</sup> Data</b>		
Sample ID			H501221	H501222	H501223	H501224	H501225	H501226	H501227	H501228
Hole ID			BD10-98	BD10-50	BD10-50	BD10-50	BD10-56	BD10-56	BD10-56	BD10-56
Lithology			50.8 FL T	9 HFI	15.6 FLT	50.6 FLT	7.4 FLT	21.2 FLT	29.9 FL T	41.4 FLT
Alteration			Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl	Chl	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	74.1	54.83	74.15	70.79	72.28	32.36	37.01	67.44
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	10.86	13.1	10.56	12.06	13.9	17.04	9.97	8.67
$Fe_2O_3(T)$	%	FUS-ICP	4.48	13.69	7.04	4.86	2.63	8.69	24.22	10.11
MnO	%	FUS-ICP	0.095	0.07	0.019	0.086	0.044	0.318	0.114	0.048
MgO	%	FUS-ICP	2.3	4.75	0.32	2.68	2.98	24.79	11.18	1.4
CaO	%	FUS-ICP	0.04	0.05	0.08	0.07	0.07	1.33	0.17	0.05
Na <sub>2</sub> O	%	FUS-ICP	0.19	0.22	0.23	0.23	0.29	0.02	0.03	0.19
K <sub>2</sub> O	%	FUS-ICP	2.72	2.41	2.76	2.79	3.22	0.03	0.13	2.08
TiO <sub>2</sub>	%	FUS-ICP	0.154	0.162	0.161	0.177	0.182	0.243	0.15	0.132
$P_2O_5$	%	FUS-ICP	< 0.01	0.01	0.02	0.03	0.02	0.1	0.02	0.01
LOI	%	FUS-ICP	4.45	9.44	4.75	4.56	4.27	14.96	16.08	7.54
Total	%	FUS-ICP	99.39	98.73	100.1	98.33	99.89	99.89	99.09	97.67
Ba S-	ppm	FUS-ICP	1217	1441	1193	1035	2370	30	69	971
SI V	ppm	FUS-ICP	37	12	38	42	42	38 78	4	33
Sc	ppm	FUS-ICP	11	13	11	12	14	16	10	9
Zr	ppm	FUS-ICP	130	158	142	141	156	205	125	104
Be	ppm	FUS-ICP	1	< 1	< 1	1	1	< 1	< 1	< 1
V	ppm	FUS-ICP	8	< 5	17	6	16	12	< 5	< 5
Hg	ppb	FIMS	< 5	333	23	< 5	22	16	85	104
Cr	ppm	HF/HNO3	<8.9	<8.9	11.54	<8.9	<8.9	<8.9	<8.9	<8.9
Co	ppm	HF/HNO3	4.41	6.04	1.99	3.61	<0.9	1.28	47.23	1.12
NI Cu	ppm	HF/HNO3	398.02	108.98	1167.01	33.62	22.82	42 54	541.56	3173 38
Zn	ppm	HF/HNO3	58 19	151 30	70.33	84.13	107.68	474 89	127 19	150.08
As	ppm	HF/HNO3	3.58	17.89	10.62	9.48	4.35	1.96	46.02	47.86
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	50.19	<26.7
Br	ppm	HF/HNO3	228.50	142.76	142.58	146.94	157.50	141.61	151.63	132.76
Mo	ppm	HF/HNO3	3.58	7.57	3.15	5.47	1.01	4.04	30.74	4.81
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Ca Sn	ppm	HF/HNO3	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9	< 0.9
Sh	ppm	HF/HNO3	2.18	7.16	1.58	1.69	1 10	9.29	9.05	5 27
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	4.59	<4.3
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	3.70	1.71	10.87	1.74	3.77	1.17	3.32	4.30
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HF/HNO3	0.42	5.33	3.74	0.80	0.22	0.42	14.65	8.27
Nb	ppm	Na2O2 Sinter	7.80	8.10	6.52	7.50	8.51	11.47	12.56	5.98
La	ppm	Na2O2 Sinter	18.94	19.53	18.55	20.65	23.76	32.82	20.65	12.24
Ce	ppm	Na2O2 Sinter	38.54	40.85	37.15	42.75	48.50	67.33	43.03	25.33
Pr	ppm	Na2O2 Sinter	4.80	5.35	4.73	5.44	6.19	8.43	5.45	3.19
Nd	ppm	Na2O2 Sinter	19.59	22.78	20.25	22.93	26.69	36.34	23.78	13.42
Sm	ppm	Na2O2 Sinter	4.43	5.62	4.88	5.79	6.64	8.81	5.80	3.88
Eu	ppm	Na2O2 Sinter	1.04	1.12	0.69	1.13	1.53	2.29	0.95	0.93
Tb	ppm	Na2O2 Sinter	0.92	1.20	0.93	1.13	1.15	1.97	1.11	0.87
Dy	ppm	Na2O2 Sinter	5.79	8.00	6.08	7.52	7.48	13.19	7.51	5.89
Но	ppm	Na2O2 Sinter	1.22	1.61	1.24	1.57	1.57	2.85	1.59	1.16
Er	ppm	Na2O2 Sinter	3.78	5.05	3.88	4.74	4.91	8.49	3.70	3.41
Tm	ppm	Na2O2 Sinter	0.60	0.74	0.60	0.73	0.75	1.23	0.75	0.56
Yb	ppm	Na2O2 Sinter	4.02	4.91	4.10	4.82	5.10	8.25	5.22	3.76
Lu	ppm	Na2O2 Sinter	0.61	0.79	0.64	0.75	0.82	1.29	0.81	0.67
Hf	ppm	Na2O2 Sinter	4.83	6.26	4.34	5.30	5.91	6.46	6.46	3.83
Та	ppm	Na2O2 Sinter	0.30	0.31	0.25	0.31	0.33	0.43	0.43	0.23
Th	ppm	Na2O2 Sinter	6.09	6.37	5.13	5.81	6.45	9.14	8.16	4.24
AlOH WL	nm	Terraspec	2,204	2,196	2,201	2,201	2,200	#DIV/0!	#DIV/0!	2,203
AlOH Depth	%	Terraspec	13.20%	2.16%	8.14%	10.63%	13.53%	#DIV/0!	#DIV/0!	5.49%
FeOH WL	nm	I erraspec	2,252	2,252	#DIV/0!	2,254	2,252	2,250	2,251	#DIV/0!
AlOH denth/FeOH denth	70	Terraspec	1.84%	2.94%	#DIV/0! #DIV/0!	2.00%	2.41% 5.61	1.92% #DIV/01	3.4/% #DIV/01	#DIV/0! #DIV/0!
ruon acpuir con acptil		renaspee	/.1/	0.75	$\pi D (V/0!$	5.50	5.01	πD1 ¥/U!	π131 ¥/0!	πD1 ¥/U:
Abbreviations:	viations: Units HFL= Hanging wall flow HBX= Hanging wall breccia			FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow		Alteration Qtz-Ser= Quartz-sericite Chl-Ser= Chlorite-sericite		<b>Other</b> WL= Wavelength		

HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow Chl=Chlorite INT= Intrusion

Appendix	idix B: Table B1.1 Whole-rock Geochemistry and Terras							ec <sup>TM</sup> Data		
Sample ID			H501229	H501230	H501231	H501232	H501233	H501234	H501235	H501236
Hole ID			BD10-56	BD10-49	BD10-49	BD10-49	BD10-49	BD10-34	BD10-34	BD10-34
Depth (m)			49.2	10.9	11.8	25	34.6	11.5	26	33.6
Lithology			FLT	FLT	FFL	FLT	FLT	HFL	HBX	FLT
Alteration	0/	FUG LOD	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
	%0 0/	FUS-ICP	00.75	/3.42	//.01	12.00	05.45	12.26	10.52	/0.8
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	9.04	10.52	11.28	12.09	11.9	13.36	12.74	10.39
$Fe_2O_3(1)$	%	FUS-ICP	12.87	6.1	3.57	5.8	8.94	/.18	2.08	3.06
MnO MgO	%0 0/.	FUS-ICP	0.01/	0.032	0.036	0.145	0.081	0.11/	0.069	0.089
CaO	/0 %	FUS-ICP	0.06	0.06	0.06	0.13	0.12	0.08	0.08	0.39
Na <sub>2</sub> O	%	FUS-ICP	0.2	0.22	0.21	0.2	0.23	0.00	0.00	0.18
K <sub>2</sub> O	%	FUS-ICP	2 46	2.84	2.87	2.64	2.86	3 55	3 48	2.89
TiO	%	FUS-ICP	0.14	0.159	0.148	0 327	0 174	0.168	0.156	0.158
P <sub>2</sub> O <sub>2</sub>	%	FUS-ICP	0.01	0.01	< 0.01	0.06	0.09	0.02	0.02	0.02
LOI	%	FUS-ICP	7.97	4.83	3.42	5.47	6.98	6.17	3.44	4.04
Total	%	FUS-ICP	100.1	98.97	100.6	100.2	98.97	98.6	99.34	99.62
Ba	ppm	FUS-ICP	970	1292	1242	943	1280	1249	1211	782
Sr	ppm	FUS-ICP	7	10	10	11	11	7	12	13
Y	ppm	FUS-ICP	33	40	36	36	40	48	46	37
Sc Zr	ppm	FUS-ICP	9	11	141	17	12	11	10	11
ZI Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	107	136	155
V	ppm	FUS-ICP	< 5	6	< 5	71	6	< 5	7	7
Hg	ppb	FIMS	29	17	< 5	11	22	870	22	116
Cr	ppm	HF/HNO3	<8.9	12.94	<8.9	25.90	12.07	9.73	11.13	<8.9
Co	ppm	HF/HNO3	2.23	7.28	3.58	9.39	4.63	<0.9	<0.9	1.46
Ni	ppm	HF/HNO3	<10	<10	<10	20.41	<10	<10	<10	<10
Cu	ppm	HF/HNO3	2450.53	4906.70	249.57	184.23	389.14	472.52	<19	1503.49
Zn	ppm	HF/HNO3	84.62	5 20	/8.53	94.79	27.71	27.01	165.98	× 10
Se	nnm	HF/HNO3	<26.7	<267	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	133.78	162.94	163.63	158.49	159.93	262.27	274.49	151.87
Mo	ppm	HF/HNO3	5.61	8.77	6.41	6.07	8.35	3.89	8.97	1.18
Ag	ppm	HF/HNO3	<1	1.41	<1	<1	<1	<1	<1	1.48
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	36.60	1.16	3.86
Sn	ppm	HF/HNO3	4.93	6.51	6.06	7.12	8.93	6.04	4.67	6.04
Sb T-	ppm	HF/HNO3	1.53	0.77	0.45	3.10	6.31	3.26	19.38	2.03
IC	ppm	HF/HNO3	<4.5	4.31 <37	<4.5	<37	0.73 <37	<4.5	<4.5	<4.3
W	ppm	HF/HNO3	3.67	4.23	3.48	6.33	3.45	2.69	2.42	6.22
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	61.37	69.83	<25.3	340.65
Bi	ppm	HF/HNO3	4.86	10.50	3.65	3.69	8.37	1.01	0.20	1.58
Nb	ppm	Na2O2 Sinter	4.99	5.04	4.86	5.00	5.24	6.16	7.94	7.02
La	ppm	Na2O2 Sinter	14.09	13.37	13.10	17.15	14.29	18.86	19.83	18.69
Ce	ppm	Na2O2 Sinter	29.50	28.68	27.20	36.03	29.32	39.77	41.37	40.04
Pr	ppm	Na2O2 Sinter	3.71	3.61	3.49	4.61	3.72	5.12	5.17	5.09
Nd	ppm	Na2O2 Sinter	16.08	15.29	14.49	19.08	15.82	21.38	21.93	22.48
Sm	ppm	Na2O2 Sinter	4.49	4.19	3.64	4.77	4.04	5.34	5.54	5.42
Eu	ppm	Na2O2 Sinter	1.06	0.77	0.44	0.45	1.35	0.87	1.17	1.08
Tb	ppm	Na2O2 Sinter	0.88	0.78	0.73	0.82	0.83	1.01	1.15	1.05
Dy	ppm	Na2O2 Sinter	5.75	5.23	4.86	5.37	5.58	6.82	8.07	7.24
Ho	ppm	Na2O2 Sinter	1.09	0.99	1.04	1.14	1.19	1.47	1.78	1.63
Er	ppm	Na2O2 Sinter	3.05	3.17	3.68	3.72	4.51	5.47	5.15	3.91
Tm	ppm	Na2O2 Sinter	0.52	0.45	0.49	0.56	0.56	0.69	0.82	0.81
Yb	ppm	Na2O2 Sinter	3.53	3.19	3.40	3.86	3.84	4.76	5.68	5.30
Lu	ppm	Na2O2 Sinter	0.61	0.55	0.52	0.62	0.59	0.71	0.82	0.80
Hf	ppm	Na2O2 Sinter	3.80	3.74	2.68	3.16	2.73	4.23	4.19	4.47
Та	ppm	Na2O2 Sinter	0.21	0.20	0.19	0.21	0.19	0.23	0.30	0.29
Th	ppm	Na2O2 Sinter	3.88	3.84	3.71	4.44	3.80	5.16	6.13	5.77
AIOH WL	nm	Terraspec	2,202	2,203	2,203	2,202	2,201	2,203	2,206	2,206
AlOH Depth	%	Terraspec	6.51%	12.88%	10.54%	11.15%	6.41%	9.57%	5.61%	4.96%
FeOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	2,253	2,252	2,255	2,255	#DIV/0!	#DIV/0!
FeOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	1.03%	1.43%	1.38%	1.18%	#DIV/0!	#DIV/0!
AIOH deptn/FeOH depth		1 erraspec	#DIV/0!	#DIV/0!	10.24	7.82	4.64	8.11	#DIV/0!	#DIV/0!
Abbreviations:	Units HFL=	= Hanging wall flow	,	FT= Footwall FBX= Footwa	tuff Ill breccia	Alteration Qtz-Ser= Qua	rtz-sericite		Other WL= Waveler	ngth

HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow Chl-Ser= Chlorite-sericite Chl=Chlorite INT= Intrusion

Аррениіх	x B: Table B1.1 whole-rock Geochemist					iennsu y	anu .	rerrasp	Let L	<i>v</i> ata
Sample ID			H501237	H501238	H501239	H501240	H501241	H501242	H501243	H501244
Hole ID			BD10-34	BD10-33	BD10-33	BD10-33	BD10-33	BD10-33	BD10-45	BD10-45
Depth (m)			18.6	9.5	18.3	27.1	36.3	46.2	8.1	18.4
Lithology			HFL	HFL	FLT	HBX	FLT	FFL	HFL	HFL
Alteration			Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	72.35	60.66	74.96	72.76	64.89	76.19	58.85	76
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	11.04	15.82	11.62	13.19	12.98	9.72	18.41	10.78
$Fe_2O_3(T)$	%	FUS-ICP	4.19	4.76	3.3	3.67	7.02	4.78	6.53	5.21
MnO	%	FUS-ICP	0.177	0.329	0.084	0.136	0.106	0.089	0.201	0.067
MgO	%	FUS-ICP	3.14	5.42	1.89	2.52	1.89	2.47	4.04	1.45
CaO	%	FUS-ICP	0.07	0.17	0.07	0.17	0.47	0.06	0.1	0.07
Na <sub>2</sub> O	%	FUS-ICP	0.19	0.18	0.16	0.18	0.23	0.16	0.37	0.24
K <sub>2</sub> O	%	FUS-ICP	2.71	3.98	3	3.57	3.31	2.33	4.35	2.53
TiO <sub>2</sub>	%	FUS-ICP	0.135	0.194	0.19	0.164	0.309	0.133	0.217	0.131
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	0.03	0.02	0.03	0.06	0.02	0.03	0.03
LOI	%	FUS-ICP	5.27	6.84	3.71	4.54	6.55	4.32	7.24	4.4
Total	%	FUS-ICP	99.27	98.37	99.02	100.9	97.81	100.3	100.3	100.9
Ba	ppm	FUS-ICP	811	1135	1385	1409	2988	721	2472	3586
Sr	ppm	FUS-ICP	9	8	8	10	18	7	20	13
Y	ppm	FUS-ICP	38	59	37	80	41	35	52	26
Sc	ppm	FUS-ICP	9	14	12	13	16	10	15	9
Zr	ppm	FUS-ICP	135	189	134	178	129	119	216	128
Be	ppm	FUS-ICP	1	1	1	1	1	< 1	1	1
V	ppm	FUS-ICP	< 5	< 5	16	< 5	61	< 5	< 5	< 5
Hg	ppb	FIMS	143	10	115	19	22	56	8	5
Cr C-	ppm	HF/HNO3	9.63	<8.9	<8.9	<8.9	27.63	<8.9	<8.9	< 8.9
C0 Ni	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	0.25	1.00	<0.9	1.43
NI Cu	ppm	HF/HNO3	113.06	31.00	180.75	20.47	1346.30	359.61	<10	<10
Zu Zn	ppm	HE/HNO3	1475 50	134.13	2193.91	285.67	121.86	496.65	165.88	86.99
As	ppm	HF/HNO3	3.16	2.87	6.29	3.31	21.22	11.44	1.94	3.29
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	256.49	149.18	143.95	120.25	128.58	142.30	139.08	134.06
Mo	ppm	HF/HNO3	1.90	1.73	<1	5.06	3.10	1.88	3.27	2.03
Ag	ppm	HF/HNO3	<1	<1	2.70	<1	1.91	<1	<1	<1
Cd	ppm	HF/HNO3	4.51	<0.9	6.23	1.16	0.97	1.74	<0.9	<0.9
Sn	ppm	HF/HNO3	4.68	4.73	10.08	4.63	5.79	4.07	5.08	5.12
Sb	ppm	HF/HNO3	1.46	1.47	7.85	1.31	2.07	3.44	1.23	0.98
Te	ppm	HF/HNO3	<4.3	<4.3	4.38	<4.3	<4.3	<4.3	<4.3	<4.3
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.12	2.11	4.19	3.00	/.60	1.82	1.13	1.80
PD D;	ppm	HF/HNO3	<25.5	<25.5	2402.52	<25.5	02.93	155.94	<25.5	<25.5
DI NI	ppm	Na2O2 Sintar	5.40	0.20	0.42	0.55	0.00	1.11	0.29	1.01
ND	ppm	Na2O2 Sinter	5.40	6.08	9.50	/.45	8.08	6.92	0.18	10.26
La	ppm	Na2O2 Sinter	16.59	17.13	21.11	18.30	21.03	16.77	18.39	23.19
Ce	ppm	Na2O2 Sinter	35.30	35.05	44.44	37.94	44.20	35.94	37.89	48.85
Pr	ppm	Na2O2 Sinter	4.50	4.44	5.56	4.82	5.62	4.56	4.78	6.19
Nd	ppm	Na2O2 Sinter	18.89	19.53	23.25	20.93	24.42	19.63	20.35	26.73
Sm	ppm	Na2O2 Sinter	4.84	4.76	7.10	4.91	6.15	4.77	4.84	6.36
Eu	ppm	Na2O2 Sinter	0.98	0.93	1.67	1.16	1.17	0.90	0.91	0.91
Tb	ppm	Na2O2 Sinter	0.91	0.97	1.53	0.96	1.42	0.98	0.88	1.29
Dy	ppm	Na2O2 Sinter	5.99	6.43	10.23	6.39	9.87	6.78	5.85	8.77
Но	ppm	Na2O2 Sinter	1.28	1.44	1.98	1.42	2.39	1.57	1.30	1.93
Er	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	4 30	6.08	4 40	7.92	4 62	3 99	6.13	3 64
Tm	nnm	Na2O2 Sinter	0.59	0.66	0.94	0.68	1.22	0.71	0.63	0.95
Vh	ppin	Na2O2 Sinter	4.01	4.52	6.49	4.45	9.42	4.74	4.05	6.59
IU Lu	ppm	Na2O2 Sinter	4.01	4.55	1.10	4.45	0.45	4.74	4.05	0.58
Lu	ppm	Na2O2 Sinter	0.62	0.66	1.10	0.68	1.51	0.71	0.63	0.97
Ht	ppm	Na2O2 Sinter	3.59	3.98	5.97	4.88	5.46	4.06	3.18	5.73
Та	ppm	Na2O2 Sinter	0.22	0.26	0.42	0.30	0.31	0.25	0.24	0.43
Th	ppm	Na2O2 Sinter	4.61	5.05	6.68	5.25	6.19	5.12	4.77	8.03
AlOH WL	nm	Terraspec	2,201	2,203	2,205	2,206	2,202	2,205	2,203	2,200
AIOH Depth	%	Terraspec	4.62%	12.05%	6.92%	8.94%	12.38%	3.57%	7.89%	9.93%
FeOH WL	nm	Terraspec	2,254	2,251	#DIV/0!	#DIV/0!	2,257	2,252	#DIV/0!	#DIV/0!
FeOH Depth	%	1 erraspec	1.32%	2.94%	#DIV/0!	#DIV/0!	1.27%	1.08%	#DIV/0!	#DIV/0!
AIOH deptn/reOH depth		1 erraspec	3.50	4.11	#DIV/0!	#DIV/0!	9./4	3.51	#DIV/0!	#DIV/0!
Abbreviations:	Unite			FT= Footwall	tuff	Alteration			Other	
	HFL= Hanging wall flow		w	FI=Footwall tuff FBX=Footwall breccia		Qtz-Ser= Quartz-sericite		e WL= Wavelength		
	HBX	= Hanging wall bre	ccia	FFL= Footwal	ll flow	Chl-Ser= Chlo	rite-sericite			-
	FLT= Footwall lapilli tuff		ff	INT= Intrusio	n	Chl=Chlorite				

прреник	IX D. TADIC DI.1 WHOIL-TOCK OCOCI					inclinistry and retraspec Data				
Sample ID			H501245	H501246	H501247	H501248	H501249	H501250	H501251	H501252
Hole ID			BD10-45	BD10-45	BD10-28	BD10-28	BD10-28	BD11-127	BD11-127	BD11-127
Depth (m)			31.4	46.7	6	20.4	34.3	67	16.4	21.1
Lithology			51.4 EI T	FDV	FIT	20.4 FI T	FIT			21.1 EI T
Littlology			I'LI	TDA OL G	I'LI	I'LI	CLLG			I'LI
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	69.73	70.61	74.05	71.3	69.5	75.85	71.33	74.86
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	11.61	12.64	9.76	13.39	10.11	10.8	12.43	10.99
Fe O (T)	0/2	EUS ICD	6 30	4.28	4.61	2.00	5 72	2.74	5.24	4.56
10203(1)	/0	FUS-ICF	0.57	4.20	4.01	2.77	0.000	2.74	0.110	4.50
MnO	%	FUS-ICP	0.07	0.11	0.189	0.135	0.288	0.128	0.118	0.112
MgO	%	FUS-ICP	1.38	2.67	3.85	1.39	3.99	2.48	2.38	1.92
CaO	%	FUS-ICP	0.1	0.06	0.06	1.37	0.96	0.18	0.07	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.26	0.22	0.07	0.14	0.1	0.26	0.32	0.28
K.O	0/2	FUS-ICP	3.15	3.02	23	4.15	2 30	2.0	3 25	2.96
R20	70	105-101	5.15	5.02	2.5	4.15	2.57	2.)	5.25	2.90
T1O <sub>2</sub>	%	FUS-ICP	0.172	0.173	0.135	0.189	0.228	0.132	0.153	0.165
$P_2O_5$	%	FUS-ICP	0.03	0.03	< 0.01	0.01	0.03	0.01	0.01	0.01
LOI	%	FUS-ICP	5.77	4.3	5.09	5.07	6.44	4.58	5.65	4.99
Total	%	FUS-ICP	98.66	98.12	100.1	100.1	99 76	100.1	101	100.9
Ba	nnm	FUS-ICP	1140	970	318	602	413	863	919	861
S.	ppm	FUS ICD	140	0	510	15	415	11	14	12
Sf	ppm	FUS-ICP	14	8	4	15	13	11	14	12
Ŷ	ppm	FUS-ICP	37	45	37	48	28	36	45	33
Sc	ppm	FUS-ICP	12	13	9	14	14	9	11	11
Zr	ppm	FUS-ICP	138	154	121	170	108	130	155	137
Be	ppm	FUS-ICP	1	1	1	1	< 1	1	1	1
V	ppm	FUS-ICP	7	< 5	< 5	< 5	56	< 5	< 5	11
Нσ	nnh	FIMS	17	< 5	14	11	37	89	83	16
Cr	ppo	HE/HNO3	< 8.0	< 8.0	< 8.9	<8.0	18 30	< 8 0	12.41	10.14
Ca	ppm	UE/UNO2	2.61	<0.0	<0.0	<0.0	7.94	<0.0	1.20	2.95
20	ppm		2.01	<0.9	<0.9	<0.9	/.04	<0.9	1.20	5.65
NI	ppm	HF/HNO3	<10	<10	<10	<10	13.23	<10	<10	<10
Cu	ppm	HF/HNO3	1440.83	348.74	<19	<19	<19	97.50	435.41	117.09
Zn	ppm	HF/HNO3	65.80	68.54	131.26	96.97	203.60	352.67	708.98	119.76
As	ppm	HF/HNO3	7.39	4.89	4.33	9.59	9.68	4.81	8.04	10.39
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	141 44	154 99	256 10	261 95	239.05	228 56	267 04	245 28
Mo	ppm	HE/HNO3	6.72	4.12	12 79	4 90	6 55	3 80	7 92	2 42
100	ppm	III/IINO3	<1	4.12	-1	4.50	-1	5.00	<1	2.42
Ag	ppm	HF/HNO3	<0.0	<0.0	<0.0	<0.0	<0.0	1 28	1 47	<0.0
Ca	ppm	HF/HNO5	<0.9	<0.9	<0.9	<0.9	<0.9	1.28	1.4/	<0.9
Sn	ppm	HF/HNO3	6.32	3.63	3.49	4.25	3.41	2.92	4.79	4.02
Sb	ppm	HF/HNO3	1.38	1.61	1.52	1.42	1.37	50.51	2.35	2.69
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.64	3.90	1.15	2.26	1.91	1.77	2.76	3.01
Ph	nnm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	31.93	<25.3
Bi	ppm	HE/HNO3	4.01	0.44	1.64	0.90	1.63	0.61	0.62	2 10
DI	ppm		4.01	0.44	1.04	0.70	1.05	0.01	0.02	2.17
Nb	ppm	Na2O2 Sinter	6.43	8.16	7.41	5.62	7.90	4.42	5.96	6.75
La	ppm	Na2O2 Sinter	18.19	22.35	19.65	17.13	22.17	14.37	16.67	19.18
Ce	ppm	Na2O2 Sinter	37 29	46.61	41.06	35.02	47 30	30.31	35 34	39 40
D-	ppm	No202 Sintar	4.72	5.00	4.00	4.42	5.04	2.07	1.50	5.00
Pf	ppm	Na2O2 Sinter	4.72	5.80	4.98	4.45	5.94	3.87	4.56	5.00
Nd	ppm	Na2O2 Sinter	20.24	25.14	21.33	18.69	25.29	16.60	19.40	21.41
Sm	ppm	Na2O2 Sinter	4.71	5.74	4.67	4.45	6.02	3.90	4.63	5.03
En		Na2O2 Sinter	0.20	0.75	1.02	1.26	1.55	1 19	0.40	0.06
Eu	ppm	N 202 5. 4	0.57	0.75	1.02	1.50	1.55	1.10	0.49	0.90
Тв	ppm	Na2O2 Sinter	0.80	1.02	1.13	0.96	1.16	0.71	0.94	1.15
Dy	ppm	Na2O2 Sinter	5.36	6.65	7.64	6.34	8.09	4.93	6.25	7.87
Но	nnm	Na2O2 Sinter	1.15	1 49	1.56	1 40	1 70	1.03	1 23	1.61
E	ppm	No O. Sintor	4.52	4.71	1.50	5.40	2.20	2.77	1.20	2.02
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.53	4./1	4.14	5.43	3.28	3.77	4.99	3.83
Tm	ppm	Na2O2 Sinter	0.54	0.72	0.73	0.62	0.83	0.52	0.60	0.73
Yb	ppm	Na2O2 Sinter	3.79	4.62	5.05	4.10	5.82	3.43	4.12	5.04
Lu.	P.P	No2O2 Sinter	0.56	0.71	0.60	0.60	0.87	0.54	0.61	0.80
LU	ppm	Na2O2 Sinter	0.56	0./1	0.69	0.60	0.8/	0.54	0.61	0.80
Hf	ppm	Na2O2 Sinter	4.17	4.06	4.00	2.33	4.34	2.95	3.53	4.13
Та	ppm	Na2O2 Sinter	0.26	0.30	0.26	0.22	0.29	0.18	0.24	0.26
ть	nom	Na2O2 Sinter	5.01	5.06	5 77	1 12	6.19	3.02	5.05	5 50
	ppm	Taman T	3.01	3.90	3.72	4.42	0.16	3.93	3.05	3.30
AIOH WL	nm	1 erraspec	2,203	2,205	2,208	2,208	2,205	2,205	2,205	2,204
AIOH Depth	%	Terraspec	11.48%	13.10%	5.75%	10.99%	9.49%	13.86%	10.20%	7.34%
FeOH WL	nm	Terraspec	#DIV/0!	2,253	2,251	#DIV/0!	2,254	#DIV/0!	#DIV/0!	#DIV/0!
FeOH Depth	%	Terraspec	#DIV/0!	1.53%	1.58%	#DIV/0!	1.45%	#DIV/0!	#DIV/0!	#DIV/0!
AlOH depth/FeOH depth		Terraspec	#DIV/0!	8.54	3.64	#DIV/0!	6.56	#DIV/0!	#DIV/0!	#DIV/0!
		•								
Abbreviations:	Units			FT= Footwall	huff	Alteration			Other	
	HFI =	Hanging wall flow	v	FBX= Footwa	ll breccia	Otz-Ser= Ouar	tz-sericite		WI = Waveler	ngth
	HDV.	= Hanging wall bear	Noia	FEI = Footwal	flow	Chl-Ser Chlor	ita_coriaita			
	TIDA.	manging wan blee	<i>i</i> a	TTT- FOOLWAI	110 W	Cin-Sei- Cilloi	ne-serienc			

Chl=Chlorite

FLT= Footwall lapilli tuff

Annondiv D. Table D1 1	Whole week Coochemistry or	d Tannaan aa <sup>TM</sup> Data
Appendix B: Table B1.1	Whole-rock Geochemistry an	d Terraspec <sup>11</sup> Data

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пррепата	D. TADIC DI.I WHOIC-TOCK O			GUUL	ienniser,	y and i	Cilaspec Data			
Sample ID			H501253	H501254	H501255	H501256	H501257	H501258	H501259	H501260
Hole ID			BD11-127	BD11-127	BD11-128	BD11-128	BD11-128	BD11-116	BD11-116	BD11-116
Depth (m)			37.9	46.5	12.6	15.5	26.5	13.6	16.8	29.2
Lithology			FLT	FFL	HFL	FLT	FLT	FLT	FT	FLT
Alteration			Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	70.47	75.83	73.78	73.34	78.17	68.54	75.19	68.97
Al-O	0/0	FUS-ICP	11.69	11.3	12.23	10.69	10.27	14.63	10.67	9.6
$F_2 O(T)$	0/	FUS ICD	4.60	2.17	2.25	2.00	4.01	5.25	10.07	10.67
$Fe_2O_3(1)$	70	FUS-ICP	4.09	5.17	2.77	5.00	4.01	3.23	4.30	10.67
MnO	%	FUS-ICP	0.184	0.109	0.078	0.166	0.045	0.082	0.071	0.052
MgO	%	FUS-ICP	3.87	1.86	1.37	3.05	1.01	2.1	2.54	2.34
CaO	%	FUS-ICP	0.08	0.11	0.32	0.49	0.07	0.05	0.06	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.38	0.36	0.22	0.2	0.19	0.26	0.17	0.15
K <sub>2</sub> O	%	FUS-ICP	2.72	3.1	3.5	2.83	2.92	3.76	2.38	2.09
TiO <sub>2</sub>	%	FUS-ICP	0.178	0.157	0.147	0.181	0.147	0.228	0.127	0.131
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.02	0.02	0.05	0.04	0.03	0.01	0.02	0.01
LOI	%	FUS-ICP	6.58	4.6	3.98	5.6	3.76	5.31	3.71	6.57
Total	%	FUS-ICP	100.9	100.6	98.45	100.5	100.6	100.2	99.49	100.6
Ba	ppm	FUS-ICP	701	924	673	543	596	2579	1797	1239
Sr	ppm	FUS-ICP	17	15	13	17	9	12	8	6
Y	ppm	FUS-ICP	44	40	42	37	33	42	36	24
Sc	ppm	FUS-ICP	13	11	10	13	10	15	9	8
Zr	ppm	FUS-ICP	143	148	158	137	134	177	127	124
Be	ppm	FUS-ICP	1	1	1	1	1	1	< 1	< 1
V	ppm	FUS-ICP	8	< 5	< 5	18	< 5	15	< 5	< 5
Hg	ppb	FIMS	17	11	< 5	10	208	113	25	16
Cr	ppm	HF/HNO3	<8.9	9.46	<8.9	9.69	26.17	8.99	15.72	<8.9
Co	ppm	HF/HNO3	1.69	<0.9	1.09	2.42	1.78	1.96	<0.9	1.81
Ni	ppm	HF/HNO3	<10	<10	<10	<10	17.59	<10	<10	<10
Cu	ppm	HF/HNO3	<19	447.77	73.79	<19	21.63	66.42	109.15	182.71
Zn	ppm	HF/HNO3	163.13	45.31	50.04	89.79	1322.82	110.38	126.51	61.06
As	ppm	HF/HNO3	4.48	4.11	8.72	6.40	9.75	27.05	5.55	14.79
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	239.92	271.78	126.95	112.29	179.92	235.32	264.42	263.65
Мо	ppm	HF/HNO3	4.08	4.82	2.35	2.98	10.90	10.11	4.33	21.81
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Cď	ppm	HF/HNO3	< 0.9	< 0.9	< 0.9	<0.9	4.58	< 0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	4.15	3.78	3.50	3.74	3.16	5.82	5.45	4.42
Sb	ppm	HF/HNO3	1.42	3.80	1.56	1.42	1.50	13.27	2.08	1.77
Те	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.61	8.05	2.72	2.10	2.13	4.86	1.62	2.05
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	42.97	<25.3
Bi	ppm	HF/HNO3	0.59	0.26	0.55	0.54	1.32	2.91	0.41	2.46
Nh	nnm	Na2O2 Sinter	5 58	6.46	11.85	7 24	6 40	6.00	13 39	10.23
Lo	ppin	Na2O2 Sinter	17.49	18 15	15.60	20.05	15.22	15.37	21.99	16.45
La	ppm	Na2O2 Sinter	17.40	18.15	15.00	20.93	15.55	15.57	21.88	10.45
Ce	ppm	Na2O2 Sinter	37.24	37.40	34.15	43.39	32.08	32.14	40.48	54.87
Pr	ppm	Na2O2 Sinter	4.73	4.85	4.40	5.34	4.04	4.02	5.89	4.45
Nd	ppm	Na2O2 Sinter	19.68	20.68	17.72	22.42	17.52	17.25	24.45	18.74
Sm	ppm	Na2O2 Sinter	4.79	5.24	4.70	4.43	3.98	3.79	5.77	4.48
Eu	ppm	Na2O2 Sinter	0.81	0.95	0.92	0.76	0.85	0.87	1.19	0.97
Th	nnm	Na2O2 Sinter	0.94	1 17	0.89	1.01	0.91	0.79	1.01	0.85
Du	ppin	Na2O2 Sinter	6.16	7 70	6.06	6.00	6.01	5.79	6.06	5.70
Dy H-	ppm	Na2O2 Sinter	1.21	1.70	1.20	1.20	1.24	1.00	0.90	1.20
но	ppm	Na2O2 Sinter	1.21	1.55	1.30	1.39	1.24	1.08	1.40	1.20
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.56	4.35	4.45	3.87	3.42	4.57	3.73	2.87
Tm	ppm	Na2O2 Sinter	0.57	0.73	0.65	0.67	0.59	0.55	0.73	0.60
Yb	ppm	Na2O2 Sinter	3.87	4.90	4.33	4.55	4.01	3.67	4.81	4.06
Lu	ppm	Na2O2 Sinter	0.59	0.73	0.66	0.66	0.56	0.51	0.73	0.63
Hf	ppm	Na2O2 Sinter	3.49	3.94	4.14	4.01	3,13	3,53	5,88	4,65
T.a	nnm	Na2O2 Sinter	0.22	0.25	0.34	0.27	0.23	0.22	0.41	0.30
ть	Phil	Na2O2 Sintar	4 00	5.23	7.55	5.21	4.50	4.21	0.02	6.30
	ppm	Tarrage -	4.89	5.04	1.55	5.50	4.50	4.31	9.02	0./3
AIOH WL	nm	1 erraspec	2,200	2,205	2,205	2,203	2,204	2,204	2,202	2,202
AIOH Depth	%	1 erraspec	11.90%	8.57%	9.53%	10.26%	8.14%	4.57%	/.16%	8.50%
FeOH WL	nm 9/	Terraspec	2,252	#DIV/0!	#DIV/0!	2,250	#DIV/0!	#DIV/0!	2,252	2,250
Alou double for a local	70	1 erraspec	1.40%	#DIV/0!	#DIV/0!	1.10%	#DIV/0!	#DIV/0!	1.92%	1.4/%
AIOH deptn/reOH depth		1 erraspec	8.52	#DIV/0!	#DIV/0!	8.85	#DIV/0!	#D1V/0!	3.12	5.80
Abbreviations:	Unite			FT= Footwall	tuff	Alteration			Other	
1 tool eviations.	bbreviations: Units			FT= Footwall tuff		Otz-Ser= Oug	rtz-sericite	e WL=Wavelength		
	HFL= Hanging wall flow HBX= Hanging wall bree			FBA= Footwall b FEI = Footwall fk		all flow Chl-Ser= Chlorite agricite			ite	
	FI T-	Footwall logilli to	ff	INT= Intrucio	n	Chl=Chlorita	seriene			
	1.1.1=	rootwan iapini tu		$11 \times 1 = 100 \text{ usio}$	u	Cin-Cinorite				

Annondiy D. Table D1 1	Whole week Coochemisters	and Tarwagnas <sup>TM</sup> Data
Appendix B: Table B1.1	Whole-rock Geochemistry	and Terraspec Data

Appendix	Appendix B: Table B1.1 whole-rock Geochemistry						y anu i	errasp	ec I	Jata
Sample ID			H501261	H501262	H501263	H501264	H501265	H501266	H501267	H501268
Hole ID			BD11-116	BD11-116	BD11-120	BD11-120	BD11-120	BD10-88	BD10-7	BD10-7
Depth (m)			38.1	49.7	9.4	34.7	42.2	33.5	11.2	17.7
Lithology			FT	FLT	HFL	HBX	HFL	FLT	FLT	FLT
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl	Qtz-Ser	Chl
SiO <sub>2</sub>	%	FUS-ICP	63.36	50.52	83.04	72.57	65.98	40.75	69.7	34.12
$Al_2O_3$	%	FUS-ICP	15.4	8.22	8.06	10.02	12.94	14.95	12.94	8.59
$Fe_2O_3(T)$	%	FUS-ICP	7.04	23.38	3.17	5.38	6.4	17.95	6.03	28.8
MnO	%	FUS-ICP	0.058	0.036	0.051	0.11	0.169	0.178	0.01	0.307
MgO	%	FUS-ICP	2.19	0.41	0.92	1.92	3.42	11.34	1.03	10.02
CaO	%	FUS-ICP	0.09	0.03	0.1	0.21	0.76	0.1	0.03	0.2
Na <sub>2</sub> O	%	FUS-ICP	0.26	0.16	0.14	0.18	0.24	0.3	0.16	0.01
K <sub>2</sub> O	%	FUS-ICP	4.04	2.26	2.15	2.58	3.13	1.22	3.64	0.04
TiO <sub>2</sub>	%	FUS-ICP	0.316	0.114	0.1	0.165	0.216	0.254	0.176	0.111
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.04	< 0.01	< 0.01	0.01	0.03	0.03	0.02	0.1
LOI	%	FUS-ICP	6.08	13.6	2.7	5.08	6.95	13.51	4.68	15.96
Total	%	FUS-ICP	98.89	98.74	100.5	98.24	100.2	100.6	98.42	98.25
Ba	ppm	FUS-ICP	2181	1167	813	1094	1208	332	1749	14
Sr	ppm	FUS-ICP	11	6	7	10	19	31	8	2
Y	ppm	FUS-ICP	48	25	30	38	42	42	51	25
Sc	ppm	FUS-ICP	18	9	7	11	13	13	13	9
Zr	ppm	FUS-ICP	162	106	98	132	166	184	170	92
Be	ppm	FUS-ICP	1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	57	< 5	< 5	12	7	14	< 5	< 5
Hg	ppb	FIMS	92	785	31	18	22	9	785	36
Cr C-	ppm	HF/HNO3	47.76	< 8.9	13.41	< 8.9	12.84	14.35	35.14	32.01
C0 Ni	ppm	HF/HNO3	20.88	12.59	<0.9	1./3	4.98	10.38	1.55	343.80
Cu	ppm	HE/HNO3	037.63	127 77	×10 830.61	720 77	438.07	<10	154.46	19.39
Zn	nnm	HE/HNO3	96.85	30.40	99.01	78.62	122.61	194.48	6300.16	4096 77
As	ppm	HF/HNO3	50.33	159.18	26.50	102.87	8 90	11.86	18 66	81.83
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	30.13	55.28
Br	ppm	HF/HNO3	255.24	242.90	283.43	152.74	266.53	228.98	236.81	198.56
Mo	ppm	HF/HNO3	6.99	32.86	7.71	36.71	8.96	18.63	9.35	6.95
Ag	ppm	HF/HNO3	1.80	6.20	<1	<1	<1	<1	<1	2.47
Cd	ppm	HF/HNO3	0.96	< 0.9	<0.9	<0.9	<0.9	<0.9	27.28	<0.9
Sn	ppm	HF/HNO3	5.02	4.71	4.17	4.93	4.48	3.49	15.80	7.32
Sb	ppm	HF/HNO3	11.79	196.01	11.19	4.08	2.32	0.86	5.02	1.49
Te	ppm	HF/HNO3	<4.3	19.07	<4.3	<4.3	<4.3	<4.3	<4.3	16.21
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	8.19	2.96	3.61	4.64	5.73	13.61	5.05	7.10
PD D:	ppm	HF/HNO3	33.29	44.//	<25.5	<25.5	<23.3	<25.5	57.98	<25.3
Ы	ppm	No2O2 Sinter	0.97	42.21	1.95	9.64	5.27	4.01	19.90	10.00
ND	ppm	Na2O2 Sinter	/.86	10.71	6.90	7.03	5.71	9.93	6.55	4.45
La	ppm	Na2O2 Sinter	17.10	21.98	13.22	12.09	16.88	21.41	20.79	15.73
Ce	ppm	Na2O2 Sinter	36.02	46.92	27.65	26.05	35.32	43.69	43.95	33.82
Pr	ppm	Na2O2 Sinter	4.54	5.93	3.43	3.23	4.51	5.38	5.52	4.39
Nd	ppm	Na2O2 Sinter	18.31	25.17	13.99	13.78	19.50	22.22	24.34	19.00
Sm	ppm	Na2O2 Sinter	4.36	6.41	3.43	3.46	4.64	5.41	5.28	4.54
Eu	ppm	Na2O2 Sinter	0.96	1.52	0.90	0.75	1.07	0.97	1.03	0.71
Tb	ppm	Na2O2 Sinter	0.71	1.28	0.70	0.69	0.89	1.05	1.25	0.91
Dy	ppm	Na2O2 Sinter	4.50	8.59	4.73	4.77	6.01	7.24	8.00	5.67
Но	ppm	Na2O2 Sinter	0.94	1.77	0.97	1.01	1.27	1.52	1.67	1.06
Er	nnm	Na <sub>2</sub> O <sub>2</sub> Sinter	5 58	3.01	3.11	3.89	4 92	4 88	5 19	3.16
Tm	nnm	Na2O2 Sinter	0.42	0.87	0.46	0.48	0.61	0.74	0.78	0.47
Vh	ppin	Na2O2 Sinter	2.02	5.87	2 22	2.24	4.21	5.12	5.10	3.08
IU Lu	ppm	Na2O2 Sinter	2.95	0.02	0.40	0.51	4.21	0.78	0.74	0.45
Lu	ppm	Na2O2 Sinter	0.44	0.92	0.49	0.51	0.60	0.78	0.74	0.45
Ht	ppm	Na2O2 Sinter	3.93	6.09	3.79	3.10	3.30	5.26	3.95	2.32
Ta	ppm	Na2O2 Sinter	0.22	0.37	0.19	0.24	0.21	0.33	0.23	0.16
Th	ppm	Na2O2 Sinter	6.15	7.10	3.85	4.38	4.25	7.12	5.56	3.26
AIOH WL	nm	Terraspec	2,203	2,204	2,203	2,205	2,203	2,192	2,205	#DIV/0!
AIOH Depth	%	Terraspec	5.22%	4.02%	7.54%	9.52%	3.32%	4.00%	18.50%	#DIV/0!
FeOH WL	nm	1 erraspec	2,252	#DIV/0!	2,252	2,255	#DIV/0!	2,252	#DIV/0!	2,250
AlOH donth/FaOH donth	70	Terraspec	2./0%	#DIV/0!	1.14%	1.50%	#DIV/0!	5.1/% 0.77	#DIV/0!	4.05% #DIV/01
AIOT depui/reOH depth		retraspec	1.95	#D1V/0!	0.02	0.55	#D1V/0!	0.//	#DIV/0!	#D1V/0!
Abbreviations:	Unite			FT= Footwall	tuff	Alteration			Other	
	HFL= Hanging wall flow		w	r I = Footwall tuff FBX= Footwall breccia		Qtz-Ser= Quartz-sericite		e WL= Wavelength		
	HBX= Hanging wall breccia		ccia	FFL= Footwal	l flow	Chl-Ser= Chlo	orite-sericite			-
	HBX= Hanging wall breccia FLT= Footwall lapilli tuff		ff	INT= Intrusion	n	Chl=Chlorite				
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Appendix B: Table B1.1 whole-rock Geochemistry and Terr							rerrasp	Dec L	<i>v</i> ata	
Sample ID			H501269	H501270	H501271	H501272	H501273	H501274	H501275	H501276
Hole ID			BD10-7	BD10-22	BD10-22	BD10-22	BD10-22	BD10-9	BD10-9	BD10-9
Depth (m)			27.8	4.6	13.9	17.9	23.1	4.1	13.5	20.4
Lithology			FLT	HBX	FLT	FFL	FLT	HBX	FLT	FBX
Alteration			Chl	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	22.81	63.05	71.11	72.07	72.17	64.31	78.03	72.4
$Al_2O_3$	%	FUS-ICP	17.24	13.04	12.48	13.25	11.84	12.54	11.88	12.93
$Fe_2O_3(T)$	%	FUS-ICP	22.88	5.76	4.24	3.15	4.46	5.61	2.22	3.04
MnO	%	FUS-ICP	0.161	0.195	0.11	0.251	0.075	0.215	0.019	0.145
MgO	%	FUS-ICP	20.67	8.06	1.79	2.95	1.23	7.65	1.05	1.88
CaO N- O	%	FUS-ICP	0.06	0.06	0.06	0.06	0.15	0.06	0.05	0.17
Na <sub>2</sub> O	%0 0/	FUS-ICP	< 0.01	0.08	0.14	0.17	2.11	0.07	0.1	0.12
K <sub>2</sub> O	%0 0/	FUS-ICP	< 0.01	2.04	5.49	3.55	2.49	2.23	5.49	3.99
110 <sub>2</sub>	70	FUS-ICP	0.246	0.180	0.164	0.203	0.139	0.178	0.103	0.182
P <sub>2</sub> O <sub>5</sub>	70 0/.	FUS-ICP	0.05	5.84	0.05	0.01	4.01	5.88	0.02	0.02
Total	%	FUS-ICP	99.78	98 33	97.89	100.1	98 73	98 77	99.67	98.85
Ba	ppm	FUS-ICP	8	1962	2922	2110	1312	1554	1819	1852
Sr	ppm	FUS-ICP	< 2	4	8	11	26	5	8	10
Y	ppm	FUS-ICP	66	45	39	48	38	48	39	51
Sc	ppm	FUS-ICP	21	12	13	14	12	11	12	13
Zr	ppm	FUS-ICP	235	162	162	168	141	150	150	167
Be	ppm	FUS-ICP	< 1	1	1	1	1	1	1	1
V Ha	ppm	FUS-ICP	< 5	20	1260	/	1200	6 72	25	9
пg Cr	ppo	HF/HNO3	29.66	9.98	<8.9	<8.9	<8.9	11 39	<8.9	<8.9
Co	ppm	HF/HNO3	41.05	10.59	2.07	<0.9	3.70	12.93	<0.9	1.02
Ni	ppm	HF/HNO3	19.46	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	295.60	301.01	2858.49	183.18	777.63	177.05	386.27	672.49
Zn	ppm	HF/HNO3	406.50	301.32	8466.09	628.35	6724.56	607.21	1295.26	3357.90
As	ppm	HF/HNO3	9.15	7.94	14.81	<1.6	45.93	6.81	17.83	7.63
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	212.21	241.52	235.82	254.93	500.71 6.02	157.39	157.06	158.46
Δσ	ppm	HF/HNO3	1.03	1.04	8.16	<1	<1	2 38	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	31.16	1.27	22.51	<0.9	4.93	12.83
Sn	ppm	HF/HNO3	2.04	6.56	16.28	4.96	17.44	5.27	15.52	36.40
Sb	ppm	HF/HNO3	1.43	4.20	13.11	1.94	10.46	2.64	7.05	4.21
Te	ppm	HF/HNO3	32.13	12.41	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	9.53	7.87	3.28	2.28	1.51	6.12	3.55	1.64
PD	ppm	HF/HNO3	<25.5 35.85	28.44	13.65	<25.5	2 07	402.41 5.32	28.25	<25.5
Nb	ppm	Na2O2 Sinter	11 40	12.50	11.17	12.10	2.07	7.82	6.23	7.20
Io	ppm	Na2O2 Sinter	28.22	22.10	22.22	22.45	19.49	21.00	20.56	24.24
La	ppm	Na2O2 Sinter	77.26	18 76	47.01	40.75	29.72	21.90	20.50	50.94
Dr.	ppm	Na2O2 Sinter	0.60	5.00	5.91	6.18	4 00	5 76	5.42	6.32
ri Nd	ppm	Na2O2 Sinter	9.00	25.09	24.09	25.84	4.90	24.75	22.62	0.33
Nd	ppm	Na2O2 Sinter	41.55	25.08	24.08	25.84	21.14	24.75	23.62	27.21
Sm	ppm	Na2O2 Sinter	8.80	0.03	5.91	0.18	5.54	5.70	5.55	0.30
Eu	ppm	No2O2 Sinter	0.98	1.14	2.01	1.58	1.50	0.96	1.48	1.51
1b	ppm	Na2O2 Sinter	1.//	1.20	0.99	1.17	0.93	1.20	0.99	1.38
Dy	ppm	Na2O2 Sinter	11.39	7.98	0.50	/.96	6.33	8.20	0.55	9.26
Но	ppm	Na2O2 Sinter	2.31	1.69	1.41	1.65	1.29	1.74	1.42	1.94
Er	ppm	$Na_2O_2$ Sinter	7.14	5.15	4.47	5.17	4.11	5.27	4.33	5.88
Im	ppm	Na2O2 Sinter	1.06	0.79	0.68	0.78	0.63	0.80	0.68	0.88
Yb	ppm	Na2O2 Sinter	7.36	5.47	4.66	5.38	4.33	5.25	4.68	6.04
Lu	ppm	Na2O2 Sinter	1.04	0.85	0.72	0.86	0.66	0.79	0.68	0.89
Hf	ppm	Na2O2 Sinter	5.95	5.70	5.10	5.70	4.96	4.90	4.48	5.06
Та	ppm	Na2O2 Sinter	0.34	0.37	0.34	0.35	0.29	0.27	0.25	0.28
Th	ppm	Na2O2 Sinter	8.05	7.59	6.99	7.17	5.91	5.70	5.51	6.34
AIOH WL	nm	Terraspec	#DIV/0!	2,205	2,210	2,208	2,210	2,210	2,210	2,213
AIOH Depth	% nm	1 erraspec	#DIV/0!	11.08%	13.83%	31.00%	8.27% #DIV/01	10.67%	20.38% #DIV/01	50.28% #DIV/01
FeOH Denth	1111 %	Terraspec	2,230	2,231	2,234	2,235	#DIV/0!	6.00%	#DIV/0!	#DIV/0!
AlOH depth/FeOH depth	/0	Terraspec	#DJV/0!	1.32	12.35	18 18	#DIV/0!	1.78	#DIV/0!	#DIV/0!
eon depair					- 2.00					
Abbreviations:	Units FT= Footwall tuff Alteration Other									
	HFL= Hanging wall flow		FBX= Footwa	ll breccia	Qtz-Ser= Qua	rtz-sericite	WL= Wavelength			
	HBX	= Hanging wall bree	ccia	FFL= Footwal	ll flow	Chl-Ser= Chlo	orite-sericite			
	FLT=	Footwall lapilli tuf	t	INT= Intrusio	n	Chl=Chlorite				

Appendix	dix B: Table B1.1 whole-rock Geochemistry and Terraspe					ec D	ลเล			
Sample ID			H501277	H501278*	H501279	H501280	H501281	H501282	H501283	H501284
Hole ID			BD10-9	BD10-11	BD10-11	BD10-11	BD10-11	BD10-21	BD10-21	BD10-21
Depth (m)			30.1	4	9.8	17.1	28.2	2.4	15.9	30.1
Lithology			FBX	HFL	HFL	FLT	FBX	HFL	FLT	FBX
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	75.06	73.94	70.41	80.77	72.22	66.63	43.91	79.58
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	11.98	11.79	13.66	8.65	13.92	18.73	7.93	8.23
$Fe_2O_3(T)$	%	FUS-ICP	2.55	2.34	3.31	2.45	1.69	2.82	18.49	1.69
MnO	%	FUS-ICP	0.142	0.095	0.148	0.177	0.112	0.015	0.153	0.24
MgO	%	FUS-ICP	2.49	2.46	3.02	1.23	2.76	2.05	3.87	2.24
CaO	%	FUS-ICP	0.06	0.16	0.04	0.05	0.09	0.64	0.09	1.31
Na <sub>2</sub> O	%	FUS-ICP	0.26	1.97	0.1	0.08	0.13	3.19	0.09	0.75
K <sub>2</sub> O	%	FUS-ICP	3.28	2.29	3.68	2.66	3.99	3.43	1.33	1.86
TiO <sub>2</sub>	%	FUS-ICP	0.162	0.144	0.169	0.12	0.181	0.253	0.13	0.109
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.01	0.01	0.02	0.01	0.01	0.05	0.01	0.02
LOI	%	FUS-ICP	3.58	2.93	3.67	3.02	3.85	2.79	11.62	3.68
Total	%	FUS-ICP	99.59	98.15	98.23	99.21	98.95	100.6	87.63	99.71
Ba	ppm	FUS-ICP	1060	1408	1982	1642	1410	764	945	682
Sr	ppm	FUS-ICP	8	22	6	5	10	138	5	35
Y	ppm	FUS-ICP	43	39	47	29	42	73	28	30
Sc	ppm	FUS-ICP	12	10	12	8	13	19	8	8
Zr	ppm	FUS-ICP	155	145	174	111	162	245	103	103
Be	ppm	FUS-ICP	1	1	1	< 1	2	2	< 1	< 1
V	ppm	FUS-ICP	< 5	< 5	< 5	< 5	13	7	21	< 5
Hg	ppb	FIMS	606	12	96	641	222	16	4050	250
Cr	ppm	HF/HNO3	< 8.9	< 8.9	<8.9	15.90	< 8.9	< 8.9	< 8.9	36.47
Ni	ppm	HE/HNO3	<0.9	<10	2.64	<0.9	<0.9	<0.9	60.92 <10	1.74
Cu	ppm	HE/HNO3	156.05	83.36	142.11	300 //	96.60	<10	18118 48	84.41
Zn	nnm	HF/HNO3	1636.20	150.64	576.76	2115.76	604 56	121.17	32065.80	859.15
As	ppm	HF/HNO3	3 12	2 12	17 17	45 27	7 31	<1.6	102.23	<1.6
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	36.86	<26.7
Br	ppm	HF/HNO3	164.94	176.10	227.26	228.59	242.65	214.24	208.28	202.61
Mo	ppm	HF/HNO3	2.70	<1	2.61	3.08	3.87	<1	11.05	7.87
Ag	ppm	HF/HNO3	<1	<1	1.42	<1	<1	<1	30.34	<1
Cd	ppm	HF/HNO3	6.21	<0.9	1.63	8.82	1.98	<0.9	112.27	2.09
Sn	ppm	HF/HNO3	6.20	5.61	11.47	28.25	11.76	4.15	28.20	3.34
Sb	ppm	HF/HNO3	1.51	0.82	2.83	4.22	1.25	0.41	44.07	1.75
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	7.43	<4.3
l	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	1.8/	1.32	3.50	2.08	3.71	2.19	2.37	2.03
PD D:	ppm	HF/HNO3	<25.5	<25.5	2 05	/8.81	<25.5	<25.5	6050.88	<25.5
DI NI	ppm	Na2O2 Sinter	0.85	0.30	2.05	0.09	1.05	0.21	41.00	0.74
IND	ppm	No2O2 Sinter	0.45	6.07	7.08	4.32	0.80	14.37	0.23	4.57
La	ppm	Na2O2 Sinter	21.20	20.65	21.64	15.37	14.68	28.08	14.47	13.57
Ce	ppm	Na2O2 Sinter	44.41	43.88	45.65	32.40	30.79	60.54	29.72	28.71
Pr	ppm	Na2O2 Sinter	5.54	5.49	5.79	4.10	3.94	7.71	3.71	3.67
Nd	ppm	Na2O2 Sinter	24.00	23.07	24.53	17.02	16.50	32.43	15.50	15.98
Sm	ppm	Na2O2 Sinter	5.76	5.55	5.88	4.15	4.15	8.05	3.81	3.93
Eu	ppm	Na2O2 Sinter	1.27	1.19	1.36	0.84	1.22	2.04	1.21	0.85
Tb	ppm	Na2O2 Sinter	1.15	1.04	1.18	0.74	0.99	1.81	0.74	0.77
Dy	ppm	Na2O2 Sinter	7.48	6.72	7.72	4.97	6.95	12.21	4.86	5.17
Но	ppm	Na2O2 Sinter	1.58	1.48	1.65	1.07	1.52	2.55	1.05	1.12
Er	ppm	Na2O2 Sinter	4.68	4.53	5.23	3.28	4.68	8.09	3.26	3.34
Tm	nnm	Na2O2 Sinter	0.72	0.68	0.79	0.53	0.71	1 24	0.49	0.51
Vh	ppm	Na2O2 Sinter	4.95	4.80	5.47	3.56	5.01	8.52	3.26	3 59
ID In	ppin	Na2O2 Sinter	0.72	4.30	0.94	0.55	0.72	1.27	0.52	0.57
Lu	ppm	Na2O2 Sinter	0.75	0.74	0.84	0.55	0.75	1.27	0.32	0.37
HI	ppm	Na2O2 Sinter	4.80	4.70	4./1	3.20	4.72	8.96	4.11	3.70
la	ppm	Na2O2 Sinter	0.28	0.25	0.28	0.18	0.26	0.53	0.19	0.20
Th	ppm	ma202 Sinter	5.67	5.59	6.11	4.14	5.44	9.84	5.28	3.81
AIOH WL	nm	Terraspec	2,210	2,213	2,211	2,212	2,210	2,205	#DIV/0!	2,206
AIOH Depth	%	1 erraspec	29.25%	14.85%	12.28%	12.34%	25.58%	30.53%	#DIV/0!	23.90%
FeOH WL FeOH Denth	nm %	Terraspec	2,252	2,252	2,250	2,253	2,251	2,200	2,251	2,251
AlOH denth/FaOH denth	/0	Terraspec	2.7070 Q Q1	2.4270 6.14	7 51	6 35	1.4470	5.5570	#DIV/01	1.5/70
Alon depuir con depui		Terraspee	9.01	0.14	7.51	0.55	17.70	5.52	#D1 ¥/0:	17.40
Abbreviations:	Units			FT= Footwall to	uff	Alteration			Other	
	HFL=	Hanging wall flow	w	FBX= Footwall	breccia	Qtz-Ser= Quar	tz-sericite		WL= Waveler	igth
	HBX	= Hanging wall bre	ccia	FFL= Footwall	flow	Chl-Ser= Chlor	rite-sericite		* Indicates le	ast altered
	FLT=	Footwall lapilli tu	ff	INT= Intrusion		Chl=Chlorite				

	nnondiv	D.	Tabla	<b>D1</b>	1 X	Vho	la raa	17.6	Cooo	homist	ry and	Torre	(SDOO <sup>TM</sup>	Data
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Appendix B: Table B1.1 whole-rock Geochemistry and Terraspec L						ala				
Sample ID			H501285	H501286	H501287	H501288	H501289	H501290	H501291	H501292
Hole ID			BD10-21	BD10-26	BD10-26	BD10-26	BD10-26	BD10-26	BD11-146	BD11-146
Depth (m)			31.1	9.2	18.5	20.6	29.1	37	6.2	11.8
Lithology			FBX	FLT	FLT	FLT	FLT	FFL	FLT	FLT
Alteration			Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	67.49	81.44	70.94	67.29	69.08	75.62	77.68	75.31
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	15.24	9.57	16.29	21.34	11.18	12.33	10.62	11.99
$Fe_2O_3(T)$	%	FUS-ICP	2.98	2.49	2.43	0.42	9.31	3.86	2.53	3.11
MnO	%	FUS-ICP	0.155	0.012	0.023	0.023	0.029	0.027	0.024	0.08
MgO	%	FUS-ICP	4.55	0.86	1.7	0.9	0.43	2.01	1.37	1.43
CaO	%	FUS-ICP	0.59	0.03	0.05	0.04	0.04	0.05	0.03	0.04
Na <sub>2</sub> O	%	FUS-ICP	0.42	0.1	0.16	0.23	0.12	0.12	0.08	0.1
K <sub>2</sub> O	%	FUS-ICP	3.74	2.93	4.79	6.4	3.24	3.16	3.05	3.56
TiO <sub>2</sub>	%	FUS-ICP	0.206	0.15	0.217	0.301	0.15	0.166	0.147	0.176
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.03	0.02	0.03	< 0.01	0.02	0.03	0.01	< 0.01
LOI	%	FUS-ICP	5.44	2.64	3.64	3.15	6.36	3.34	3.14	3.67
Total	%	FUS-ICP	100.8	100.2	100.3	100.1	99.95	100.7	98.69	99.48
Ba	ppm	FUS-ICP	1443	1447	1999	2603	1475	1092	1327	1473
Sr	ppm	FUS-ICP	25	6	7	12	5	6	5	5
Y	ppm	FUS-ICP	57	28	68	18	103	45	32	38
Sc	ppm	FUS-ICP	16	10	19	12	13	12	10	12
Zr	ppm	FUS-ICP	202	147	203	276	149	163	134	155
Be	ppm	FUS-ICP	1	< 1	1	1	< 1	< 1	1	1
V	ppm	FUS-ICP	< 5	27	7	7	7	< 5	28	6
Hg	ppb	FIMS	98	214	903	45	132	7	522	342
Cr	ppm	HF/HNO3	34.22	27.96	32.87	26.92	15.42	28.18	<8.9	<8.9
Co	ppm	HF/HNO3	1.08	0.96	0.96	<0.9	6.05	1.28	<0.9	<0.9
Ni	ppm	HF/HNO3	20.10	18.25	20.89	16.59	<10	18.43	<10	<10
Cu Za	ppm	HF/HNO3	32.39	296.53	49.29	<19	654.56	21.31	557.08	92.69
Zfi	ppm	HF/HNO3	400.29	935.20	4140.84	133.80	53.96	72 20	51.37	948.30
Se	ppm	HE/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	nnm	HF/HNO3	230.24	197.87	219.23	195 34	230.99	192 30	222.12	215 74
Mo	ppm	HF/HNO3	10.08	6.59	11.00	9.89	4 10	7 75	4 31	3 96
Ag	ppm	HF/HNO3	<1	<1	1.50	1.01	1.94	<1	<1	1.06
Cď	ppm	HF/HNO3	<0.9	3.16	13.49	<0.9	1.17	<0.9	7.36	2.84
Sn	ppm	HF/HNO3	5.06	5.39	120.76	10.30	9.68	16.31	10.28	12.32
Sb	ppm	HF/HNO3	2.20	3.29	10.42	0.95	13.31	5.17	6.20	6.52
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	3.45	3.99	3.46	6.69	6.62	3.76	3.65	4.76
Pb	ppm	HF/HNO3	<25.3	264.73	1353.53	<25.3	83.87	<25.3	375.03	554.99
Bı	ppm	HF/HNO3	0.27	3.50	2.35	0.24	0.77	4.65	2.35	0.21
Nb	ppm	Na2O2 Sinter	8.22	5.92	7.86	8.91	8.28	6.52	5.42	6.09
La	ppm	Na2O2 Sinter	21.38	20.35	29.34	1.72	26.23	19.53	19.98	22.49
Ce	ppm	Na2O2 Sinter	45.49	41.62	62.91	3.12	51.99	41.13	40.79	47.48
Pr	ppm	Na2O2 Sinter	5.67	5.19	7.98	0.35	6.38	5.11	5.10	5.92
Nd	ppm	Na2O2 Sinter	24.90	22.55	33.99	1.39	26.86	21.99	21.67	24.74
Sm	ppm	Na2O2 Sinter	6.09	5.01	8.25	0.48	6.87	5.48	5.01	5.68
Eu	ppm	Na2O2 Sinter	1.59	1 47	1 36	0.07	2.17	0.97	1.50	1 45
Th	nnm	Na2O2 Sinter	1 38	0.72	1.56	0.23	1.76	1.02	0.87	0.99
Dv	ppm	Na2O2 Sinter	0.15	4.62	10.28	2.01	13.44	6.90	5.52	6.70
Dy He	ppin	Na2O2 Sinter	2.02	1.02	2.25	2.01	2.04	1.50	1.16	1.40
Б	ppm	Na O Sinter	2.05	1.02	2.23	0.34	3.04	1.30	1.10	1.49
Er	ppm	$N_{2}O_{2}$ Sinter	6.17	3.18	6.83	2.03	10.04	4.63	3.53	4.82
Tm	ppm	Na2O2 Sinter	0.93	0.53	1.05	0.36	1.61	0.70	0.55	0.78
Yb	ppm	Na2O2 Sinter	6.26	3.84	6.81	2.83	11.68	4.72	3.76	5.70
Lu	ppm	Na2O2 Sinter	0.96	0.59	1.07	0.45	1.88	0.72	0.56	0.90
Hf	ppm	Na2O2 Sinter	5.30	3.82	4.91	8.82	11.98	3.89	3.98	4.47
Та	ppm	Na2O2 Sinter	0.34	0.24	0.33	0.39	0.28	0.28	0.22	0.25
Th	ppm	Na2O2 Sinter	6.90	4.94	6.50	8.75	6.22	5.48	5.00	5.85
AlOH WL	nm	Terraspec	2,208	2,212	2,210	2,204	2,205	2,205	2,211	2,211
AlOH Depth	%	Terraspec	12.68%	19.88%	16.78%	47.73%	4.64%	12.93%	10.53%	11.79%
FeOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	2,254	#DIV/0!	#DIV/0!
FeOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	2.77%	#DIV/0!	#DIV/0!
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	4.67	#DIV/0!	#DIV/0!
Abbreviations: Units HFL= Hanging wall flow HBX= Hanging wall breecia		FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow		Alteration Qtz-Ser= Quartz-sericite Chl-Ser= Chlorite-sericite		<b>Other</b> WL= Wavelength				
	FLT=	FLT= Footwall lapilli tuff		INT= Intrusion	n	Chl=Chlorite				

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Annondiv D. Tabla D1 1	Whole real (	ooohomistry.	and Tarragnaa <sup>+</sup>	" Data
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тррения	<b>D</b> .		51 VVIIO		GLUCI	iennisei j	y anu i	crrasp		ata
Sample ID			H501293	H501294	H501295	H501296	H501297	H501298	H501299	H501300
Hole ID			BD11-146	BD11-143	BD11-143	BD11-143	BD11-143	BD11-134	BD11-134	BD11-134
Depth (m)			14.9	4.3	7.9	16.7	28.1	10.9	21.5	33.1
Lithology			FLT	FLT	FLT	FLT	FFL	FLT	FLT	FLT
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	55.61	75.69	78.11	79.36	80.81	62.43	53.49	74.01
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	8.65	7.7	7.89	7.94	10.22	10.21	11.38	9.35
$Fe_2O_3(T)$	%	FUS-ICP	18.16	7.86	6.11	4.83	1.8	13.91	11.46	8.32
MnO	%	FUS-ICP	0.065	0.007	0.027	0.017	0.014	0.023	0 193	0.007
MgO	%	FUS-ICP	1 44	0.38	2.3	17	1 29	1 48	12.37	0.38
CaO	%	FUS-ICP	0.02	0.03	0.02	0.03	0.03	0.03	1.43	0.03
Na <sub>2</sub> O	%	FUS-ICP	0.07	0.11	0.08	0.09	0.12	0.13	0.02	0.14
K-0	0/0	FUS-ICP	2.51	2.14	1.61	1.86	2 74	2 71	0.15	2.69
TiO	0/2	FUS-ICP	0.122	0.118	0.103	0.116	0.147	0.146	0.179	0.137
110 <sub>2</sub>	70 07	FUS ICD	6.122	6.118	0.105	0.110	0.147	6.140	0.175	6.157
P <sub>2</sub> O <sub>5</sub>	70	FUS-ICP	< 0.01	< 0.01	0.01	< 0.01	0.02	< 0.01	0.02	< 0.01
LOI	70 0/	FUS-ICP	08.14	5.47	3.96	3.22	2.2	0.77	9.03	3.20
Pa	70	FUS-ICP	98.14	99.3 608	520	99.18 603	99.4	1000	100.5	762
Da Sr	ppm	FUS-ICP	1000	5	320	4	6	8	33	703
v	ppm	FUS-ICP	16	27	20	33	36	36	16	3/
Sc	nnm	FUS-ICP	12	8	8	9	11	11	13	10
Zr	ppm	FUS-ICP	120	105	103	111	136	130	152	121
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
v	ppm	FUS-ICP	19	< 5	< 5	< 5	< 5	< 5	6	< 5
Hg	ppb	FIMS	1410	12	11	9	< 5	30	16	18
Cr	ppm	HF/HNO3	<8.9	29.63	<8.9	34.09	30.72	9.09	<8.9	<8.9
Co	ppm	HF/HNO3	<0.9	19.63	12.46	14.29	2.49	8.06	4.15	10.26
Ni	ppm	HF/HNO3	<10	16.30	<10	19.58	19.84	<10	<10	<10
Cu	ppm	HF/HNO3	177.31	3026.08	963.65	1147.85	47.36	702.85	139.90	22.90
Zn	ppm	HF/HNO3	1794.40	84.39	82.09	88.66	68.46	114.18	348.40	26.90
As	ppm	HF/HNO3	481.71	18.34	16.13	33.61	1.97	65.73	6.24	34.43
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	249.22	185.88	216.22	233.90	211.15	265.11	251.45	264.97
Mo	ppm	HF/HNO3	2.89	9.20	3.81	13.69	9.92	4.49	5.29	4.41
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Cd	ppm	HF/HNO3	5.71	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	10.03	4.17	3.81	4.10	4.18	6.04	3.15	3.94
Sb	ppm	HF/HNO3	26.83	1.66	1.27	1.38	0.82	1.96	1.67	2.11
le	ppm	HF/HNO3	<4.3	7.92	4.32	<4.3	<4.3	12.05	<4.3	<4.3
l W	ppm	HF/HNO3	< 3 /	<3/	<3/	<3/	< 3 /	< 3 /	<3/	<3/
w Ph	ppm	HE/HNO3	59.29	/.0/	25.2	25.2	0.33	72.04	/./1	4.00
Bi	ppm	HE/HNO3	0.19	<23.5 0.01	5 23	<23.3 4.13	1.82	17.73	40.15	<23.3 4.07
NIL	ppm	Na2O2 Sinter	4.60	4.70	5.25	5.26	5.92	7.42	10.00	5.04
ND	ppm	Na2O2 Sinter	4.60	4.70	3.34	3.20	5.82	12.66	10.09	3.94
La	ppm	Na2O2 Sinter	12.41	10.38	13.01	13.24	16.30	13.66	20.32	15.11
Ce	ppm	Na2O2 Sinter	25.64	21.78	28.69	27.67	32.67	29.51	43.41	32.13
Pr	ppm	Na2O2 Sinter	3.13	2.69	3.65	3.47	4.08	3.76	5.49	4.09
Nd	ppm	Na2O2 Sinter	13.18	11.00	15.40	14.54	18.17	16.06	22.96	17.16
Sm	ppm	Na2O2 Sinter	2.79	2.64	3.68	3.63	4.73	4.17	6.02	4.32
Eu	ppm	Na2O2 Sinter	0.66	0.73	0.98	0.88	1.02	1.01	1.48	0.84
Тb	ppm	Na2O2 Sinter	0.41	0.60	0.71	0.82	0.94	0.87	1.20	0.77
Dv	ppm	Na2O2 Sinter	2 73	4 34	4 48	5 60	5 95	6.03	7 59	5 19
Ho	nnm	Na2O2 Sinter	0.60	0.99	0.96	1 20	1.28	1.22	1.64	1.06
E.	ppin	Na.O. Sinter	1.05	2.04	2.97	2.65	1.20	2.01	5.01	2.16
EI	ppm	No2O2 Sinter	1.93	3.04	2.87	3.03	4.04	3.91	5.01	5.10
Im	ppm	Na2O2 Sinter	0.32	0.45	0.44	0.56	0.60	0.61	0.75	0.50
Yb	ppm	Na2O2 Sinter	2.20	3.17	3.00	3.65	4.23	4.08	5.24	3.41
Lu	ppm	Na2O2 Sinter	0.33	0.49	0.46	0.54	0.69	0.67	0.83	0.52
Hf	ppm	Na2O2 Sinter	3.45	3.16	4.07	2.80	4.46	4.98	6.20	3.84
Та	ppm	Na2O2 Sinter	0.17	0.17	0.22	0.19	0.24	0.22	0.30	0.21
Th	ppm	Na2O2 Sinter	3.63	3.83	4.59	3.93	4.85	5.28	6.28	4.35
AlOH WL	nm	Terraspec	2,211	2,204	2,201	2,202	2,204	2,205	#DIV/0!	2,205
AlOH Depth	%	Terraspec	13.52%	20.60%	16.25%	17.58%	33.75%	14.50%	#DIV/0!	21.43%
FeOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	2,253	2,253	2,254	2,253	2,250	#DIV/0!
FeOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	6.28%	4.56%	2.03%	1.28%	17.43%	#DIV/0!
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	2.59	3.85	16.65	11.33	#DIV/0!	#DIV/0!
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	<ul> <li>Hanging wall flow</li> </ul>	w .	FBX= Footwa	II breccia	Qtz-Ser= Qua	rtz-sericite		WL= Waveler	ngth
	HBX	= Hanging wall bre	ccia	FFL= Footwal	I flow	Chi-Ser= Chlo	orite-sericite			

Chl=Chlorite

FLT= Footwall lapilli tuff

Annondiv B	• Tahla	R1 1	Whole_roc	k Gooch	omistry	and Torr	asnee <sup>TM</sup>	Nata
Appendix B	: I able	<b>DI</b> .I	w noie-roc	k Geoch	lemistry	and terr	aspec	Data

Appendix B: Table B1.1 whole-rock Geochemistry and Terraspec Data								ata		
Sample ID			H501301	H501302	H501303	H501304	H501305	H501306	H501307	H501308
Hole ID			BD11-134	BD11-154	BD11-154	BD11-154	BD11-158	BD11-158	BD11-158	BD11-158
Depth (m)			43	10.9	26.1	30.5	5.8	14	25.4	39.8
Lithology			FBX	FLT	FLT	FLT	FLT	FLT	FLT	FLT
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	70.04	75.55	76.35	76.39	22.68	44.36	75.98	72.15
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	9.54	12.77	10.4	12.91	15.85	8.25	10.21	11.75
$Fe_2O_3(T)$	%	FUS-ICP	10.29	4.29	5.45	3.24	17.59	8.53	3.4	4.34
MnO	%	FUS-ICP	0.031	0.007	0.013	0.012	0.1	0.204	0.036	0.047
MgO	%	FUS-ICP	1.5	0.46	0.48	0.61	22.54	15.7	5.27	3.85
CaO	%	FUS-ICP	0.12	0.03	0.05	0.04	0.21	6.53	0.08	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.12	0.2	0.18	0.2	0.03	0.01	0.06	0.08
K <sub>2</sub> O	%	FUS-ICP	2.49	3.72	2.96	3.76	0.03	0.02	1.85	2.6
TiO <sub>2</sub>	%	FUS-ICP	0.144	0.185	0.156	0.18	0.23	0.19	0.143	0.171
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	< 0.01	< 0.01	< 0.01	0.01	0.08	0.01	0.02
LOI	%	FUS-ICP	6.41	3.74	4.07	3.38	14.93	14.36	3.85	4.01
Total	%	FUS-ICP	100.7	101	100.1	100.7	94.2	98.22	100.9	99.07
Ba	ppm	FUS-ICP	728	1611	1022	1395	19	15	624	657
Sr	ppm	FUS-ICP	9	12	10	13	4	120	4	5
Y	ppm	FUS-ICP	35	16	19	27	23	54	36	43
Sc	ppm	FUS-ICP	10	8	8	13	14	11	10	13
Zr	ppm	FUS-ICP	125	160	139	156	200	198	123	152
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	< 5	< 5	6	7	< 5	< 5	6	5
Hg	ррь	FIMS UE/UNO2	30	12.67	12.02	17.01	2530	/4	14	/
Co	ppm	HF/HNO3	2 70	2.14	6 77	8 44	<0.9 80.15	<0.9 18 63	<0.9 1.50	<8.9 1.00
Ni	ppm	HE/HNO3	2.79 <10	<10	<10	6.44 <10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	731 55	<10	<10	<10	23776.05	310.99	145 74	131.84
Zn	ppm	HF/HNO3	52 37	32.65	33 15	32.02	9964 99	686.97	176 31	238 35
As	ppm	HF/HNO3	175.59	17.96	30.15	6.84	106.22	25.32	6.53	46.66
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	251.86	248.03	238.46	230.34	295.54	236.62	315.08	259.59
Mo	ppm	HF/HNO3	2.74	6.26	4.58	9.30	<1	8.16	3.30	3.67
Ag	ppm	HF/HNO3	<1	1.12	<1	<1	5.76	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	36.17	1.45	<0.9	<0.9
Sn	ppm	HF/HNO3	4.57	5.19	4.76	6.76	14.22	8.09	14.92	9.14
Sb	ppm	HF/HNO3	11.04	1.29	1.19	1.64	1.76	<0.3	1.19	2.30
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	11.47	<4.3	<4.3	<4.3
l W	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	2.15	4.02	4.27	5.28	5.25	1.92	0.74	5.14
Bi	ppm	HE/HNO3	7 28	3.05	3 3 3	1.67	~25.5	0.83	1.05	2.03
DI Nik	ppm	Na2O2 Sinter	6.52	9.75	7 49	8.06	8.20	6.75	5.00	2.05
NU	ppm	No2O2 Sinter	0.32	8.57	7.46	8.00	8.50	0.73	3.00	0.11
La	ppm	Na2O2 Sinter	15.19	7.82	5.75	19.40	5.01	46.75	17.17	19.75
Ce	ppm	Na2O2 Sinter	32.12	12.93	7.42	36.56	6.18	98.67	35.73	41.61
Pr	ppm	Na2O2 Sinter	4.01	1.40	0.89	4.20	0.80	12.27	4.45	5.28
Nd	ppm	Na2O2 Sinter	16.79	5.35	3.59	16.58	3.60	51.98	18.80	22.47
Sm	ppm	Na2O2 Sinter	4.30	1.04	0.92	3.26	1.09	11.61	4.49	5.35
Eu	ppm	Na2O2 Sinter	1.23	0.34	0.29	0.69	0.18	1.20	1.32	1.41
Tb	ppm	Na2O2 Sinter	0.83	0.30	0.33	0.66	0.37	1.69	0.86	1.05
Dy	ppm	Na2O2 Sinter	5.65	2.37	2.65	4.39	2.98	9.91	5.71	6.93
Но	ppm	Na2O2 Sinter	1.19	0.59	0.64	0.95	0.74	1.97	1.26	1.51
Er	ppm	Na2O2 Sinter	3.52	2.22	2.28	3.16	2.57	5.81	3.81	4.55
Tm	ppm	Na2O2 Sinter	0.55	0.40	0 39	0.50	0.42	0.87	0.58	0.70
Vh	nnm	Na2O2 Sinter	3.65	3 27	2 97	3 79	3 31	5 78	4 04	4 80
In In	ppm	Na2O2 Sinter	0.58	0.48	0.46	0.50	0.55	0.00	0.62	0.73
Lu	ppm	Na2O2 Sinter	4.20	0.48	0.40	0.59	6.16	5.25	2.59	2.66
HI	ppm	No2O2 Sinter	4.39	4.54	4.41	4.94	0.10	5.25	3.58	3.00
1a	ppm	Na2O2 Sinter	0.24	0.32	0.28	0.31	0.31	0.28	0.21	0.26
1h	ppm	Na2O2 Sinter	4.72	5.99	5.34	6.37	6.22	6.52	4.63	5.52
AIOH WL	nm	Terraspec	2,204	2,204	2,204	2,205	#DIV/0!	#DIV/0!	2,207	2,207
AIOH Depth	%	1 erraspec	11.60%	22.75% #DIV/01	25.58%	30.14%	#DIV/0!	#DIV/0!	14.30%	10.06%
FeOH WL FeOH Donth	nm o/	Тегтазрес	2,255	#DIV/0!	#DIV/0!	#DIV/0!	2,249	2,248	2,231	2,252
AlOH denth/FeOH denth	70	Terraspec	0.50	#DIV/0!	#DIV/0! #DIV/0!	#DIV/0!	4.22% #DIV/01	1./1% #DIV/01	0.05% 2.00	2.60% 2.61
raon acpuirteon acpui		remaspec	7.37	πD1 V/U!	πD1¥/0!	πD1 v/0!	πD1 ¥/0!	#D11/0!	4.07	2.01
Abbreviations:	Units			FT= Footwall tuff		Alteration		Other		
	HFL=	Hanging wall flor	w	FBX= Footwa	ll breccia	Qtz-Ser= Qua	rtz-sericite		WL= Waveler	ıgth
	HBX	= Hanging wall bre	ccia	FFL= Footwal	l flow	Chl-Ser= Chlo	rite-sericite			
	FLT=	Footwall lapilli tu	ff	INT= Intrusion	n	Chl=Chlorite				

Annend's D. Table D1 1	Whale we als Case ab amigtory		
Appendix B: Table B1.1	Whole-rock Geochemistry a	and Terraspec <sup>111</sup> Da	ata

Appendix	<b>D.</b> I	able D1.		C-IUCK	Geoti	iennsti j	y anu l	i ei i asp	Let L	Jala
Sample ID			H501309	H501310	H501311	H501312	H501313	H501314	H501315	H501316
Hole ID			BD11-136	BD11-136	BD11-136	BD11-195	BD11-195	BD11-195	BD11-195	BD10-6
Depth (m)			10.7	23.4	31.9	7.9	9.5	16.8	23.8	4.6
Lithology			FLT	FLT	FLT	FLT	FFL	FLT	FFL	HFL
Alteration			Qtz-Ser	Chl	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	73.45	25.79	72.61	73.61	68.86	75.05	72.89	76.59
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	7.25	20.66	10.28	10.08	11.99	10.82	12.03	10.21
$Fe_2O_3(T)$	%	FUS-ICP	8.61	15.94	6.84	6.95	6.77	5.02	4.25	2.24
MnO	%	FUS-ICP	0.014	0.136	0.007	0.014	0.071	0.019	0.049	0.1
MgO	%	FUS-ICP	0.68	22.87	0.77	0.56	4.4	1.52	3.54	3.29
CaO	%	FUS-ICP	0.03	0.04	0.04	0.04	0.06	0.04	0.05	0.05
Na <sub>2</sub> O	%	FUS-ICP	0.1	< 0.01	0.13	0.11	0.09	0.1	0.09	0.08
K-0	%	FUS-ICP	2	< 0.01	2.89	3.02	2.51	2.91	2.92	2.5
TiO	0/2	FUS-ICP	0 105	0.317	0.148	0.144	0.191	0.144	0.167	0.126
110 <sub>2</sub>	0/	FUS ICD	< 0.01	0.01	0.140	0.144	0.171	< 0.01	0.107	0.120
P <sub>2</sub> O <sub>5</sub>	70 0/	FUS-ICP	< 0.01	12.8	5.27	4.08	5.79	< 0.01	0.02	2.11
LOI	70 0/.	FUS-ICP	0.05	12.8	00.01	4.98	3.78	4.00	4.19	08.22
10tai Ba	70 DDM	FUS-ICP	1/20	13	1082	99.52	779	037	820	2076
Sr.	ppm	FUS-ICP	7	< 2	8	5	5	6	6	2070
V	ppm	FUS-ICP	46	49	41	36	45	36	44	39
Sc	ppm	FUS-ICP	10	17	11	10	12	11	13	9
Zr	ppm	FUS-ICP	94	274	130	128	154	127	157	127
Be	ppm	FUS-ICP	< 1	< 1	<1	< 1	< 1	< 1	< 1	1
V	ppm	FUS-ICP	< 5	6	< 5	< 5	< 5	< 5	< 5	< 5
Hg	ppb	FIMS	117	< 5	13	45	19	34	28	11
Cr	ppm	HF/HNO3	14.86	<8.9	<8.9	13.19	<8.9	<8.9	<8.9	<8.9
Co	ppm	HF/HNO3	2.84	34.35	5.34	<0.9	<0.9	<0.9	<0.9	1.06
Ni	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	1782.60	101.31	92.00	2618.66	103.67	200.60	<19	70.07
Zn	ppm	HF/HNO3	306.96	511.51	76.65	178.41	174.03	72.41	97.77	180.71
As	ppm	HF/HNO3	203.53	48.49	10.07	70.64	9.15	6.39	4.26	2.00
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	278.55	317.36	271.89	225.88	240.39	241.35	227.99	255.04
Mo	ppm	HF/HNO3	7.23	5.18	3.06	5.76	3.58	3.64	4.05	<1
Ag	ppm	HF/HNO3	1.01	1.06	<1	<1	<1	<1	<1	1.30
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	21.43	13.46	13.33	4.40	3.78	4.54	3.24	12.94
Sb	ppm	HF/HNO3	12.71	1.08	0.99	4.16	3.74	3.86	2.40	1.96
le	ppm	HF/HNO3	<4.5	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
I W	ppm	HF/HNO3	<3/ 2.05	<3/ 17.26	<3/ 2.07	<37	<3/	<3/	<37 2.67	< 37
Ph	ppm	HE/HNO3	2.05	<25.3	<2.57	26.30	<25.3	<25.3	<25.3	128.82
Bi	ppm	HF/HNO3	18 59	1 78	2 58	2 19	1.52	1 79	0.40	1 94
DI Nik	ppin	Na2O2 Sinter	2.42	12.70	5.12	2.1)	0.71	6.24	7.70	5.00
NU	ppm	Na2O2 Sinter	3.42	13.79	3.15	7.40	9.71	0.34	7.70	3.09
La	ppm	Na2O2 Sinter	25.54	24.68	21.50	16.14	19.80	18.77	20.06	18.06
Ce	ppm	Na2O2 Sinter	57.09	49.26	45.44	33.83	41.15	37.87	42.33	37.11
Pr	ppm	Na2O2 Sinter	7.58	5.89	5.87	4.32	5.16	4.74	5.34	4.75
Nd	ppm	Na2O2 Sinter	32.96	24.56	25.53	18.29	22.34	20.29	22.63	19.98
Sm	ppm	Na2O2 Sinter	7.98	5.38	6.27	4.17	5.26	4.77	5.47	4.90
Eu	ppm	Na2O2 Sinter	1.70	1.10	1.07	1.22	1.35	1.14	1.56	1.18
Tb	ppm	Na2O2 Sinter	1.28	1.14	1.05	0.91	1.22	0.86	1.13	0.94
Dv	ppm	Na2O2 Sinter	7.70	8.17	6.60	6.07	8.20	5.70	7.50	6.41
Но	nnm	Na2O2 Sinter	1 54	1 94	1 33	1.28	1 73	1 24	1.66	1.36
Fr.	ppm	Na.O. Sinter	1.54	6.47	4.06	4.02	5.06	2.02	5.02	4.20
EI Ter	ppm	No2O2 Sinter	4.40	1.02	4.00	4.02	5.00	0.50	5.05	4.29
1111	ppm	Na2O2 Sinter	0.04	1.08	0.60	0.03	0.80	0.39	0.78	0.00
Yb	ppm	Na2O2 Sinter	4.28	7.70	3.95	4.10	5.33	4.02	5.38	4.40
Lu	ppm	Na2O2 Sinter	0.63	1.23	0.62	0.62	0.80	0.62	0.77	0.71
Hf	ppm	Na2O2 Sinter	2.52	9.60	3.68	4.02	3.92	4.49	5.27	4.31
Та	ppm	Na2O2 Sinter	0.14	0.52	0.21	0.26	0.35	0.24	0.30	0.23
Th	ppm	Na2O2 Sinter	3.02	11.75	4.48	5.26	6.42	5.24	6.43	4.64
AlOH WL	nm	Terraspec	2,206	#DIV/0!	2,206	2,208	2,207	2,205	2,207	2,213
AlOH Depth	%	Terraspec	10.24%	#DIV/0!	19.40%	13.70%	15.90%	9.60%	12.40%	19.93%
FeOH WL	nm	Terraspec	#DIV/0!	2,250	#DIV/0!	#DIV/0!	2,253	2,252	2,251	2,252
FeOH Depth	%	Terraspec	#DIV/0!	12.39%	#DIV/0!	#DIV/0!	1.46%	2.95%	3.64%	2.16%
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	10.87	3.25	3.41	9.24
Abbreviations:	Units HFL=	<ul> <li>Hanging wall flow</li> </ul>	r .	FT= Footwall FBX= Footwa	l tuff all breccia	Alteration Qtz-Ser= Qua	rtz-sericite		Other WL= Waveler	ngth

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Appendix B:	I able BLL	whole-rock	Geochemistry	and Lerrasn	oec Data

HFL= Hanging wall now HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow Chl-Ser= Chlorite-sericite INT= Intrusion Chl=Chlorite

ampic ID			H501317	H501318	H501319	H501320	H501326	H501327	H501328	H50132
lole ID Jonth (m)			BD10-6	BD10-6	BD10-6	BD10-6	BD10-113	BD10-113	BD10-113	BD10-11
ithology			FLT	FLT	FBX	FBX	HFL	FLT	FLT	40.8 FLT
Iteration			Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Qtz-Ser	Qtz-Sei
iO <sub>2</sub>	%	FUS-ICP	65.02	72.29	76.48	75.46	78.95	53.65	72.47	76.4
$l_2O_3$	%	FUS-ICP	11.35	12.49	11.31	12.51	11.22	11.33	13.27	10.32
$e_2O_3(T)$	%	FUS-ICP	8.49	3.9	3.01	2.34	1.6	11.18	2.4	1.9
4nO	%	FUS-ICP	0.164	0.158	0.136	0.089	0.024	0.269	0.183	0.27
/IgO	%	FUS-ICP	3.11	1.95	1.55	2.24	2.38	11.35	2.83	2.25
aO	%	FUS-ICP	0.04	0.06	0.23	0.14	0.03	0.04	0.35	0.34
la <sub>2</sub> O	%	FUS-ICP	0.08	0.12	0.67	0.22	0.09	0.03	0.66	0.11
20	%	FUS-ICP	2.56	3.54	3.1	3.55	3	0.48	3.3	2.98
iO <sub>2</sub>	%	FUS-ICP	0.14	0.178	0.156	0.159	0.129	0.162	0.182	0.16
2O5	%	FUS-ICP	0.01	0.02	0.02	0.02	0.03	0.02	0.03	0.05
OI	%	FUS-ICP	6.14	4.23	3.62	3.61	2.61	9.51	4.23	3.81
otal	%	FUS-ICP	97.1	98.94	100.3	100.3	100.1	98.02	99.9	98.59
a	ppm	FUS-ICP	1776	1881	1737	1175	2481	344	1437	1206
r	ppm	FUS-ICP	5	7	15	9	5	< 2	18	17
	ppm	FUS-ICP	36	41	33	28	37	37	39	39
c	ppm	FUS-ICP	11	13	10	11	10	10	13	11
1	ppm	FUS-ICP	120	154	158	151	130	150	101	147
e	ppm	FUS-ICP	22	12	1	1	1	< 1	1	1
a	ppin	FIMS	1580	357	313	546	30	14	307	103
ъ r	ppo	HF/HNO3	12.73	<8.9	<8.9	11.20	<8.9	<8.9	<8.9	<8.9
0	ppm	HF/HNO3	18.74	1.20	<0.9	<0.9	<0.9	30.03	2.73	0.92
i	ppm	HF/HNO3	<10	<10	<10	<10	<10	<10	<10	<10
'n	ppm	HF/HNO3	2252.77	223.85	363.20	626.73	231.49	2080.79	226.52	128.03
n	ppm	HF/HNO3	5509.07	1355.64	1355.19	1698.80	298.16	796.81	1897.05	609.64
S	ppm	HF/HNO3	183.87	55.90	24.80	14.20	4.10	109.44	35.94	21.95
e	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
r	ppm	HF/HNO3	267.29	296.89	269.17	258.91	263.76	259.46	284.07	270.59
lo	ppm	HF/HNO3	18.90	5.69	2.66	4.15	2.04	8.04	2.97	2.38
g	ppm	HF/HNO3	9.78	1.28	<1	<1	<1	4.40	<1	<1
a	ppm	HF/HNO3	13.47	4.57	4.75	5.91	< 0.9	5.55	5.05 21.50	1.44
h	ppm	HE/HNO3	25.57	13.13	17.33	9.39	3.95	10.53	3.42	2 20
e	nnm	HF/HNO3	<4 3	<4 3	<4 3	<4 3	<4 3	4 61	<4 3	<4 3
	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
1	ppm	HF/HNO3	3.32	2.95	2.75	3.64	2.39	3.48	1.70	1.20
b	ppm	HF/HNO3	1331.20	49.99	84.52	<25.3	88.35	195.20	145.10	<25.3
i	ppm	HF/HNO3	9.38	1.74	1.67	0.56	0.60	14.69	0.66	0.16
b	ppm	Na2O2 Sinter	5.04	6.46	5.83	6.03	5.03	6.51	7.26	6.09
a	ppm	Na2O2 Sinter	16.75	21.49	16.93	19.75	18.87	21.93	24.68	19.45
e	ppm	Na2O2 Sinter	34.47	45.15	35.95	41.70	39.06	45.22	52.15	40.95
r	ppm	Na2O2 Sinter	4.35	5.70	4.51	5.29	5.01	5.73	6.58	5.18
d	ppm	Na2O2 Sinter	18 36	23.98	19.04	22.05	21.52	23.91	28.06	21.96
m	ppm	Na2O2 Sinter	4 48	5 69	4 61	5.03	5 15	5 57	6 44	5 31
	ppm	Na2O2 Sinter	1 41	1 47	1.05	1.00	1 11	1 49	1 72	1 32
• h	PDm	Na2O2 Sinter	0.87	1.06	0.84	0.70	0.05	1.02	1.72	1.52
v	ppm	Na2O2 Sinter	5 20	7 16	5 77	5 11	6.24	6.92	7.24	7.15
y	ppm	Na2O2 Sinter	5.80	/.10	3.72	3.11	0.54	0.85	1.24	1.15
U	ppm	No O Sinter	1.25	1.55	1.21	1.0/	1.39	1.40	1.54	1.56
r	ppm		3.92	4.73	3.71	5.45	4.33	4.48	4.85	4.70
m	ppm	Na2O2 Sinter	0.58	0.74	0.57	0.54	0.65	0.68	0.76	0.72
b	ppm	Na2O2 Sinter	4.04	5.00	4.00	3.79	4.47	4.62	5.25	4.87
u	ppm	Na2O2 Sinter	0.62	0.78	0.63	0.62	0.67	0.70	0.81	0.71
f	ppm	Na2O2 Sinter	3.77	4.19	4.03	3.95	3.94	3.84	5.32	4.41
a	ppm	Na2O2 Sinter	0.21	0.27	0.24	0.25	0.21	0.25	0.29	0.23
h	ppm	Na2O2 Sinter	4.18	5.38	4.78	5.01	4.88	5.43	6.32	5.23
IOH WL	nm	Terraspec	2,209	2,210	2,213	2,211	2,210	2,204	2,209	2,209
lOH Depth	%	Terraspec	10.41%	13.83%	21.15%	18.60%	18.50%	5.14%	14.95%	19.70%
eOH WL	nm	Terraspec	2,253	#DIV/0!	#DIV/0!	#DIV/0!	2,252	2,250	#DIV/0!	2,250
eOH Depth	%	Terraspec	3.39%	#DIV/0!	#DIV/0!	#DIV/0!	2.16%	4.88%	#DIV/0!	1.09%
IOH depth/FeOH depth		Terraspec	3.07	#DIV/0!	#DIV/0!	#DIV/0!	8.56	1.05	#DIV/0!	18.07
bbreviations:	Units HFL= HBX=	<ul> <li>Hanging wall flo</li> <li>Hanging wall bre</li> </ul>	w ccia	FT= Footwall FBX= Footwa FFL= Footwa	tuff all breccia ll flow	Alteration Qtz-Ser= Qua Chl-Ser= Chlo	artz-sericite orite-sericite		Other WL= Waveler	ngth

Annondiv D. Table D1 1	Whole woold (	Tooohomistmy	and Townson	TM Data
ADDENUIX D. LADIE DI.I	W HUIE-FOCK U	JEUCHEIIIISUIV	and refraspec	Data

Appendix	D	Table D1.	I WILL	пе-госк	Geoci	lennstry	anu	rerrasp	ec D	่าสเส
Sample ID			H501332	H501333	H501334	H501335	H501336	H501337	H501338	H501339
Hole ID			BD10-3	BD10-3	BD10-3	BD10-3	BD10-3	BD10-18	BD10-18	BD10-18
Depth (m)			6.6	8.2	16.3	30.7	37.9	4.3	9.9	19.7
Lithology			HFL	FLT	FLT	FLT	FLT	FLT	FFL	FBX
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	74.87	70.98	70.38	55.23	29.41	61.73	76.76	77.83
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	12.33	9.95	13.14	14.36	15.55	9.09	10.26	10.35
$Fe_2O_3(T)$	%	FUS-ICP	2.46	7.34	5.35	13.34	14.64	13.5	5.03	4.55
MnO	%	FUS-ICP	0.041	0.024	0.025	0.091	1.166	0.058	0.016	0.007
MgO	%	FUS-ICP	1.19	0.69	0.79	1.95	20.18	2.02	0.94	0.49
CaO	%	FUS-ICP	0.03	0.03	0.03	0.04	1.94	0.05	0.12	0.03
Na <sub>2</sub> O	%	FUS-ICP	0.16	0.12	0.16	0.18	< 0.01	0.07	0.09	0.09
K <sub>2</sub> O	%	FUS-ICP	3.6	2.95	3.95	4.13	0.04	2.44	3.08	3.14
TiO <sub>2</sub>	%	FUS-ICP	0.171	0.156	0.186	0.195	0.191	0.131	0.138	0.148
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	< 0.01	< 0.01	< 0.01	0.02	0.02	0.02	0.01
1.01	%	FUS-ICP	3 16	5 49	4 61	9 3 3	14 92	8.86	4 18	3.76
Total	%	FUS-ICP	98.01	97.73	98.62	98.86	98.07	97.97	100.6	100.4
Ba	ppm	FUS-ICP	2778	2538	2461	1664	15	858	1037	933
Sr	ppm	FUS-ICP	8	6	8	10	21	3	5	5
Y	ppm	FUS-ICP	44	29	47	49	56	36	28	43
Sc	ppm	FUS-ICP	13	10	13	16	23	10	10	11
Zr	ppm	FUS-ICP	154	117	166	174	173	120	130	132
Be	ppm	FUS-ICP	1	< 1	1	1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	< 5	6	5	< 5	11	5	< 5	< 5
Hg	ppb	FIMS	481	1460	1110	551	60	31	18	9
Cr	ppm	HF/HNO3	<8.9	8.94	<8.9	<8.9	<8.9	<8.9	11.06	<8.9
Со	ppm	HF/HNO3	<0.9	<0.9	<0.9	0.99	24.01	<0.9	1.78	<0.9
Nı	ppm	HF/HNO3	<10	13.76	<10	<10	<10	<10	<10	<10
Cu	ppm	HF/HNO3	269.22	69.57	120.75	137.50	154.03	3709.45	553.25	640.67
Zn	ppm	HF/HNO3	3140.30	8869.23	5/06.9/	13/6.58	2.50	302.94	129.46	521.15
AS So	ppm	HE/HNO3	<26.7	<26.7	<26.7	<26.7	<267	<26.7	<26.7	<267
Br	ppm	HF/HNO3	233.63	237 31	20.7	201.31	214.99	254 75	275 79	229.06
Mo	ppm	HF/HNO3	255.05	3 22	3 17	6.69	5 15	39.47	4 15	2 59
Ag	ppm	HF/HNO3	2.95	1.62	<1	1.02	<1	<1	<1	1 44
Cd	ppm	HF/HNO3	11.26	31.42	23.58	5.14	9.55	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	12.68	5.73	5.60	50.67	5.62	9.98	10.48	8.93
Sb	ppm	HF/HNO3	6.55	8.06	6.74	28.65	94.47	2.63	2.45	1.54
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	18.88
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	3.85	4.81	5.78	2.26	14.36	3.93	3.53	5.23
Pb	ppm	HF/HNO3	3140.32	303.07	34.61	109.40	165.01	47.75	37.17	<25.3
Bi	ppm	HF/HNO3	0.21	1.44	0.33	19.98	2.26	11.78	4.98	26.09
Nb	ppm	Na2O2 Sinter	7.92	8.09	9.38	10.46	9.48	5.06	4.96	5.44
La	ppm	Na2O2 Sinter	18.38	17.52	21.02	21.10	30.44	22.70	15.88	16.63
Ce	ppm	Na2O2 Sinter	38.12	35.94	43.92	44.84	65.07	47.94	33.46	35.41
Pr	ppm	Na2O2 Sinter	4.78	4.62	5.51	5.64	8.25	6.12	4.27	4.48
Nd	ppm	Na2O2 Sinter	20.36	19.95	23.83	24.77	36.12	26.68	18.14	19.25
Sm	nnm	Na2O2 Sinter	4 84	4 55	5 72	5.92	8 69	6.24	4 37	5.08
En	ppm	Na2O2 Sinter	1.32	1.08	1 38	1 13	1.20	1.30	0.85	0.95
Th	ppm	Na2O2 Sinter	0.00	0.70	1.50	1.15	1.20	1.00	0.85	1.11
10	ppm	Na2O2 Sinter	0.99	0.79	1.19	1.19	1.05	1.02	0.74	1.11
Dy	ppm	Na2O2 Sinter	6.75	5.19	/.90	8.08	10.73	6.48	4.94	1.33
Но	ppm	Na2O2 Sinter	1.56	1.15	1.74	1.82	2.26	1.38	1.08	1.58
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.82	3.57	5.31	5.81	6.65	4.10	3.27	4.75
Tm	ppm	Na2O2 Sinter	0.76	0.56	0.85	0.90	0.99	0.63	0.52	0.71
Yb	ppm	Na2O2 Sinter	5.21	3.77	5.62	6.10	6.48	4.18	3.68	4.70
Lu	ppm	Na2O2 Sinter	0.80	0.57	0.87	0.96	1.03	0.63	0.59	0.72
Hf	ppm	Na2O2 Sinter	5.75	3.82	5.03	5.54	6.51	3.65	3.27	3.27
Та	ppm	Na2O2 Sinter	0.32	0.24	0.30	0.35	0.33	0.20	0.21	0.22
Th	ppm	Na2O2 Sinter	6.31	4.99	6.63	6.87	7.35	4.22	4.43	5.47
AIOH WL	nm	Terraspec	2,210	2,210	2,209	2,209	#DIV/0!	2,210	2,210	2,208
AlOH Depth	%	Terraspec	9.74%	7.17%	10.30%	10.94%	#DIV/0!	10.91%	12.65%	17.90%
FeOH WL	nm	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	2,250	2,252	#DIV/0!	#DIV/0!
FeOH Depth	%	Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	8.54%	2.22%	#DIV/0!	#DIV/0!
AlOH depth/FeOH depth		Terraspec	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	#DIV/0!	4.91	#DIV/0!	#DIV/0!
Abbreviations: Units		FT= Footwall tuff		Alteration		Other W/ = Wayslangth				
	HBX	= Hanging wall bree	cia	FFL= Footwal	l flow	Chl-Ser= Chlor	ite-sericite			
	FLT=	Footwall lapilli tuff	f	INT= Intrusion	1	Chl=Chlorite				

		TM -
Annondiv R. Tahla R1 1	Whole_rock (Leochemistr	v and Tarracnae''' Nata
Appendix D. Lavie D1.1		y and i criaspec Data

Аррения	<b>D</b> . 1		51 VVIIU	JIC-I UCK	GUULI	icinisti	y anu i	l ci i asp	L D	ala
Sample ID			H501340	H501341	H501342	H501343	H501344	H501345	H501346	H501347
Hole ID			BD10-18	BD10-18	BD11-190	BD11-190	BD11-190	BD11-189	BD11-189	BD11-189
Depth (m)			27.2	31.3	9.8	24.9	39.3	11.3	19.5	20.7
Lithology			FFL	FBX	FLT	FLT	FLT	FLT	FFL	FLT
Alteration			Otz-Ser	Chl	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser
SiO	0/2	EUS ICD	75.27	63.27	76.25	69.47	72 57	71.07	61.57	64.03
5102	/0	FUS-ICF	0.05	11.00	11.05	10.04	12.57	/1.//	10.02	10.11
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	9.95	11.09	11.95	10.94	11.24	11.4	10.92	18.11
$Fe_2O_3(T)$	%	FUS-ICP	5.48	7.97	3.63	7.9	4.77	6.75	10.99	4.46
MnO	%	FUS-ICP	0.026	0.254	0.039	0.051	0.113	0.105	0.339	0.092
MgO	%	FUS-ICP	1.02	9.13	1.54	1.13	2.7	2.34	3.98	2.22
CaO	%	FUS-ICP	0.04	0.07	0.05	0.04	0.07	0.05	0.1	0.08
Na <sub>2</sub> O	%	FUS-ICP	0.09	0.03	0.16	0.17	0.16	0.17	0.18	0.31
K O	0/	FUS ICD	2.02	0.05	2	2.97	2.51	2.61	2.24	1.96
<b>K</b> <sub>2</sub> <b>O</b>	70	rus-icr	2.82	0.9	5	2.87	2.31	2.01	2.24	4.80
TiO <sub>2</sub>	%	FUS-ICP	0.146	0.158	0.17	0.155	0.179	0.168	0.226	0.256
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	0.04	0.02	< 0.01	0.03	0.03	0.02	0.04
LOI	%	FUS-ICP	4.27	6.8	3.45	5.9	4.05	4.65	8.4	5.26
Total	%	FUS-ICP	99.12	99.71	100.3	98.64	98.39	100.2	98.96	99.73
Ba	ppm	FUS-ICP	793	270	1262	1009	992	1119	917	2046
Sr	ppm	FUS-ICP	6	2	9	9	9	7	10	16
Y	nnm	FUS-ICP	34	41	43	40	37	41	37	60
Sc	nnm	FUS-ICP	10	10	12	11	11	12	12	19
Zr	ppm	FUS-ICP	129	148	144	141	141	147	175	240
Pa	ppm	FUS ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	1
De V	ppm	FUS-ICP	< 1	- 1	< 1 6	< 1	0	< 1	< 1	1
v H-	ppm	FUS-ICF	< 3	3	0	< 3	9	< 3	0	< 3
Hg	рро	FIMS	11	24	< 5	12	8	< 5	115	8
Cr	ppm	HF/HNO3	9.69	<8.9	<8.9	15.30	15.70	34.71	36.99	26.36
Co	ppm	HF/HNO3	<0.9	<0.9	1.24	4.8/	1.05	6.21	20.61	1./1
Ni	ppm	HF/HNO3	<10	13.17	<10	<10	<10	27.63	20.04	15.20
Cu	ppm	HF/HNO3	234.51	287.67	963.89	784.11	381.03	20.48	1279.01	<19
Zn	ppm	HF/HNO3	120.82	416.08	86.39	167.04	77.22	136.43	172.84	97.96
As	ppm	HF/HNO3	96.88	30.38	23.31	14.48	2.80	16.77	21.19	3.83
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	262.65	252.70	249.99	195.49	222.96	215.16	203.30	213.63
Mo	ppm	HF/HNO3	2.98	4.46	3.41	5.98	5.52	11.33	16.27	9.31
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Cď	ppm	HF/HNO3	< 0.9	< 0.9	<0.9	<0.9	< 0.9	<0.9	< 0.9	< 0.9
Sn	ppm	HF/HNO3	9 78	7 87	11 41	5 26	4 30	5.17	4 18	4 36
Sh	nnm	HE/HNO3	2.69	4 57	1.25	3.83	1.27	1 34	6.14	1.70
Te	ppm	HE/HNO3	<13	<13	<13	<13	<13	<13	<13	<13
ie I	ppm	HE/INO2	<4.3	<4.5	<4.5	<4.5	<4.3	<4.5	~4.5	<4.5
1 W	ppm	HF/HNO3	6.12	2.24	2.66	< <u>5</u> 7	2.00	2 70	<37 14.07	<37 6.17
W	ppm	HF/HNO3	0.13	2.34	3.00	0.00	3.90	3.70	14.07	0.17
Pb	ppm	HF/HNO3	31.06	<25.3	<25.3	<25.3	<25.3	<25.3	28.66	<25.3
Bi	ppm	HF/HNO3	6.89	1.90	1.39	7.70	0.82	1.05	7.25	0.31
Nb	ppm	Na2O2 Sinter	4.92	6.59	5.77	7.46	8.33	6.53	9.03	9.86
La	ppm	Na2O2 Sinter	17.26	17.74	19.23	16.86	17.26	18.14	16.49	31.65
Ce	nnm	Na2O2 Sinter	36.49	37.20	39.95	35 55	35.88	37 77	31.42	64 64
D.	ppm	No2O2 Sinter	4 (7	1.72	5 10	4.51	1.54	4.00	2.92	8.00
Pf	ppm	Na2O2 Sinter	4.67	4.72	5.10	4.51	4.54	4.69	3.82	8.00
Nd	ppm	Na2O2 Sinter	19.47	20.42	22.03	19.27	19.11	20.53	16.01	35.59
Sm	ppm	Na2O2 Sinter	4.66	5.03	5.37	4.75	4.73	5.24	4.29	9.23
Eu	ppm	Na2O2 Sinter	0.90	1.41	0.89	0.91	0.77	1.03	0.73	1.65
Th	nnm	Na2O2 Sinter	0.82	1.08	1.06	1.00	0.93	1.00	0.80	1.52
10	ppm	N-2O2 Sinter	0.82	1.08	1.00	1.00	0.93	1.00	0.89	1.52
Dy	ppm	Na2O2 Sinter	5.45	7.30	7.17	6.76	6.34	6.78	6.20	9.71
Но	ppm	Na2O2 Sinter	1.21	1.60	1.55	1.42	1.33	1.42	1.36	2.03
Er	ppm	Na2O2 Sinter	3.66	4.87	4.77	4.28	4.16	4.35	4.16	6.07
Tm	nnm	Na2O2 Sinter	0.57	0.74	0.73	0.66	0.60	0.67	0.66	0.95
1111	ppm	No2O2 Sinter	0.57	0.74	0.75	0.00	0.00	0.07	0.00	0.75
Yb	ppm	Na2O2 Sinter	3.95	4.97	4.84	4.36	4.13	4.37	4.84	6.71
Lu	ppm	Na2O2 Sinter	0.62	0.77	0.76	0.68	0.67	0.75	0.82	1.08
Hf	ppm	Na2O2 Sinter	2.76	4.36	4.22	3.53	4.59	4.08	5.49	6.36
Та	ppm	Na2O2 Sinter	0.21	0.25	0.25	0.25	0.28	0.28	0.30	0.38
 TL	PPm	Na2O2 Sintar	4 41	5.20	5.10	5.00	£ 10	4.02	6.50	0.00
	ppm	T	4.41	5.27	5.12	5.00	5.19	4.93	0.07	8.03
AIOH WL	nm	1 erraspec	2,207	2,204	2,203	2,203	2,201	2,202	2,198	2,202
AIOH Depth	%	Terraspec	24.18%	5.25%	16.55%	14.80%	18.78%	14.13%	2.96%	12.65%
FeOH WL	nm	Terraspec	#DIV/0!	2,251	2,255	#DIV/0!	2,254	2,255	2,256	2,254
FeOH Depth	%	Terraspec	#DIV/0!	10.31%	3.09%	#DIV/0!	4.11%	3.37%	1.19%	1.55%
AlOH depth/FeOH depth		Terraspec	#DIV/0!	0.51	5.36	#DIV/0!	4.57	4.19	2.49	8.15
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	Hanging wall flor	w	FBX= Footwa	all breccia	Qtz-Ser= Qua	rtz-sericite		WL= Waveler	ngth
	HBX	= Hanging wall bre	ccia	FFL= Footwa	ll flow	Chl-Ser= Chlo	orite-sericite			

HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow Chl=Chlorite INT= Intrusion

Аррения	<b>D</b> . 1		51 10 110	IC-IUCK	ututi	iennsei,	y anu i	l ci i asp	tt D	ata
Sample ID			H501348	H501349	H501350	H501351	H501352	H501353	H501354	H501355
Hole ID			BD11-189	BD11-189	BD10-106	BD10-106	BD10-106	BD10-106	BD10-106	BD10-106
Depth (m)			23.7	38.2	9.6	13.9	17.1	18.4	21.8	29.4
Lithology			INT	INT	FLT	FBX	FT	FLT	FBX	FLT
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser
SiO <sub>2</sub>	%	FUS-ICP	71.29	77.44	73.49	70.53	71.62	72.04	72.58	67.6
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	10.66	10.55	10.26	11.22	12	11.29	11.96	12.4
$Fe_2O_2(T)$	%	FUS-ICP	6.48	3.44	4.13	3.18	4.86	4.54	4.38	5.89
MnO	%	FUS-ICP	0.171	0.049	0.136	0.167	0.06	0.118	0.064	0 195
MgO	%	FUS-ICP	2.87	1 37	4 76	6.93	2.09	2.88	1.95	4 91
CaO	%	FUS-ICP	0.06	0.04	0.1	0.43	0.06	0.14	0.05	0.07
Na <sub>2</sub> O	%	FUS-ICP	0.15	0.17	0.16	0.14	0.23	0.19	0.22	0.23
K O	0/.	FUS ICP	2.24	2.64	2.04	1.79	2.08	2.61	3.02	2.64
K <sub>2</sub> O	/0	FUS ICP	2.24	2.04	2.04	1.78	5.08	2.01	5.02	2.04
1102	%0	FUS-ICP	0.154	0.145	0.183	0.167	0.187	0.18	0.197	0.208
$P_2O_5$	%	FUS-ICP	0.02	0.01	0.05	0.03	0.03	0.03	0.04	0.02
LOI	%	FUS-ICP	4.68	2.67	4.92	5.28	4.72	4.54	4.34	6.59
Total	%	FUS-ICP	98.79	98.55	100.2	99.85	98.95	98.55	98.8	100.7
Ba	ppm	FUS-ICP	913	1309	846	778	1176	1183	1113	939
Sr	ppm	FUS-ICP	8	7	8	16	10	9	10	13
Y	ppm	FUS-ICP	43	34	29	42	40	43	45	42
Sc	ppm	FUS-ICP	12	11	11	12	12	11	12	13
Zr	ppm	FUS-ICP	152	133	121	151	145	147	156	167
Be	ppm	FUS-ICP	< 1	< 1	< 1	< 1	< 1	< 1	< 1	< 1
V	ppm	FUS-ICP	< 5	< 5	21	/	10	9	9	12
Hg	ррв	FIMS	5	< 5	22,80	20.16	15	20.02	/	< 5
Cr Cr	ppm	HF/HNO3	27.13	20.49	35.89	29.16	< 8.9	30.93 <0.0	< 8.9	< 8.9
CO NI	ppm	HE/INO2	5.26	1.01	21.41	1.34	<10	<0.9	1.38	<0.9
NI Gu	ppm	HE/INO2	13.12	<10	21.41	20.92	224.78	1860.28	54.62	<10
Cu Zn	ppm	HE/INO2	20.50	100.86	142.78	36.79	524.76 80.55	75 42	56.42	110.06
	ppm	HE/HNO3	4 70	100.80	2.00	1 9 2	80.33	/ 5.42	14 45	5.08
AS So	ppin	HE/HNO3	4.70	<1.0	2.09	26.7	26.7	4.09	-26.7	267
Br.	ppin	HE/HNO3	201.08	215.01	~20.7	~20.7	~20.7	242.7	~20.7	217.40
Bi Mo	ppin	HE/HNO3	204.08	7 49	11.47	214.05	4 70	242.41	5 86	217.40
A.a.	ppin	HE/HNO3	/.90	/.40	<1	/.94	4.70	0.70	5.80	4.01
Ag Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	3 78	6.48	2.86	2 53	3 79	3.69	3 35	3.64
Sh	nnm	HF/HNO3	1.63	0.74	1.06	0.78	2 50	1 23	1.61	0.88
Te	nnm	HF/HNO3	<4.3	<43	<4 3	<43	<4 3	<4.3	<4.3	<4.3
I	nnm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	nnm	HF/HNO3	4 78	4 60	6.87	6.48	2.82	3.63	3 36	2 21
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HF/HNO3	0.82	0.77	0.92	0.33	3.06	4 06	2.64	0.51
Nh	nnm	Na2O2 Sinter	6.61	5 94	5.14	6.86	8 4 8	9.26	8.48	10.08
Ind	ppm	Na2O2 Sinter	16.20	15.06	16 47	10.17	17.79	15.00	21.26	20.55
La	ppm	Na2O2 Sinter	16.20	13.90	10.47	19.17	17.78	13.99	21.20	20.33
Ce	ppm	Na2O2 Sinter	33.31	33.10	33.04	40.23	38.41	35.32	42.70	43.31
Pr	ppm	Na2O2 Sinter	4.17	4.20	4.00	4.95	4.71	4.45	5.06	5.36
Nd	ppm	Na2O2 Sinter	18.38	17.95	17.44	21.31	20.01	18.92	20.50	23.26
Sm	ppm	Na2O2 Sinter	4.91	4.46	4.40	5.15	4.64	4.59	4.77	5.49
Eu	ppm	Na2O2 Sinter	0.76	0.68	0.78	0.93	0.90	0.82	0.87	1.00
Th	ppm	Na2O2 Sinter	1.02	0.84	0.81	1.01	0.91	1 04	1.04	1.05
Dv	nnm	Na2O2 Sinter	6.95	5 72	5.26	6.68	6.14	7 27	7.16	7.04
Цо	ppin	Na2O2 Sinter	1.49	1.21	1.00	1.47	1.20	1.54	1.55	1.57
110	ppm	No O Sintor	1.40	1.21	1.09	1.47	1.29	1.54	1.55	1.57
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	4.47	3.65	3.32	4.53	4.13	4.56	4.66	4.90
Tm	ppm	Na2O2 Sinter	0.67	0.57	0.51	0.70	0.62	0.70	0.70	0.74
Yb	ppm	Na2O2 Sinter	4.62	3.71	3.54	4.66	4.01	4.67	4.73	5.05
Lu	ppm	Na2O2 Sinter	0.76	0.65	0.59	0.73	0.64	0.73	0.71	0.77
Hf	ppm	Na2O2 Sinter	4.26	4.15	3.83	4.99	4.08	5.10	4.66	5.66
Та	ppm	Na2O2 Sinter	0.26	0.25	0.21	0.26	0.31	0.35	0.32	0.36
Th	npm	Na2O2 Sinter	4 00	4.62	3.07	5.20	6 20	6.28	6.12	7.08
AIOH WI	ppm	Terraspec	7.20 2.200	2 202	2 202	2 201	2 205	2 204	2 204	2 200
AlOH Denth	%	Terraspec	10.83%	2,202	12 78%	11 52%	13 30%	10.68%	15 98%	12 73%
FeOH WL	nm	Terraspec	2 254	21.0070	2 252	2 251	2 252	2 252	2 254	2 253
FeOH Denth	%	Terraspec	4 24%	3 56%	4 80%	6 31%	1 48%	2,20%	1 16%	3 90%
AlOH denth/FeOH denth	/0	Terraspec	2.55	6.12	2 66	1.83	8 99	4 85	13 73	3 26
		- erraspee	2.00	0.12	2.00	1.05	0.77			5.40
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	Hanging wall flo	w	FBX= Footwa	ll breccia	Qtz-Ser= Qua	rtz-sericite		WL= Waveler	ngth
	HFL= Hanging wall flow HBX= Hanging wall breecia					Qtz-Ser= Quartz-sericite WL= Wavelength				-

Chl=Chlorite

FLT= Footwall lapilli tuff

Annendix B. Table B1 1	Whole-rock Geochemistr	v and Terraspec <sup>TM</sup> Data
Appendix $D_1$ i abie $D_{1,1}$		y and i criaspec Data

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Appendix	ndix B: Table B1.1 Whole-rock Geochemistry and Terraspec <sup>TM</sup> Data									ata	
Sample ID			H501356	H501357	H501358	H501359	H501360	H501361	H501363	H501364	
Hole ID			BD10-106	BD10-106	BD10-106	BD10-106	BD10-106	BD11-129	BD10-9	BG10-7	
Depth (m)			36.9	40.5	44.3	51.2	68.5	28.3	10.1	6.5	
Lithology			FLT	FT	FLT	FBX	FFL	FFL	HBX	HFL	
Alteration			Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	
SiO <sub>2</sub>	%	FUS-ICP	71.34	68.77	75.54	77.33	77.81	74.77	67.44	80.41	
$Al_2O_3$	%	FUS-ICP	12.45	12.74	11.59	10.91	11.16	8.54	14.24	8.8	
$Fe_2O_3(T)$	%	FUS-ICP	3.58	4.68	3.8	1.49	2.6	6.9	4.94	2.53	
MnO	%	FUS-ICP	0.134	0.121	0.068	0.095	0.108	0.006	0.197	0.017	
MgO	%	FUS-ICP	2.96	4.26	2.19	2.21	1.59	0.27	3.22	1.29	
CaO	%	FUS-ICP	0.07	0.09	0.05	0.05	0.06	0.02	0.04	0.03	
Na <sub>2</sub> O	%	FUS-ICP	0.24	0.24	0.21	0.19	0.21	0.18	0.11	0.12	
K <sub>2</sub> O	%	FUS-ICP	3.08	2.89	2.88	2.76	3.14	2.33	3.43	2.24	
11O <sub>2</sub>	%	FUS-ICP	0.204	0.294	0.196	0.17	0.182	0.114	0.191	0.118	
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	0.03	0.05	0.03	0.03	0.03	< 0.01	0.03	0.02	
LOI	%	FUS-ICP	4.37	5.59	4.16	3.25	3.87	5.02	4.19	2.61	
l otal De	%	FUS-ICP	98.48	99.72	100.7	98.49	100.7	98.16	98.04	98.19	
Da Sr	ppm	FUS-ICP	1156	1025	1084	1085	1043	7	6	1343	
Y	ppm	FUS-ICP	44	43	41	34	40	30	49	28	
Sc	ppm	FUS-ICP	13	16	13	12	12	9	15	8	
Zr	ppm	FUS-ICP	166	134	142	137	146	104	184	110	
Be	ppm	FUS-ICP	< 1	1	1	1	1	< 1	2	< 1	
V	ppm	FUS-ICP	< 5	55	18	16	6	6	7	7	
Hg	ppb	FIMS	< 5	10	< 5	< 5	11	56	198	8	
Cr	ppm	HF/HNO3	34.20	61.58	20.42	<8.9	<8.9	18.56	<8.9	13.60	
Co	ppm	HF/HNO3	< 0.9	3.78	1.31	<0.9	1.85	9.48	< 0.9	1.91	
N1 Cu	ppm	HF/HNO3	17.32	29.54	<10	<10	<10	<10	<10	<10	
Cu Zn	ppm	HE/HNO3	156.15	21.64	×19 88.06	29.14	55.10	34.30	000.28	51.81	
As	ppm	HF/HNO3	2 24	5.05	2 37	<1.6	2 92	12 74	1.88	2 61	
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	
Br	ppm	HF/HNO3	238.58	189.06	237.68	238.81	239.32	224.09	216.37	284.03	
Mo	ppm	HF/HNO3	9.55	9.32	3.46	3.52	12.12	7.85	2.28	3.90	
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1	
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	<0.9	<0.9	2.46	<0.9	
Sn	ppm	HF/HNO3	3.81	3.02	5.20	4.12	2.98	3.41	6.70	3.51	
Sb T-	ppm	HF/HNO3	0.85	2.00	1.35	1.33	1.04	1.87	2.01	1.84	
Ie	ppm	HF/HNO3	<4.3	<4.3	<4.5	<4.3	<4.3	<4.3	<4.5	<4.3	
W	ppm	HF/HNO3	4 59	8.08	7 37	616	4 20	3.12	4 09	1.68	
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	493.51	<25.3	
Bi	ppm	HF/HNO3	0.75	1.05	0.43	< 0.1	2.41	8.06	0.41	1.78	
Nb	ppm	Na2O2 Sinter	7.68	5.94	8.21	8.30	7.84	4.90	10.92	6.30	
La	ppm	Na2O2 Sinter	19.28	18.37	16.09	19.57	18.52	13.07	22.99	14.68	
Ce	ppm	Na2O2 Sinter	40 92	39 37	34 96	41.57	38 70	26 79	48 42	30.97	
Pr	ppm	Na2O2 Sinter	5.11	5 16	4 44	5.17	4 93	3 38	6.14	3 91	
Nd	nnm	Na2O2 Sinter	22.54	22.72	18 74	21.25	20.92	14 36	26.15	16.97	
Sm	ppm	Na2O2 Sinter	5.68	5.82	4 53	5.05	4 88	3 51	6.24	4 13	
5m En	ppm	Na2O2 Sinter	0.03	0.88	0.85	0.82	4.00	0.84	1.07	4.15	
Th	ppm	Na2O2 Sinter	1.05	1.02	0.05	0.82	0.75	0.69	1.97	0.78	
10 Dec	ppm	Na2O2 Sinter	7.15	1.02	0.90	0.82	0.97	0.08	0.24	0.73	
Dy	ppm	Na2O2 Sinter	7.15	0.82	0.37	5.15	0.01	4.49	8.34	5.05	
Но	ppm	Na2O2 Sinter	1.59	1.47	1.39	1.12	1.36	0.97	1./8	1.08	
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	5.06	4.57	4.13	3.59	4.11	2.76	5.49	3.21	
Tm	ppm	Na2O2 Sinter	0.75	0.68	0.62	0.56	0.62	0.43	0.81	0.47	
Yb	ppm	Na2O2 Sinter	5.22	4.43	4.28	4.04	4.29	2.92	5.61	3.38	
Lu	ppm	Na2O2 Sinter	0.82	0.69	0.60	0.65	0.66	0.45	0.94	0.50	
Hf	ppm	Na2O2 Sinter	5.22	4.03	4.60	5.63	5.69	3.66	6.36	3.44	
Та	ppm	Na2O2 Sinter	0.29	0.24	0.31	0.33	0.34	0.20	0.41	0.25	
Th	ppm	Na2O2 Sinter	5.76	4.69	5.53	5.71	6.46	4.80	8.39	5.02	
AlOH WL	nm	Terraspec	2,204	2,201	2,202	2,204	2,205	2,203	2,207	2,205	
AlOH Depth	%	Terraspec	9.87%	11.04%	12.50%	13.58%	9.90%	4.12%	7.55%	11.05%	
FeOH WL	nm	Terraspec	2,252	2,252	2,252	2,253	#DIV/0!	#DIV/0!	2,254	2,251	
FeOH Depth	%	Terraspec	1.31%	3.45%	1.87%	1.33%	#DIV/0!	#DIV/0!	2.83%	2.04%	
AIOH depth/FeOH depth		1 erraspec	7.53	3.20	6.68	10.25	#DIV/0!	#DIV/0!	2.66	5.43	
Abbreviations:	bbreviations: Units HFL= Hanging wall flow HBX= Hanging wall brave			FT= Footwall FBX= Footwa FFL= Footwa	tuff all breccia Il flow	Alteration Qtz-Ser= Qua Chl-Ser= Chlc	rtz-sericite prite-sericite		<b>Other</b> WL= Wavelength		

HBX= Hanging wall how HBX= Hanging wall breccia FLT= Footwall lapilli tuff

FFL= Footwall flow Chl=Chlorite INT= Intrusion

Аррениіх	<b>D</b> .	able D1.	I WINU	DIE-IOCK	Geoci	iennsu y	anu .	i ci i asp	L D	ala
Sample ID			H501365	H501366	H501367	H501368	H501369	H501370	H501371	H501372
Hole ID			BG10-7	BG10-7	BG10-7	BG10-7	BG10-7	BD11-183	BD11-183	BD11-183
Depth (m)			18.7	31.9	47.2	55.2	63	8.2	21.1	34.3
Lithology			FLT	FLT	FLT	FBX	FLT	HFL	HBX	HFL
Alteration			Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	%	FUS-ICP	70.74	63.81	74.74	68.63	62.27	77.92	61.63	74.63
Al <sub>2</sub> O <sub>3</sub>	%	FUS-ICP	14.24	11.45	11.31	12.99	13.76	7	15.55	11.19
$Fe_2O_3(T)$	%	FUS-ICP	3.46	9.88	5.69	4.91	7.24	7.13	8.43	4.56
MnO	%	FUS-ICP	0.026	0.041	0.02	0.148	0.263	0.018	0.08	0.059
MgO	%	FUS-ICP	1.65	2.29	0.94	3.38	4.73	0.37	2.05	1.47
CaO	%	FUS-ICP	0.04	0.05	0.04	0.07	0.07	0.02	0.07	0.05
Na <sub>2</sub> O	%	FUS-ICP	0.2	0.15	0.16	0.16	0.15	0.11	0.23	0.16
K <sub>2</sub> O	%	FUS-ICP	3.75	2.75	3.2	3.11	2.98	1.99	4.1	2.93
TiO <sub>2</sub>	%	FUS-ICP	0.209	0.179	0.153	0.195	0.21	0.088	0.234	0.135
P2O5	%	FUS-ICP	0.03	0.02	0.02	0.03	0.05	0.01	0.04	0.01
LOI	%	FUS-ICP	3.92	7.22	4.51	4.84	6.71	4.64	6.22	3.89
Total	%	FUS-ICP	98.27	97.83	100.8	98.48	98.42	99.29	98.65	99.09
Ba	ppm	FUS-ICP	1940	1226	1137	1044	957	1341	1904	1202
Sr	ppm	FUS-ICP	10	7	8	8	8	4	9	6
Y	ppm	FUS-ICP	48	40	37	46	52	33	54	40
Sc	ppm	FUS-ICP	15	13	11	13	14	5	15	9
Zr	ppm	FUS-ICP	177	134	137	167	165	88	196	146
Be	ppm	FUS-ICP	1	< 1	< 1	1	1	< 1	1	< 1
V	ppm	FUS-ICP	9	15	6	< 5	< 5	6	5	8
Hg	ppb	FIMS	13	100	12	< 5	14	6	10	< 5
Cr Co	ppm	HF/HNO3	< 8.9	< 8.9	10.74	20.30	< 8.9	<8.9	35.80	< 8.9
Ni	ppm	HE/HNO3	<10	8.55 <10	<0.9	<0.9 15.24	<10	<10	14.01	5.71
Cu	ppm	HF/HNO3	<10	45.67	<10	923 75	595 75	<10	22.51	<10
Zn	nnm	HF/HNO3	42 44	49.03	25.94	147 75	337.96	32.16	144 77	53.97
As	ppm	HF/HNO3	8.33	120.65	13.84	17.12	68.63	25.65	15.59	5.49
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	235.74	228.26	230.56	201.20	245.48	216.07	216.01	221.52
Mo	ppm	HF/HNO3	3.80	5.71	6.59	7.61	3.44	8.07	60.91	9.19
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	<0.9	<0.9	< 0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	3.84	3.68	3.83	4.09	8.81	4.50	4.69	4.22
Sb	ppm	HF/HNO3	3.16	6.98	2.45	1.35	17.11	1.84	2.65	1.51
Te	ppm	HF/HNO3	4.28	5.05	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
I	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	5.54 -25.2	3.70	4.35	2.81	1.80	1.57	12.96	3.76
PD Di	ppm	HF/HNO3	<25.5	<25.5	<25.5 2.82	<25.5	<25.5	<25.5	<23.3	<25.5
DI NIL	ppm	No2O2 Sinter	4.09	7.55	2.02	5.51	2.54	2.97	2.09	1.14
ND	ppm	Na2O2 Sinter	9.38	8.70	/.38	/.10	7.05	4.84	9.55	7.62
La	ppm	Na2O2 Sinter	22.36	18.41	15.99	17.86	19.37	9.66	21.96	17.68
Ce	ppm	Na2O2 Sinter	46.68	38.52	32.96	37.08	40.90	20.22	45.92	36.75
Pr	ppm	Na2O2 Sinter	5.90	4.94	4.17	4.62	5.15	2.59	5.85	4.64
Nd	ppm	Na2O2 Sinter	25.86	20.98	17.85	19.60	22.50	11.48	24.65	19.92
Sm	ppm	Na2O2 Sinter	6.12	5.15	4.34	5.14	5.49	3.19	6.14	4.86
Eu	ppm	Na2O2 Sinter	1.64	1.23	1.80	1.08	1.17	0.53	0.97	0.73
Tb	ppm	Na2O2 Sinter	1.23	1.00	0.88	1.06	1.25	0.83	1.24	0.96
Dy	ppm	Na2O2 Sinter	8.11	6.62	5.93	7.10	8.57	5.42	8.49	6.25
Но	ppm	Na2O2 Sinter	1.78	1.45	1.29	1.59	1.84	1.13	1.89	1.35
Er	ppm	Na <sub>2</sub> O <sub>2</sub> Sinter	5 30	4 45	4 08	4 85	5 69	3 35	5 71	4 23
Tm	ppm	Na2O2 Sinter	0.81	0.66	0.59	0.74	0.86	0.49	0.84	0.63
Vh	ppin	Na2O2 Sinter	5.28	4.57	4.22	4.02	5.01	2.26	5.80	4.32
IU Lu	ppm	Na2O2 Sinter	0.92	4.37	4.23	4.92	0.01	0.50	0.00	4.32
Lu	ppm	Na2O2 Sinter	0.85	0.69	0.65	0.84	0.91	0.50	0.90	0.68
Ht	ppm	Na2O2 Sinter	5.37	4.69	3.63	4.21	4.36	2.99	5.53	4.58
Ta	ppm	Na2O2 Sinter	0.35	0.29	0.29	0.27	0.27	0.17	0.32	0.32
Th	ppm	Na2O2 Sinter	7.71	6.41	5.63	5.50	5.64	4.08	6.54	6.13
AIOH WL	nm	Terraspec	2,203	2,203	2,205	2,204	2,203	2,205	2,203	2,203
AIOH Depth	%	Terraspec	11.83%	10.64%	14.08%	12.30%	13.08%	17.50%	6.48%	19.83%
FeOH WL	nm %	Torraspec	2,255	2,254	#DIV/0!	2,254	2,252	#DIV/0!	2,200	2,254
AlOH denth/FaOH donth	70	Terraspec	1.10%	1.20%	#DIV/0! #DIV/0!	7 10	2.54%	#DIV/0!	1.54%	2.33%
AIOTI deput/reOri deptil		renaspec	10.73	0.00	#D1V/0!	1.19	5.57	#D1V/0!	4.00	1.10
Abbreviations:	Units			FT= Footwall	tuff	Alteration			Other	
	HFL=	Hanging wall flow	7	FBX= Footwa	ll breccia	Qtz-Ser= Quart	z-sericite		WL= Waveler	ngth
	HBX= Hangi		cia	FFL= Footwal	l flow	Chl-Ser= Chlor	ite-sericite			
	FLT= Footwall lapilli tuff		f	INT= Intrusion	n	Chl=Chlorite				

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A	XX/L - L L	Cool	J T	
Appendix B: Table B1.1	wnole-rock	Geocnemistry	and Terraspec	Data

Аррспиіх	<b>D</b> .	able DI	51 WHU	1C-1 UCK	Geoti	iennsu y	anu i	errasp	et Data
Sample ID			H501373	H501374	H501375	H501376	H501377	H501378	H501379
Hole ID			BD11-183	BD11-183	BG10-2	BG10-2	BG10-2	BG10-2	BG10-2
Depth (m)			52	68	10.1	20.4	27.6	31.9	44.3
Lithology			HFL	HFL	HFL	HFL	FLT	FBX	FLT
Alteration			Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiO <sub>2</sub>	%	FUS-ICP	78.91	58.3	77.05	75.8	68.11	76.29	71.48
Al-O	0/0	FUS-ICP	937	22.62	11 14	11.88	11.55	10.56	12 39
Fa O (T)	0/	FUS ICD	116	2 62	2 00	2.24	7	2.08	4.40
$Fe_2O_3(1)$	70	FUS-ICP	4.10	5.02	3.88	5.24	0.000	5.08	4.49
MnO	%	FUS-ICP	0.033	0.082	0.091	0.074	0.206	0.151	0.115
MgO	%	FUS-ICP	0.75	2.88	2.06	1.51	2.74	2.12	2.44
CaO	%	FUS-ICP	0.07	0.13	0.04	0.06	1.22	1.08	0.06
Na <sub>2</sub> O	%	FUS-ICP	0.15	0.31	0.14	0.17	0.14	0.13	0.18
K <sub>2</sub> O	%	FUS-ICP	2.58	6.11	2.69	3.23	2.87	2.79	2.84
TiO <sub>2</sub>	%	FUS-ICP	0.145	0.363	0.137	0.15	0.186	0.149	0.171
P <sub>2</sub> O <sub>5</sub>	%	FUS-ICP	< 0.01	0.04	0.02	0.03	0.03	0.02	0.02
LOI	%	FUS-ICP	3.39	5.72	3.19	3.03	6.51	4.2	3.81
Total	%	FUS-ICP	99.56	100.2	100.4	99.17	100.6	100.6	97.99
Ва	ppm	FUS-ICP	1115	2666	951	1120	925	831	1000
Sr	ppm	FUS-ICP	9	13	5	7	24	26	11
Y	ppm	FUS-ICP	30	61	43	42	42	34	46
Sc	ppm	FUS-ICP	10	23	10	11	13	11	13
Zr	ppm	FUS-ICP	117	291	138	147	143	140	164
Be	ppm	FUS-ICP	1	2	< 1	1	< 1	1	1
V	ppm	FUS-ICP	20	18	6	< 5	13	7	8
Hg	ppb	FIMS	6	72	65	9	11	< 5	< 5
Cr	ppm	HF/HNO3	28.93	<8.9	<8.9	27.42	<8.9	11.64	<8.9
Co	ppm	HF/HNO3	3.42	< 0.9	3.75	3.03	12.63	1.46	5.80
Ni	ppm	HF/HNO3	<10	<10	<10	15.87	<10	<10	<10
Cu	ppm	HF/HNO3	<19	<19	280.10	85.66	86.05	<19	<19
Zn	ppm	HF/HNO3	39.04	71.33	430.29	148.83	108.64	114.32	272.57
As	ppm	HF/HNO3	8.52	19.99	2.17	4.85	44.66	3.91	1.95
Se	ppm	HF/HNO3	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7	<26.7
Br	ppm	HF/HNO3	240.46	211.77	229.02	194.50	263.31	190.16	218.00
Mo	ppm	HF/HNO3	8.23	11.69	1.97	11.55	6.24	4.19	3.44
Ag	ppm	HF/HNO3	<1	<1	<1	<1	<1	<1	<1
Cd	ppm	HF/HNO3	<0.9	<0.9	2.01	<0.9	<0.9	<0.9	<0.9
Sn	ppm	HF/HNO3	3.39	4.27	3.50	3.32	12.28	4.16	5.01
Sb	ppm	HF/HNO3	3.66	12.17	2.06	1.57	2.66	1.76	1.28
Te	ppm	HF/HNO3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3	<4.3
Ι	ppm	HF/HNO3	<37	<37	<37	<37	<37	<37	<37
W	ppm	HF/HNO3	6.73	6.58	4.80	3.62	1.64	1.64	2.48
Pb	ppm	HF/HNO3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3	<25.3
Bi	ppm	HF/HNO3	2.05	2.00	0.42	0.38	3.40	0.65	<0.1
Nb	ppm	Na2O2 Sinter	6.19	15.02	7.37	6.42	6.05	8.03	9.05
La	ppm	Na2O2 Sinter	10.36	31.18	17.10	18.63	22.70	16.29	20.50
Ce	ppm	Na2O2 Sinter	21.71	67.10	35.88	38.19	47.78	33.97	42.80
Pr	ppm	Na2O2 Sinter	2.66	8 47	4 49	4 74	6.07	4 22	5 28
Nd	ppm	Na2O2 Sinter	10.98	36 72	18.92	20.70	25.83	17.96	23.01
- Nu	ppm	Na2O2 Sinter	2.71	9.70	10.52	5.12	25.65	17.50	5.52
Sm	ppm	Na2O2 Sinter	2.71	8.70	4.65	5.13	0.55	4.23	5.53
Eu	ppm	Na2O2 Sinter	0.45	2.23	0.88	0.92	1.55	0.98	1.15
Tb	ppm	Na2O2 Sinter	0.61	1.58	0.90	0.95	1.25	0.80	1.11
Dy	ppm	Na2O2 Sinter	4.55	10.40	6.10	6.46	8.11	5.46	7.46
Но	ppm	Na2O2 Sinter	1.03	2.30	1.39	1.43	1.69	1.25	1.70
Er	ppm	Na2O2 Sinter	3.22	6.98	4.39	4.54	4.97	3.93	5.17
Tm	nnm	Na2O2 Sinter	0.49	1.11	0.65	0.69	0.77	0.60	0.79
Yh.	ppin	Na2O2 Sinter	2.42	7.24	4.45	4.99	5.11	4.15	5.20
10	ppm	Na2O2 Sinter	5.42	7.54	4.43	4.00	5.11	4.13	3.29
Lu	ppm	Na2O2 Sinter	0.52	1.12	0.66	0.77	0.79	0.65	0.86
Hf	ppm	Na2O2 Sinter	4.47	11.05	4.27	4.36	4.95	6.20	5.12
Та	ppm	Na2O2 Sinter	0.26	0.56	0.30	0.24	0.26	0.33	0.36
Th	ppm	Na2O2 Sinter	4.95	11.44	5.78	5.15	5.32	5.81	6.66
AlOH WL	nm	Terraspec	2,205	2,205	2,206	2,205	2,206	2,207	2,202
AlOH Depth	%	Terraspec	21.80%	12.10%	5.85%	27.03%	8.55%	15.72%	20.00%
FeOH WL	nm	Terraspec	2,254	#DIV/0!	2,255	2,256	2,253	2,254	2,255
FeOH Depth	%	Terraspec	1.50%	#DIV/0!	2.17%	2.46%	1.36%	2.98%	5.34%
AlOH depth/FeOH depth		Terraspec	14.58	#DIV/0!	2.69	11.00	6.31	5.28	3.75
Abbreviations:	Units	•		FT= Footwall	tuff	Alteration			Other
	HFL=	<ul> <li>Hanging wall flo</li> </ul>	W	FBX= Footwa	ll breccia	Qtz-Ser= Quar	tz-sericite		WL= Wavelength
	HBX	= Hanging wall bre	eccia	FFL= Footwal	l flow	Chl-Ser= Chlor	ite-sericite		

Chl=Chlorite

FLT= Footwall lapilli tuff

Appendix B: Lable BLL	Whole-rock Geochemistry	v and Terraspec Data

		-																		
Πh																				1
Та																			1	0.95
Ηf																		1	0.84	0.84
Νb																	1	0.8	0.94	0.93
Lu																1	0.69	0.77	0.73	0.73
γb															1	66.0	0.7	0.75	0.73	0.74
Tm														1	0.99	86.0	0.7	0.73	0.73	0.73
Но													1	0.98	0.97	0.95	0.66	0.67	0.69	0.69
Dy												1	0.99	0.96	0.94	0.92	0.64	0.63	0.67	0.67
Πb											1	0.99	0.97	0.93	0.91	0.89	0.62	0.59	0.64	0.64
Eu										1	0.69	0.69	0.69	0.67	0.65	0.64	0.47	0.49	0.47	0.48
Sm									1	0.69	0.92	0.88	0.86	0.8	0.77	0.75	0.52	0.5	0.55	0.57
ΡN								1	0.99	0.67	0.89	0.85	0.82	0.77	0.74	0.71	0.51	0.48	0.53	0.57
$\mathbf{Pr}$							1	1	86.0	0.67	0.89	0.85	0.82	0.77	0.74	0.71	0.52	0.48	0.53	0.57
Ce						1	1	1	86.0	0.67	0.89	0.85	0.82	0.77	0.74	0.72	0.53	0.49	0.55	0.58
La					1	1	66.0	66.0	76.0	0.66	0.88	0.84	0.82	0.77	0.75	0.72	0.53	0.5	0.55	0.59
Zr				1	0.55	0.54	0.53	0.53	0.53	0.42	0.58	0.59	0.62	0.65	0.66	0.65	0.7	0.69	0.76	0.78
Υ			1	0.67	0.71	0.71	0.7	0.71	0.73	0.58	0.84	0.85	0.86	0.84	0.82	0.81	0.54	0.57	0.55	0.56
$\mathrm{TiO}_2$		1	0.46	0.67	0.38	0.37	0.37	0.37	0.38	0.32	0.42	0.44	0.47	0.5	0.5	0.49	0.54	0.51	0.55	0.56
$\mathrm{Al}_{2}\mathrm{O}_{3}$	1	0.78	0.66	0.93	0.5	0.49	0.48	0.49	0.49	0.44	0.56	0.58	0.61	0.64	0.64	0.63	0.68	0.67	0.74	0.76
	$Al_2O_3\\$	$TiO_2$	Υ	Zr	La	Ce	$\mathbf{Pr}$	ΡN	Sm	Eu	$^{\mathrm{Tb}}$	Dy	Но	Tm	Yb	Lu	Νb	Ηf	Та	Th

Table C.1.1 Correlation Coefficients

Sample ID	H500606	H500607	H500608	H500609	H500610	H500611	H500612	H500613	H500614	H500615
Hole ID	BD10-112	BD10-112	BD10-112	BD10-112	BD10-112	BD11-117	BD11-117	BD11-117	BD11-117	BD11-117
Depth (m)	20.1	28.8	39.8	50.9	58.2	12.6	15.2	25.5	45.1	57.7
Lithology	HBX	FLT	FLT	FLT	FLT	HFL	HFL	FLT	FLT	FLT
Alteration	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser
SiO2	19 19	1.16	0.28	24.54	2.80	-5.93	-4 02	3.21	-25.04	11 46
A.O.	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
A <u>12</u> O3	2.24	2 22	8.02	14.86	10.07	1.00	2.54	14.00	6.43	14 70
reo Mao	5.24	3.33	8.02	14.80	10.07	1.88	2.34	14.99	0.43	14.70
MIO	0.17	-0.06	-0.08	-0.09	-0.03	-0.02	0.02	-0.04	-0.04	-0.08
MgO	1.49	-0.19	-1.83	-1.98	3.48	-0.17	1.42	0.51	-0.01	-1.76
CaO	0.17	-0.06	-0.06	-0.09	-0.08	-0.10	-0.08	-0.07	-0.11	-0.04
Na <sub>2</sub> O	-1.85	-1.83	-1.79	-1.80	-1.90	-1.84	-1.88	-1.86	-1.84	-1.81
K <sub>2</sub> O	0.16	0.52	1.00	1.04	-0.54	0.37	0.06	0.36	0.64	0.98
TiO <sub>2</sub>	0.01	0.03	0.02	0.04	0.03	0.01	0.00	0.01	0.02	0.02
$P_2O_5$	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.02	0.00
Net Oxide Change	22.58	2.89	5.56	36.51	13.84	-5.79	-1.95	17.11	-19.94	23.46
Ba	201.25	347.85	376.29	316.08	-508.43	1009.81	363.38	466.48	180.80	244.42
Sr	-13.19	-12.44	-10.64	-12.92	-14.57	-13.35	-14.96	-12.75	-13.35	-12.33
Y	-2.51	-3.95	4.16	-4.19	8.09	0.40	1.24	-5.98	-1.81	3.83
Sc	1.32	3.81	4.76	2.11	2.39	0.57	0.06	-0.75	2.11	1.05
Zr	21.08	15.40	33.32	26.05	23.51	7.80	0.85	8.23	5.49	8.36
Be	-0.37	-0.47	-0.43	-0.24	-0.38	-0.04	-0.50	-0.34	-0.57	-0.31
V	0.65	10.25	0.34	1.28	0.60	-0.10	2.53	4.10	-0.34	0.95
Hg	-8.85	-5.63	40.25	65.20	0.39	-9.60	-5.96	593.01	215.46	39.12
Cr 52	7.62	9.54	0.60	2.29	7.05	-0.17	0.03	9.82	-0.60	1.70
Co	0.39	0.58	-0.60	35.94	35.39	-0.68	1.50	1.39	0.37	48.30
Ni	1.29	0.31	0.68	2.57	1.20	-0.20	0.03	1.60	-0.68	1.91
Cu	349.48	0.44	2496.39	878.41	2925.55	531.52	12.61	71.93	622.75	625.07
Zn	84.78	1.51	-92.81	-37.66	3.65	8.04	52.61	12.98	-52.23	-100.16
As	-1.11	4.87	38.34	90.55	19.71	-0.29	2.90	100.96	39.19	37.34
Se 78	3.45	0.83	1.81	6.86	3.19	-0.52	0.08	4.29	-1.80	5.09
Br 79	7.26	-8.31	18.71	77.41	19.25	-5.65	-4.60	41.44	-28.46	50.57
Мо	1.99	24.49	10.19	32.80	15.82	2.12	1.90	25.11	18.48	27.86
Ag 107	0.13	0.03	0.98	6.87	0.12	-0.02	0.00	1.31	-0.07	0.19
Cd	0.12	0.03	0.06	0.23	0.11	1.18	0.00	0.14	-0.06	0.17
Sn	2.24	0.99	3.26	3.61	1.27	2.65	1.33	3.32	1.21	2.65
Sb	0.51	2.76	5.97	6.73	1.52	0.35	6.50	32.90	15.29	2.19
Te	0.56	0.13	0.29	14.39	4.37	-0.08	0.01	0.69	-0.29	12.98
Ι	4.78	1.15	2.51	9.50	4.42	-0.72	0.11	5.94	-2.50	7.06
W	0.82	1.96	3.34	5.75	1.32	0.18	0.41	2.29	0.80	0.86
Pb	3.27	0.79	17.44	36.79	3.03	-0.49	0.07	114.97	-1.71	4.83
Bi	0.01	0.73	-0.04	28.91	8.50	0.12	0.25	5.42	5.10	16.03
Nb	0.40	0.65	0.76	0.79	1.31	0.42	0.18	-0.20	0.17	0.17
La	-1.14	0.34	-0.35	-2.77	2.60	-2.15	-2.73	0.41	-4.84	3.17
Ce	-2.52	1.54	-1.05	-4.90	4.41	-3.76	-5.58	0.49	-9.73	6.00
Pr	-0.29	0.31	-0.11	-0.59	0.69	-0.41	-0.65	0.11	-1.12	0.73
Nd	-0.80	1.66	-0.51	-2.44	3.30	-1.16	-2.06	0.80	-4.72	3.65
Sm	-0.03	0.41	0.12	-0.51	0.76	-0.23	-0.45	0.10	-1.02	0.75
Eu	-0.21	0.23	0.25	0.19	0.27	-0.39	-0.03	0.32	-0.08	0.11
Tb	-0.05	-0.03	0.06	-0.10	0.20	-0.04	-0.07	-0.19	-0.15	0.11
Dy	-0.13	-0.06	0.57	-0.39	1.31	-0.06	-0.34	-1.25	-0.69	0.75
Но	-0.05	-0.06	0.09	-0.09	0.21	-0.07	-0.10	-0.32	-0.19	0.07
Er	0.02	-0.17	0.14	-0.28	0.64	-0.14	-0.25	-1.02	-0.63	0.14
Tm	0.01	-0.03	0.02	-0.03	0.09	-0.01	-0.02	-0.16	-0.07	0.01
Yb	-0.07	-0.47	-0.09	-0.33	0.34	-0.07	-0.13	-1.21	-0.73	-0.12
Lu	0.00	-0.08	-0.02	-0.06	0.07	0.00	-0.02	-0.19	-0.10	-0.02
Hf	-0.50	-0.33	-0.35	-0.05	-0.45	-0.33	-0.52	-0.88	-0.57	-0.60
Та	0.06	0.06	0.03	0.07	0.05	0.02	0.01	-0.04	-0.02	0.01
Th	-0.14	-0.28	-0.35	-0.05	-0.19	-0.24	-0.57	-0.79	-0.92	-0.45

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

# Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

# Alteration

Sample ID	H500616	H500617	H500618	H500619	H500621	H500622	H500623	H500624	H500625	H500626
Hole ID	BD11-117	BD10-107	BD10-107	BD10-107	BD10-108	BD10-108	BD10-108	BD10-108	BD10-108	BD10-108
Depth (m)	70.4	10.8	18.3	31.2	22.5	28.3	34	36.6	43.5	46.4
Lithology	FBX	FBX	FLT	FLT	FLT	FLT	FBX	FLT	FBX	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl-Ser	Chl-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Chl
SiO <sub>2</sub>	1.46	-5.92	-2.21	-46.09	-0.65	-29.79	-12.99	-3.04	64.02	-17.18
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	1.68	1.29	2.14	6.10	20.91	7.53	6.76	20.21	5.22	16.62
MnO	-0.06	-0.05	-0.01	-0.05	0.16	-0.04	-0.08	-0.08	-0.03	0.09
MgO	-0.47	-0.41	1.84	2.87	1.19	0.94	-1.76	-1.58	-0.58	6.87
CaO	-0.05	0.07	-0.02	0.00	0.12	-0.03	-0.12	-0.09	-0.04	-0.04
Na <sub>2</sub> O	-1.84	-1.83	-1.82	-1.90	-1.82	-1.83	-1.81	-1.79	-1.80	-1.96
K <sub>2</sub> O	0.65	0.60	0.07	-0.86	0.21	0.34	0.90	0.75	0.52	-1.92
TiO <sub>2</sub>	0.02	0.01	0.04	0.02	0.03	0.21	0.06	0.03	0.03	0.12
P <sub>2</sub> O <sub>5</sub>	-0.01	0.02	0.00	0.02	0.00	0.04	0.01	0.00	0.00	0.04
Net Oxide Change	1.40	-6.24	0.03	-39.88	20.15	-22.64	-9.03	14.41	67.32	2.63
Ва	319.65	1358.83	-37.87	-608.76	-94.69	-136.36	723.48	103.74	313.95	-1211.23
Sr	-12.72	-12.50	-11.78	-14.72	-12.13	-10.96	-11.25	-12.58	-11.65	-18.42
Y	4.29	5.67	1.87	0.73	3.32	-7.59	5.96	-1.31	7.59	13.47
Sc	2.37	-0.50	1.24	0.59	5.52	7.83	6.61	2.12	3.80	5.50
Zr	6.53	-6.23	2.13	-6.61	3.12	-40.59	-6.22	13.85	17.19	10.03
Be	-0.48	-0.05	-0.49	-0.67	-0.29	-0.58	-0.51	-0.33	-0.14	-0.40
V	0.08	-0.12	12.83	-0.84	27.12	80.69	19.98	5.58	7.85	28.51
Hg	-9.42	-7.25	-9.45	-7.36	2.11	25.35	155.12	113.19	29.41	4.70
Cr 52	11.23	-0.22	0.10	-1.50	25.65	34.18	15.73	21.46	3.23	10.56
Co	-0.15	-0.68	1.16	3.00	4.69	13.35	7.48	16.15	9.69	34.57
NI Cu	0.15	-0.25	9.93	-1.69	2.05	14.99	0.13	1.75	3.03	0.96
Cu Zn	33 88	530.70 83.78	-/3.81	26.35	-47.54	62.41	364.32	3311./1 84.27	10.03	7.00
As	-55.88	-85.78	-0.07	-20.33	189.15	19.82	38.69	292.12	11.05	17.38
A5 Se 78	0.41	-0.66	0.29	-4.51	5 48	-2.02	-0.30	41.25	9.68	2 57
Br 79	2.77	12.51	10.15	-30.18	157.84	40.01	80.90	186 30	234 96	110 44
Mo	7 24	1.17	4 61	12.27	15.78	2.40	6.18	7.85	13.58	14 82
Ag 107	0.02	-0.02	0.01	-0.17	0.21	0.60	0.66	0.89	0.36	0.10
Cď	0.01	-0.02	0.01	-0.15	0.18	-0.07	1.07	0.16	0.33	0.09
Sn	1.38	-0.90	-2.07	-1.99	2.06	-1.61	-0.41	2.06	1.41	-1.52
Sb	1.01	0.39	-0.09	0.06	0.89	3.76	5.86	12.32	1.79	0.52
Te	0.07	-0.11	0.05	11.16	12.94	-0.32	-0.05	0.74	7.15	11.02
Ι	0.57	-0.92	0.40	-6.25	7.60	-2.80	-0.42	6.40	13.42	3.56
W	0.84	0.86	2.40	2.72	9.25	10.74	7.98	4.21	4.25	6.42
Pb	0.39	-0.63	0.27	-4.28	5.20	-1.91	18.27	4.38	9.18	2.44
Bi	1.19	0.04	0.23	12.92	13.92	3.38	7.25	17.55	6.72	12.29
Nb	0.48	1.79	2.90	2.41	1.55	0.26	1.54	2.63	3.09	4.02
La	-1.81	-0.26	2.36	0.67	2.78	-/.28	-1.83	1.08	4.44	-1.12
Dr	-4.03	-0.92	4.//	0.22	4.05	-13.65	-4./3	2.55	8.52	-2.59
ri Nd	-0.49	-0.02	3.25	0.32	1.02	-7.36	-0.55	1.50	4.23	-0.50
Sm	-0.35	-0.08	0.82	0.44	0.25	-1.52	-2.20	0.41	0.89	0.32
En	-0.14	-0.42	0.32	0.25	0.50	-0.08	0.45	0.41	0.89	0.02
Th	0.05	0.06	0.13	0.09	0.14	-0.21	0.09	0.02	0.17	0.38
Dy	0.69	0.86	1.18	0.97	0.93	-0.96	0.92	0.40	0.98	3.02
Ho	0.10	0.15	0.15	0.13	0.13	-0.24	0.17	0.01	0.21	0.55
Er	0.36	0.42	0.51	0.42	0.38	-0.77	0.39	-0.07	0.59	1.62
Tm	0.06	0.08	0.11	0.07	0.05	-0.11	0.06	0.02	0.11	0.29
Yb	0.27	0.29	0.42	0.17	0.45	-0.80	0.38	0.22	0.48	1.48
Lu	0.04	0.07	0.07	0.05	0.11	-0.15	0.01	0.00	0.05	0.19
Hf	-0.56	0.90	1.38	1.34	1.01	-0.63	-0.04	0.82	0.48	0.46
Та	0.03	0.07	0.12	0.07	0.03	-0.04	-0.01	0.07	0.11	0.11
Th	-0.34	1 40	1.99	1 48	0.32	-1 64	-0.31	0.60	0.91	0.30

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow

INT= Intrusion

Alteration

Sample ID	H500627	H500628	H500629	H500642	H500643	H500644	H500645	H500649	H500650
Hole ID	BD10-108	BD10-108	BD10-108	BD10-104	BD10-104	BD11-120	BD11-120	BD10-88	BD10-88
Depth (m)	47.4	50.6	53.2	14.3	21.6	22.8	63.3	8.2	22.1
Lithology	FLT	FFL	FLT	FLT	FLT	HFL	FLT	FLT	FLT
Alteration	Chl	Qtz-Ser	Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl-Ser
SiO <sub>2</sub>	-42.48	-18.57	-5.04	0.00	21.15	27.72	56.80	18.13	13.18
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	14.64	2.02	16.96	1.13	4.72	2.03	10.88	5.71	5.21
MnO	-0.01	-0.08	0.12	-0.03	-0.05	0.03	-0.07	-0.05	0.01
MgO	5.27	-0.53	6.24	-0.77	1.46	-0.57	1.75	2.41	2.53
CaO	-0.08	-0.13	-0.11	-0.03	-0.03	0.26	-0.06	-0.10	-0.10
Na <sub>2</sub> O	-1.93	-1.82	-1.98	-1.78	-1.82	-1.83	-1.80	-1.87	-1.85
K <sub>2</sub> O	-1.29	0.38	-2.08	0.75	0.05	0.74	-0.26	-0.41	-0.63
TiO <sub>2</sub>	0.03	0.03	0.04	0.02	0.05	0.01	0.10	0.06	0.08
P <sub>2</sub> O <sub>6</sub>	0.02	0.01	0.00	0.01	0.02	0.02	0.00	0.00	0.00
Net Oxide Change	-25.83	-18.69	14 14	-0.70	25.56	28.41	67 35	23.88	18 43
Ba	-855.60	-108.17	-1269.04	-22.07	-412.68	3.33	-686 18	-399.80	-692.15
Sr	-16 67	-13.67	-20.69	-8.94	-11.58	-10.26	-9.52	-14 22	-10.87
Ŷ	5.40	-4.84	2.95	-8.87	0.08	-5.09	3.77	-4.01	-0.67
Sc	2.43	3.33	3.11	2.05	1.72	-0.87	2.48	6.85	6.07
Zr	22.85	9.98	13.62	-12.43	-21.24	-1.52	54.61	-1.16	-0.35
Be	-0.56	-0.58	-0.34	0.00	-0.35	-0.35	-0.11	-0.35	-0.38
V	1.94	4.17	4.05	4.53	18.34	0.76	17.10	20.83	23.46
Hg	2.21	-6.17	32.57	-9.49	-8.74	-2.87	-7.54	26.88	-8.91
Cr 52	4.81	-0.74	1.38	0.02	7.38	1.35	18.17	12.63	20.26
Co	24.71	2.11	24.98	0.67	2.34	-0.52	30.69	9.31	14.38
Ni	-0.56	-0.83	1.55	0.02	1.51	1.52	3.91	1.48	1.18
Cu	317.75	920.83	21918.18	-73.97	294.93	38.09	-66.70	172.37	4.51
Zn	6.12	-118.59	112.56	26.37	-25.27	-64.37	-103.75	-4.75	56.16
As	126.18	4.55	18.16	-1.32	7.16	6.95	6.99	-1.08	13.77
Se 78	-1.49	-2.23	4.15	0.06	4.04	4.06	10.44	3.95	3.16
Br 79	48.95	0.40	141.91	57.77	59.27	50.91	281.71	21.07	19.82
Mo	9.06	4.87	24.95	41.45	9.60	4.44	24.43	4.32	3.09
Ag 107	-0.06	-0.08	0.16	0.00	0.15	0.15	0.39	0.15	0.12
Cd	-0.05	-0.08	0.14	0.00	0.14	0.14	0.35	0.13	0.11
Sn	-1.93	-2.15	0.07	1.21	1.24	2.09	0.50	1.57	0.31
SD	0.80	0.34	2.37	0.22	0.46	1.21	0.53	-0.31	0.09
IC	2.15	2.00	5.87	0.01	5.92	0.05	1.08	0.04 5.47	0.51
I W	-2.07	-5.09	7.07	3.70	4.70	1.25	8 86	7.42	4.37
Ph	1.15	2.11	25.24	0.05	4.70	3.85	0.00	3 75	2.09
Bi	8 44	5 75	10.00	0.88	4.89	3 35	2.36	0.45	0.32
Nh	3.45	3.01	3 76	0.98	0.91	-0.71	1 34	-0.68	-0.62
La	1.69	0.37	-0.83	-17.67	-4.14	-2.39	-5.06	-0.60	-2.20
Ce	3.38	0.51	-2.61	-37.50	-8.98	-5.06	-12.89	-7.47	-6.38
Pr	0.45	0.00	-0.29	-4.66	-1.15	-0.65	-1.66	-1.21	-0.83
Nd	2.13	-0.38	-0.98	-19.21	-4.36	-2.15	-7.81	-4.89	-3.47
Sm	0.47	-0.17	-0.02	-4.21	-1.02	-0.83	-1.77	-1.39	-0.93
Eu	0.11	-0.05	-0.15	-0.84	-0.38	-0.32	-0.46	-0.35	-0.47
Tb	0.14	-0.11	0.19	-0.46	-0.09	-0.21	-0.17	-0.22	-0.11
Dy	1.32	-0.50	1.30	-2.26	-0.13	-1.23	-0.66	-1.59	-0.40
Но	0.27	-0.16	0.25	-0.41	0.06	-0.32	-0.04	-0.34	-0.14
Er	0.74	-0.45	0.64	-1.13	0.08	-1.01	-0.06	-0.85	-0.29
Tm	0.14	-0.06	0.11	-0.13	0.04	-0.13	-0.02	-0.12	-0.04
Yb	0.78	-0.52	0.54	-0.81	0.10	-0.91	0.05	-0.87	-0.40
Lu	0.10	-0.09	0.06	-0.13	-0.01	-0.13	0.01	-0.12	-0.05
Hf	1.39	-0.13	1.51	0.26	0.58	-0.94	-0.62	-1.46	-1.24
Та	0.08	0.08	0.06	0.06	0.05	0.00	0.04	-0.02	0.00
Th	0.89	0.49	0.52	0.42	0.82	-0.63	0.01	-0.94	-0.66

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

#### Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H500661	H500662	H500663	H500664	H500665	H500666	H500667	H500668	H500669	H500670
Hole ID	BD10-102	BD10-102	BD10-102	BD10-102	BD11-129	BD11-129	BD11-129	BD11-129	BD11-129	BD11-129
Depth (m)	6.2	16.6	25.1	43.7	12.8	24.8	38.1	35.4	38.4	49.4
Lithology	FLT	FLT	FLT	FLT	FLT	FLT	FLT	FFL	FLT	FBX
Alteration	Qtz-Ser	Chl	Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	1.56	-61.08	28.38	17.36	-14.32	14.70	28.92	34.39	13.81	-0.07
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	24.04	16.89	36.38	5.62	0.50	15.95	19.43	5.25	12.88	4.56
MnO	-0.07	0.00	0.04	-0.03	-0.05	-0.09	-0.05	-0.09	-0.09	-0.08
MgO	-2.04	9.83	8.45	1.21	1.39	-2.11	-1.75	-1.99	-2.05	-0.46
CaO	-0.05	-0.11	-0.04	-0.09	-0.10	-0.11	-0.02	-0.12	-0.08	-0.13
Na <sub>2</sub> O	-1.79	-2.01	-1.97	-1.83	-1.82	-1.75	-1.75	-1.77	-1.74	-1.80
K <sub>2</sub> O	0.89	-2.32	-2.12	-0.28	0.30	0.90	0.91	0.89	0.97	0.52
TiO <sub>2</sub>	0.08	0.05	0.06	0.08	0.02	0.05	0.05	0.03	0.03	0.03
$P_2O_5$	0.00	0.01	0.01	0.01	0.02	0.00	0.00	0.00	0.00	0.00
Net Oxide Change	22.61	-38.74	69.18	22.06	-14.06	27.54	45.74	36.58	23.73	2.55
Ba	436.97	-1390.81	-1313.44	-589.53	403.83	247.85	162.72	149.35	93.35	-4.81
Sr	-2.93	-19.66	-13.78	-10.51	-9.90	-10.52	-8.79	-13.49	-11.07	-13.43
Y	5.00	7.10	8.28	0.58	-6.15	4.05	-2.66	0.71	6.08	3.85
Sc	3.20	2.50	4.39	4.05	2.10	2.91	1.56	1.35	0.93	3.92
Zr	7.53	26.13	31.78	5.67	8.87	4.23	0.17	6.76	2.54	12.46
Be V	-0.27	-0.01	0.03	-0.50	-0.14	-0.28	-0.17	-0.29	-0.52	-0.40
v Ha	15.10	72 39	9.65	29.42	10.32	20.40	379.44	7.43 57.50	19.30	14 78
Cr 52	14 28	-0.97	24 35	15.10	-0.60	1 94	2.90	21.07	1 63	0.32
Co	36.24	83.22	272.76	5.07	-0.72	18 17	73.10	14.51	12.02	27.29
Ni	2.33	-1.09	5.28	1.38	-0.68	2.17	3.26	2.09	1.83	0.36
Cu	12107.27	21512.26	34653.18	263.39	36.60	226.37	23457.83	1325.49	-70.59	-39.84
Zn	322.35	42.01	194.25	-57.40	-5.41	-81.97	-46.69	-95.49	-49.47	-119.43
As	31.83	24.49	62.57	2.67	1.16	65.69	59.93	10.87	30.48	81.72
Se 78	6.23	13.73	112.49	3.70	-1.81	5.81	50.67	5.58	4.89	0.95
Br 79	79.50	-35.15	158.14	16.99	29.72	138.15	195.75	144.48	175.18	81.36
Mo	46.68	47.35	55.31	2.44	5.00	9.34	27.05	6.27	4.54	15.67
Ag 107	1.85	1.85	3.88	0.14	-0.07	0.96	4.50	0.21	0.18	0.04
Cd	0.21	-0.10	0.48	0.12	-0.06	0.20	0.29	0.19	0.16	0.03
Sn Sh	0.93	0.24	3.13	1.68	-2.22	0.27	2.48	-0.49	-0.80	-0.//
Та	1.01	1.64	02.37	0.01	0.14	8.33 7.97	24.77	0.90	0.79	1.11
I	8.63	-4 04	19 53	5.12	-2.51	8.05	12.06	7 74	6.77	1 32
W	4 25	19.42	30.73	10.86	2.18	3.95	3.01	4 14	4.08	2.41
Pb	5.91	-2.77	13.36	3.51	-1.72	5.51	43.18	5.30	4.63	0.90
Bi	6.14	65.13	90.21	1.95	-0.02	34.60	44.43	6.93	2.28	15.42
Nb	0.40	0.31	0.10	1.18	3.24	2.34	2.96	2.14	2.22	3.03
La	3.14	5.73	-1.49	-7.36	1.29	1.82	0.85	1.39	1.13	-1.20
Ce	4.87	7.00	-4.25	-15.21	2.26	2.84	0.52	2.10	1.45	-3.46
Pr	0.49	0.86	-0.54	-1.75	0.44	0.44	0.09	0.40	0.15	-0.39
Nd	2.45	4.24	-1.77	-6.85	2.51	3.02	0.19	1.58	0.95	-1.07
Sm	0.45	1.30	-0.70	-1.36	0.39	0.82	0.18	0.14	0.24	-0.27
Eu	0.07	-0.39	-0.68	-0.41	0.24	0.24	0.16	-0.18	0.09	0.56
10 Dv	0.07	0.20	0.09	-0.08	-0.05	0.17	0.00	0.04	0.09	0.07
Но	0.73	1.24	0.99	0.02	-0.59	0.27	-0.01	0.57	0.71	0.85
Fr	0.00	0.15	0.25	0.05	-0.14	1.08	-0.04	0.05	0.10	0.10
Tm	0.17	0.07	0.07	0.04	-0.04	0.15	-0.02	0.02	0.09	0.47
Yb	-0.05	0.47	0.65	0.12	-0.62	0.62	-0.01	0.31	0.47	0.43
Lu	0.01	0.05	0.08	0.01	-0.08	0.11	0.00	0.04	0.08	0.05
Hf	-0.46	-0.60	-0.53	-0.55	0.62	0.84	0.60	-0.20	0.29	1.24
Та	-0.06	0.02	0.02	0.02	0.07	0.07	0.03	0.05	0.00	0.09
Th	-0.50	0.34	-0.08	0.08	0.44	0.69	0.31	0.48	0.02	0.63

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

# Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H500671	H500672	H500673	H500674	H500675	H500676	H500677	H500678	H500679	H500680
Hole ID	BD11-129	BD11-129	BD11-123	BD11-123	BD11-123	BD11-123	BD11-123	BD10-87	BD10-87	BD10-87
Depth (m)	53.8	56.4	7.4	9	10.8	14.8	22.3	6.3	11	17.9
Lithology	FFL	FLT	HFL	FLT	FFL	FLT	FLT	FLT	FLT	FT
Alteration	Chl-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	-3.84	-22.60	6.24	-17.16	2.06	-4.54	-25.15	-30.60	14.29	1.01
A <sub>l2</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	6.71	3.22	1.25	2.45	1.97	3.19	2.78	14.58	14.97	5.74
MnO	-0.04	-0.03	0.02	0.02	0.15	0.11	-0.01	0.11	0.04	-0.05
MgO	3.93	2.25	0.08	0.25	0.56	1.30	-0.61	6.40	6.52	0.74
CaO	-0.13	-0.11	-0.09	-0.06	-0.09	-0.09	-0.09	-0.11	-0.11	-0.11
Na <sub>2</sub> O	-1.91	-1.86	-1.73	-1.76	-1.82	-1.81	-1.77	-1.95	-1.96	-1.81
K <sub>2</sub> O	-1.06	-0.31	0.58	0.61	0.73	0.53	0.82	-1.68	-1.81	-0.11
TiO <sub>2</sub>	0.02	0.03	0.01	0.02	0.02	0.02	0.03	0.07	0.07	0.04
$P_2O_5$	0.00	0.02	0.01	0.02	0.00	0.02	0.01	-0.01	0.00	0.00
Net Oxide Change	3.67	-19.38	6.38	-15.61	3.57	-1.27	-23.98	-13.18	32.03	5.43
Ba	-768.90	-396.97	575.56	399.93	275.42	12.65	1049.17	-1114.97	-1168.40	-506.41
Sr	-17.58	-15.29	-9.08	-10.71	-13.56	-12.83	-9.39	-15.92	-16.12	-9.77
Y	11.77	12.14	1.90	1.81	0.05	6.86	4.34	-14.67	25.68	8.80
Sc	2.14	3.41	0.76	2.16	3.72	2.23	2.61	1.15	1.76	1.12
Zr	16.15	8.41	37.97	20.86	15.43	20.10	7.88	32.44	35.80	-4.93
Ве	-0.45	-0.58	0.08	-0.13	0.06	-0.49	-0.21	-0.49	-0.27	-0.44
V	6.33	5.04	0.19	3.58	0.14	0.05	3.80	4.60	6.32	6.39
Hg	1.25	-9.90	20.29	20.13	4.89	5.33	36.07	4.22	11.52	0.23
Cr 52	0.46	-0.72	5.37	4.33	0.25	0.09	2.90	0.06	9.10	0.50
C0	15.57	/.05	1.76	2.30	-0.64	-0.65	0.57	8.13	20.62	15.94
INI Cu	0.52	-0.81	0.38	-0.00	0.28	0.10	5.30	0.07	2.35	0.50
Cu 7n	-/3.04	-73.32	-50.87	37.17	38.81	17 55	-36.12	-44.97	-09.02	-72.97
As	6.77	-57.45	4.10	-57.17	-56.81	-17.33	15 52	13.90	2 45	-80.07
A5 Se 78	1 39	-2.16	4.10	-1.76	0.74	0.07	-2.83	0.19	6.27	1.49
Br 79	95 73	24.80	102 35	60.82	118.81	92.34	48.45	106.63	197 59	99.27
Mo	11.93	6.06	23.26	11.37	7.01	4 76	5 90	11 44	16 38	20.37
Ag 107	0.05	-0.08	0.04	-0.07	0.03	0.01	-0.11	0.01	0.23	0.06
Cd	0.05	-0.07	0.03	1.25	0.02	0.01	-0.10	0.01	0.21	0.05
Sn	-2.08	-2.37	-1.27	-1.57	-0.97	-1.47	-1.79	-1.43	-0.74	-1.04
Sb	-0.07	0.19	0.58	0.79	0.82	0.37	7.32	0.94	0.21	-0.11
Те	0.22	-0.35	0.16	-0.28	0.12	0.04	-0.46	2.90	1.01	0.24
Ι	1.92	-2.99	1.41	-2.44	1.03	0.35	-3.92	0.26	8.69	2.07
W	2.42	3.36	0.66	1.87	10.25	1.85	2.94	12.17	10.01	6.34
Pb	1.31	-2.05	0.97	12.23	0.70	0.24	-2.68	0.18	5.95	1.41
Bi	2.32	5.58	0.09	1.10	0.02	0.05	1.56	4.91	7.39	2.02
Nb	2.88	3.15	0.60	0.65	1.14	1.08	1.16	2.82	2.57	0.63
La	3.30	-0.68	-2.00	-2.21	-1.84	-0.27	1.18	-16.21	12.48	16.60
Ce	6.66	-1.54	-4.48	-4.60	-4.41	-1.43	2.14	-34.64	27.41	35.98
Pr	0.80	-0.12	-0.53	-0.66	-0.51	-0.08	0.30	-4.32	3.64	4.69
Nd	3.48	-0.22	-1.78	-2.37	-2.07	-0.40	1.49	-18.14	14.47	21.82
Sm	1.47	0.29	-0.23	-0.34	-0.36	0.31	0.68	-4.20	3.71	4.70
Eu	0.20	0.05	-0.01	0.04	-0.32	-0.06	0.99	-0.87	0.02	0.03
10 Du	0.37	0.20	-0.03	-0.06	-0.01	0.09	0.00	-0.60	0.55	0.39
Бу Но	2.75	1.72	0.15	0.04	0.32	1.07	0.08	-5.12	4.50	2.40
Er	1 10	1.10	0.09	0.00	0.05	0.22	0.13	-0.57	0.09	0.40
Li Tm	0.14	0.16	0.29	0.10	0.17	0.08	0.45	-1.27	2.42	0.88
Yh	0.14	1 14	0.07	-0.12	-0.02	0.00	0.07	-0.11	1.98	0.10
In	0.15	0.20	0.25	-0.12	0.07	0.08	0.05	-0.52	0.26	_0.02
Hf	1 12	-0.19	-0.34	-0.04	-0 74	-1 04	-0.78	0.05	0.20	-1 32
Та	0.09	0.07	0.01	0.01	0.02	0.01	0.03	0.06	0.07	-0.01
Th	0.61	0.23	-0.05	-0.02	-0.11	0.03	0.06	0.58	0.82	-0.45

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H500681	H500682	H500683	H500684	H500685	H500686	H500687	H500688	H500689	H500690
Hole ID	BD10-87	BD10-77	BD10-77	BD10-77	BD10-77	BD10-77	BD10-76	BD10-76	BD10-76	BD10-76
Depth (m)	30.5	12.4	22.3	26	29.5	49.1	48.2	36	27.2	7.7
Lithology	FLT	FLT	FBX	FLT	FLT	FLT	FLT	FLT	FLT	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Chl	Qtz-Ser
SiO <sub>2</sub>	3.48	4.62	-42.34	-3.13	-10.70	16.60	23.80	21.23	-42.79	39.92
A <sub>l2</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	3.89	4.26	5.12	1.86	20.19	0.70	11.13	2.69	17.31	8.21
MnO	-0.07	0.00	-0.07	-0.04	0.06	-0.09	0.00	-0.08	0.04	-0.06
MgO	-0.58	0.61	-1.41	1.18	5.48	-1.52	3.08	-1.57	9.40	-0.87
CaO	-0.12	-0.01	-0.07	-0.07	-0.08	-0.10	-0.09	-0.08	-0.11	-0.10
Na <sub>2</sub> O	-1.79	-1.81	-1.81	-1.82	-1.89	-1.80	-1.84	-1.75	-1.99	-1.80
K <sub>2</sub> O	0.51	0.78	1.07	0.39	-0.76	0.82	-0.67	0.88	-2.00	0.80
TiO <sub>2</sub>	0.03	0.05	0.06	0.03	0.02	0.03	0.04	0.04	0.03	0.02
$P_2O_5$	0.00	0.00	0.02	0.02	0.02	0.01	0.00	0.01	-0.01	0.00
Net Oxide Change	5.34	8.48	-39.45	-1.59	12.32	14.66	35.45	21.39	-20.11	46.11
Ba	-256.68	499.52	421.01	-75.40	-818.32	-313.29	-776.25	-235.51	-1256.38	8.62
Sr	-9.98	-9.60	-12.81	-10.97	-12.72	-11.62	-13.21	-9.59	-19.08	-11.20
Y	6.88	6.10	0.40	-3.91	-15.15	1.37	9.37	4.43	6.68	4.21
Sc	2.02	2.40	1.16	2.03	-0.72	2.69	6.12	3.65	5.55	2.35
Zr	5.74	28.62	6.05	-7.63	8.71	-1.96	16.24	1.41	0.79	18.58
Be	-0.45	-0.44	-0.34	-0.50	-0.34	-0.42	-0.27	-0.38	-0.51	-0.23
v Ha	-9.27	7.05	33.97	-9.49	38 35	-9.12	-8.34	-8.90	-6.17	-2.74
Cr 52	0.41	0.57	-1.53	0.01	1 4 5	0.68	2.07	1.07	-0.17	2.42
Co	48.57	0.10	0.12	-0.66	14 62	21.30	76.20	19.06	79.74	21.64
Ni	0.46	0.64	-1.72	0.01	1.63	0.77	2.33	1.20	-0.14	2.72
Cu	61.64	535.57	-48.70	-73.98	42110.18	264.76	24.86	-18.64	-48.54	805.84
Zn	-92.77	-11.97	-82.04	-39.64	453.99	-111.57	-92.77	-116.26	44.87	3.57
As	19.29	12.22	13.80	-1.32	9.75	0.44	19.91	1.74	10.97	10.69
Se 78	1.23	1.70	-4.58	0.04	34.24	2.05	6.22	3.21	20.99	7.25
Br 79	10.77	72.64	-49.54	20.94	68.84	54.93	127.05	56.48	23.65	215.76
Mo	2.39	12.34	25.74	3.16	10.11	13.39	2.70	12.44	48.46	11.01
Ag 107	0.05	0.06	-0.17	0.00	2.88	0.08	0.23	0.12	-0.01	0.27
Cd	0.04	0.06	-0.15	0.00	0.15	0.07	0.21	0.11	-0.01	0.24
Sn	2.90	0.76	-1.30	-1.70	4.99	-1.61	-0.23	-0.36	-2.20	0.21
50 To	0.18	1./1	4.72	0.16	2.73	0.20	1.82	0.17	5.80	0.71
I	1.71	2.36	-6.35	0.01	6.01	2.84	8.01	3.73	-0.52	1.17
W	4 83	4 58	4 83	5.86	2.84	3.04	7.02	3 27	7.06	3 90
Pb	1.17	1.61	8 09	0.03	4.12	1.94	5.90	3.05	-0.36	6.88
Bi	3.62	7.60	6.75	1.29	5.69	2.34	11.97	9.24	12.21	2.57
Nb	-0.15	2.86	2.73	0.98	1.34	-0.04	1.68	0.64	2.54	1.01
La	-6.48	3.52	-1.66	-8.08	-12.33	4.09	3.73	0.21	-0.85	-2.07
Ce	-13.19	6.39	-3.06	-17.25	-27.04	9.82	8.13	-1.24	-2.36	-3.68
Pr	-1.57	0.90	-0.39	-2.12	-3.32	1.49	1.22	-0.09	-0.32	-0.38
Nd	-5.69	4.54	-1.50	-8.38	-13.66	6.25	5.89	0.21	-0.63	-1.99
Sm	-1.12	1.02	-0.22	-1.82	-3.13	1.11	1.40	-0.23	0.12	0.02
Eu	-0.68	0.30	0.18	-0.28	-0.74	-0.15	-0.38	-0.32	-0.09	-0.59
Tb	-0.08	0.21	-0.08	-0.29	-0.46	0.02	0.18	-0.01	0.13	-0.01
Dy Но	-0.01	1.45	-0.16	-1.58	-2.51	0.05	1.46	0.05	1.52	0.15
Er	-0.02	0.45	0.00	-0.25	-0.50	0.00	0.29	0.00	0.31	-0.14
ы Тт	-0.05	0.21	0.03	-0.04	-1.22	-0.08	0.88	-0.04	0.81	-0.50
Yh	-0.23	1.30	-0.01	-0.51	-0.63	0.01	0.54	0.04	0.14	-0.69
Lu	-0.04	0.15	0.01	-0.07	-0.10	-0.02	0.11	-0.02	0.10	-0.05
Hf	-0.90	0.79	-0.01	-0.61	0.40	0.08	0.64	0.35	-0.13	0.15
Та	-0.01	0.09	0.05	0.03	0.03	0.02	0.07	0.02	0.05	0.04
Th	-0.81	2.12	0.89	0.36	0.56	0.58	1.11	0.63	0.50	-0.20

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H500691	H500692	H500693	H500694	H500695	H500696	H500697	H500698	H500699	H500700
Hole ID	BD10-63	BD10-63	BD10-63	BD10-63	BD10-63	BD10-63	BD10-63	BD10-61	BD10-61	BD10-61
Depth (m)	11.4	21.3	37.5	57.3	59.1	64.1	75.4	11.1	24.9	37.9
Lithology	HBX	HFL	FLT	FLT	FBX	FFL	FBX	FLT	FLT	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Chl	Chl-Ser
SiO <sub>2</sub>	19.44	16.09	-13.69	16.24	8.65	8.78	12.30	35.70	-21.37	22.68
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	0.69	6.98	2.61	37.33	2.19	1.63	4.16	20.12	4.51	12.89
MnO	0.19	-0.05	0.05	0.09	-0.04	-0.05	-0.06	-0.09	0.10	0.00
MgO	0.62	-1.75	0.18	5.77	0.17	0.29	-0.21	-2.04	10.32	5.35
CaO	0.04	-0.10	-0.09	-0.07	-0.02	-0.08	-0.07	-0.10	0.71	-0.11
Na <sub>2</sub> O	-1.88	-1.87	-1.86	-1.91	-1.83	-1.83	-1.83	-1.81	-1.97	-1.94
K <sub>2</sub> O	1.07	1.25	0.78	-1.40	0.67	0.69	0.64	1.11	-1.84	-1.65
TiO <sub>2</sub>	0.02	0.00	0.05	0.01	0.06	0.05	0.06	0.04	0.24	0.11
$P_2O_5$	0.02	0.00	0.02	0.00	0.02	0.02	0.01	0.02	0.07	0.00
Net Oxide Change	20.22	20.55	-11.97	56.05	9.87	9.51	15.01	52.95	-9.24	37.33
Ba	539.84	1141.37	529.87	-992.22	-112.56	-182.21	-145.97	1347.30	-1110.74	-1095.55
Sr	-13.54	-14.35	-15.66	-14.26	-13.04	-14.22	-12.47	-15.08	-10.57	-19.00
Y	3.32	-2.02	-2.74	-8.06	1.34	5.45	-4.44	2.54	-8.51	7.57
Sc	0.88	-1.07	1.78	-0.33	0.09	2.22	1.92	2.12	5.24	3.52
Zr	-12.00	18.24	-16.29	7.78	5.16	2.81	5.16	9.03	-31.62	8.22
Be	0.21	-0.36	-0.55	-0.03	-0.44	-0.44	-0.40	-0.13	-0.52	-0.25
V	3.55	5.15	6.56	2.33	6.46	0.28	0.48	1.83	70.86	14.02
Hg	463.17	4030.78	80.45	69.22	-0.79	-9.22	-9.02	/8.00	0.39	4.52
Cr 52	10.35	1.23	-0.42	4.16	0.54	0.50	0.85	45.15	51.57	11.32
C0 Ni	-0.57	-0.54	0.13	37.32	0.11	-0.61	0.95	15.58	3.00	2 51
Cu	86.38	2008.26	621.22	50053.08	481.00	78.13	8.45	29.82	59.02	711.40
Zn	5513.41	37977 79	260.36	280.30	-54.12	-71.20	91.15	351.98	197.46	-27.58
As	3.68	158.79	11.51	56.84	1.34	3.49	13.42	123.51	-1.36	29.80
Se 78	2.79	3.68	-1.25	70.14	1.61	1.49	2.56	9.76	-0.63	6.70
Br 79	43.36	50.44	-26.66	160.16	12.07	30.21	49.61	105.12	11.66	103.23
Мо	6.84	6.37	3.40	19.75	16.77	2.24	4.70	20.91	1.40	21.90
Ag 107	0.10	4.83	-0.05	4.91	0.06	0.06	0.10	1.27	-0.02	0.25
Cd	16.88	120.83	0.77	0.42	0.05	0.05	0.09	1.18	-0.02	0.23
Sn	-1.69	4.92	-2.37	0.40	-0.96	-1.98	-1.68	0.57	-2.38	-1.42
Sb	3.97	14.62	4.97	4.21	1.13	0.54	1.38	5.31	-0.13	0.11
Те	0.45	0.59	-0.20	19.01	0.26	0.24	0.41	9.24	-0.10	7.96
Ι	3.87	5.09	-1.73	17.28	2.23	2.06	3.55	13.52	-0.87	9.29
W	2.38	6.55	1.73	8.19	0.37	0.89	0.54	2.90	1.09	11.63
Pb	594.12	4444.13	41.64	41.00	1.53	1.41	2.43	215.79	51.43	6.36
Bi	0.30	5.23	1.20	34.39	1.64	0.46	1.32	10.19	0.44	10.06
Nb	0.88	0.46	0.83	2.09	2.05	1.10	0.46	1.10	0.50	1.56
La Ce	-5.50	-2.57	-2.50	-5.54	2.09	-2.31	-1.45	0.55	-8.03	-4.62
Pr	-0.75	-4.01	-0.64	-1.12	0.17	-0.66	-0.44	-0.04	-2.08	-1.36
Nd	-2.55	-2.05	-2.24	-4 66	0.56	-2.70	-1.58	0.33	-8.94	-5.48
Sm	-0.60	-0.35	-0.65	-1.01	0.13	-0.54	-0.40	0.44	-1.92	-0.88
Eu	-0.25	0.38	-0.17	-0.22	-0.09	-0.18	-0.30	0.32	-0.19	-0.41
Tb	0.02	-0.03	-0.11	-0.09	-0.03	-0.02	-0.10	0.07	-0.30	0.09
Dy	0.41	0.18	-0.46	0.07	-0.10	0.34	-0.55	0.81	-1.44	0.97
Но	0.10	0.00	-0.12	0.01	-0.05	-0.04	-0.22	0.13	-0.34	0.05
Er	0.31	-0.09	-0.42	-0.14	-0.19	-0.04	-0.76	0.21	-1.01	0.28
Tm	0.07	0.00	-0.05	0.02	-0.02	-0.01	-0.11	0.05	-0.13	0.04
Yb	0.24	0.12	-0.50	-0.02	-0.32	-0.25	-0.84	0.10	-1.11	0.19
Lu	0.05	0.01	-0.08	0.01	-0.06	-0.02	-0.13	0.04	-0.16	0.03
Hf	-0.26	0.29	-0.54	-0.16	-0.14	0.07	-0.41	0.41	-0.72	-0.31
Та	0.00	0.02	-0.01	0.04	0.04	0.04	0.01	0.04	-0.01	0.04
Ih	0.32	0.58	-0.35	0.58	0.69	0.18	0.19	0.73	-1.27	0.13

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

# Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501201	H501202	H501203	H501204	H501205	H501206	H501207	H501208	H501212	H501213
Hole ID	BD10-61	BD10-61	BD10-58	BD10-58	BD10-58	BD10-58	BD10-58	BD10-58	BD10-40	BD10-40
Denth (m)	20.5	44.4	8 2	15	18.2	20.5	32.3	34.0	1	0.5
Lithology	59.5 EDV	44.4 EDV	6.2 ELT	IJ ELT	18.2 ELT	29.5 ELT	52.5	54.9 ELT	4	9.5 UDV
Alternation	Chl See	Chl San	Chl San		Chlfen	Chl San	Chl San	Chl San	Ota Sar	
Alteration	Chi-Ser	Chi-Ser	Chi-Ser	Chi	Chi-Ser	Chi-Ser	Chi-Ser	Chi-Ser	Qtz-Ser	Qtz-Ser
SIO <sub>2</sub>	28.54	17.46	34.78	-61.98	23.77	2.28	82.59	31.84	8.57	-11.23
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	9.98	5.86	28.83	31.71	19.31	32.38	51.62	14.75	3.95	4.92
MnO	-0.01	0.25	-0.01	0.06	-0.04	0.06	0.01	-0.02	-0.01	0.09
MgO	4.80	3.53	3.69	6.21	-1.32	2.58	-0.31	-0.37	-1.07	-0.14
CaO	-0.09	2.20	-0.07	-0.12	-0.07	-0.08	-0.05	-0.09	-0.11	-0.10
Na <sub>2</sub> O	-1.93	-1.88	-1.88	-1.98	-1.74	-1.87	-1.81	-1.79	-1.78	-1.75
K <sub>2</sub> O	-1.35	-0.37	-1.16	-2.06	0.74	-1.05	0.03	-0.08	0.82	0.63
TiO	0.06	0.06	0.06	0.05	0.04	-0.01	-0.01	0.00	0.00	0.01
P.O	0.02	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.01
$\Gamma_2 O_5$	0.02	0.00	0.00	0.00	0.00	0.00	122.00	0.00	0.01	0.01
Net Oxide Change	40.03	27.11	64.24	-28.12	40.68	34.29	132.08	44.23	10.39	-7.56
Ва	-943.97	-339.00	-084.01	-1233.01	137.43	-/8/.9/	-329.95	-414.32	101.29	-/0.98
Sr	-17.50	15.23	-16.21	-18.89	-6.00	-13.67	-7.82	-10.87	-12.85	-10.32
Ŷ	6.00	2.22	-10.04	-11.03	-8.60	-10.67	-10.63	0.75	23.89	4.82
Sc	2.00	3.30	3.51	2.43	4.40	1.67	1.35	5.90	1.43	0.71
Zr	12.51	-5.38	-11.79	30.07	19.78	-18.33	8.20	-5.09	15.08	3.97
Be	-0.25	-0.34	-0.03	-0.48	-0.20	-0.17	0.42	-0.21	-0.43	-0.03
V	5.00	0.82	20.67	25.47	23.10	14.17	4.59	7.04	5.50	4.32
Hg	6.00	-4.02	264.08	107.13	165.58	89.67	963.92	40.47	6.29	-3.24
Cr 52	2.23	29.64	25.57	0.16	10.35	11.37	38.33	2.62	0.64	26.95
Co	40.70	15.87	159.65	43.99	40.87	141.51	14.15	21.82	44.59	1.92
Ni	2.50	24.68	4.65	6.87	3.00	3.33	9.18	2.95	0.72	-0.13
Cu	520.59	795.31	2531.87	6935.00	12558.94	27224.77	47928.80	6162.45	3482.14	-28.87
Zn	-41.22	-63.88	-21.34	155.18	-23.69	67.52	367.82	-40.60	-94.35	128.87
As	3.52	2.36	70.41	312.41	62.71	132.82	668.06	24.45	32.27	7.58
Se 78	6.68	4.40	116.89	63.26	56.29	112.62	45.40	7.87	1.91	-0.35
Br 79	150.47	57.41	184.99	-19.88	88.70	81.01	295.40	109.78	115.97	74.50
Mo	12.43	15.67	94.22	61.49	42.25	93.02	39.19	23.09	12.15	8.22
Ag 107	0.25	0.16	0.47	0.86	0.30	3.17	9.85	0.29	0.07	-0.01
Cd	0.23	0.15	0.42	0.02	0.27	0.30	1.51	0.27	0.06	-0.01
Sn	-1.36	-1.85	0.45	-0.94	1.67	1.36	9.60	2.42	-0.89	-1.79
Sb	0.44	0.39	1.75	3.73	3.26	7.41	102.67	1.06	4.61	0.53
Те	1.08	0.71	16.87	44.59	16.81	412.04	26.70	7.77	10.52	-0.06
Ι	9.25	6.10	17.22	0.66	11.10	12.33	33.98	10.91	2.65	-0.49
W	2.07	1.43	14.97	15.09	5.72	9.51	8.50	2.64	2.76	2.00
Pb	6.33	4.17	11.78	39.79	19.67	31.29	129.42	7.47	1.82	-0.33
Bi	4.40	2.57	17.73	67.49	31.15	624.23	55.02	14.58	17.61	1.00
Nb	1.12	2.07	-0.22	2.25	0.47	-0.45	1.65	1.29	2.68	1.31
La	-3.95	4.06	-4.45	1.10	-3.93	-5.42	-2.24	-6.64	2.56	-0.07
Ce	-9.45	7.11	-11.39	0.53	-7.92	-11.96	-4.84	-14.23	5.17	-1.21
Pr	-1.06	0.97	-1.47	-0.04	-0.90	-1.42	-0.55	-1.69	0.70	-0.13
Nd	-4.77	4.10	-6.60	-0.12	-3.22	-5.94	-2.50	-7.27	2.52	-0.42
Sm	-0.96	0.87	-2.00	-0.41	-0.73	-1.05	-0.53	-1.37	1.16	0.26
Eu	-0.36	-0.38	-0.67	-0.27	0.04	-0.52	0.31	0.02	-0.34	-0.07
Tb	0.04	0.10	-0.34	-0.19	-0.21	-0.17	-0.25	0.01	0.40	0.15
Dy	0.41	0.86	-2.26	-0.70	-1.06	-0.77	-0.90	0.24	3.19	1.19
Но	-0.01	0.16	-0.48	-0.16	-0.23	-0.20	-0.29	0.14	0.62	0.17
Er	-0.19	0.24	-1.36	-0.24	-0.75	-0.45	-0.76	0.59	2.11	0.53
Tm	0.00	0.04	-0.18	-0.01	-0.07	-0.06	-0.11	0.13	0.33	0.05
Yb	-0.36	0.27	-1.27	0.09	-0.58	-0.53	-0.57	0.80	2.08	0.35
Lu	-0.06	-0.01	-0.16	0.02	-0.12	-0.04	-0.03	0.18	0.36	0.04
Hf	-0.42	0.46	0.03	0.30	0.64	0.90	1.03	0.88	0.43	0.31
Та	0.04	0.06	-0.06	0.05	0.01	-0.02	0.06	0.07	0.07	0.06
Th	0.29	0.90	-0.90	1.24	0.23	-0.17	0.74	0.36	0.17	0.13

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

#### Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration
Sample ID	H501214	H501215	H501216	H501217	H501218	H501219	H501220	H501221	H501222	H501223
Hole ID	BD10-40	BD10-40	BD10-40	BD10-98	BD10-98	BD10-98	BD10-98	BD10-98	BD10-50	BD10-50
Depth (m)	30.5	43.7	16.4	13.7	23	40.4	42.8	50.8	9	15.6
Lithology	HBX	FLT	HBX	FLT	FLT	FLT	FFL	FLT	HFL	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl-Ser	Chl	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser
SiO <sub>2</sub>	-14.61	6.45	7.71	-61.91	-25.92	-3.60	-8.25	6.70	-25.34	9.11
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	1.83	5.51	0.58	12.76	2.50	8.07	1.32	2.34	9.25	5.11
MnO	0.03	-0.02	0.04	0.05	-0.02	0.05	-0.04	0.01	-0.04	-0.08
MgO	0.19	-0.20	-0.58	8.36	4.04	8.68	-0.25	0.04	1.87	-2.17
CaO	-0.04	-0.09	0.14	-0.10	0.09	-0.09	-0.11	-0.12	-0.12	-0.07
Na <sub>2</sub> O	-1.60	-1.57	-1.77	-1.62	-1.81	-1.96	-1.81	-1.82	-1.83	-1.76
K <sub>2</sub> O	0.77	0.75	0.89	-1.88	-0.55	-1.78	0.71	0.68	-0.12	0.82
TiO <sub>2</sub>	0.01	0.03	0.01	0.02	0.05	0.05	0.02	0.02	0.00	0.04
P <sub>2</sub> O <sub>5</sub>	0.01	0.00	0.02	0.00	0.01	0.01	0.02	0.00	0.00	0.01
Net Oxide Change	-13.40	10.86	7.04	-44.34	-21.61	9.44	-8.40	7.85	-16.32	11.01
Ва	-190.39	-439.82	-14.35	-1140.48	823.63	-1170.62	-66.22	-66.99	-68.96	-22.62
Sr	0.11	-2.22	-9.09	1.26	-0.16	-16.12	-13.64	-14.29	-10.85	-3.42
Y	6.98	1.73	5.12	8.89	-1.79	5.65	7.46	1.77	3.75	5.13
Sc	0.61	0.47	1.84	1.63	6.18	1.75	3.01	2.12	2.08	2.77
Zr	14.16	8.60	18.58	26.05	-9.11	25.39	17.61	-1.75	1.82	19.90
Be	-0.12	-0.42	0.08	-0.66	-0.60	-0.41	-0.07	0.10	-0.54	-0.42
V	2.81	9.14	5.03	5.03	31.47	12.78	4.00	6.32	-0.18	17.24
Hg	4.80	14.76	-2.31	34.53	4.18	-4.95	-4.57	-9.25	297.44	14.71
Cr 52	-0.52	0.73	10.22	-1.41	-0.85	6.23	-0.32	0.45	-0.31	8.95
Co	1.07	10.01	-0.63	10.12	-0.35	8.29	1.78	3.75	4.50	1.20
Ni	-0.58	0.82	0.38	-1.58	-0.96	0.88	-0.35	0.51	-0.35	0.81
Cu	-38.96	237.23	-59.69	52.76	-75.80	-37.73	133.89	355.22	17.91	1271.84
Zn	53.85	-0.77	-10.13	608.41	41.88	86.89	-89.62	-86.52	-10.05	-68.97
As	3.68	18.88	11.30	52.08	-1.47	1.04	8.01	1.82	14.50	10.21
Se 78	-1.55	2.19	1.02	-4.22	-2.55	2.34	-0.95	1.36	-0.94	2.15
Br /9	39.67	97.84	89.68	-11.26	0.30	120.43	43.96	/5.68	-43.40	-10.50
M0	2.25	7.20	5.88	1.62	5.58	5.23	3.20	5.44	0.53	3.10
Ag 107	-0.06	0.08	0.04	-0.10	-0.10	0.09	-0.04	0.05	-0.04	0.08
Cu Sn	-0.03	0.07	2.14	2.24	-0.09	1.08	-0.03	0.05	-0.03	0.07
Sh	-0.77	0.02	-2.14	-5.24	-0.93	-1.64	-1.52	-0.87	-1.40	-0.20
Те	0.70	4.75	0.41	0.00	-0.20	0.27	0.15	0.22	0.15	0.35
I	-0.25	3.03	1 41	-5.84	-3.54	3 24	-0.13	1.89	-1.31	2.98
W	2.87	9.84	4 65	2 79	3.97	0.92	2 40	2.76	0.27	11.31
Ph	-1 47	26.07	0.96	-4 00	-2.42	2.22	13 53	1 29	-0.90	2.04
Bi	0.25	9.38	0.33	2.45	0.08	1.56	0.64	0.16	4.65	4.05
Nb	2.37	3.74	4.51	4.57	2.60	4.71	4.11	2.52	1.46	1.50
La	-3.12	-2.35	0.64	-1.72	-4.95	1.42	0.31	0.22	-2.50	0.89
Ce	-6.83	-5.75	0.80	-5.96	-11.28	2.85	0.91	-1.41	-5.92	-0.74
Pr	-0.79	-0.77	0.13	-0.80	-1.35	0.44	0.16	-0.20	-0.52	0.00
Nd	-3.41	-2.36	0.86	-2.70	-5.45	2.83	0.62	-1.48	-1.90	0.45
Sm	-0.68	-0.71	0.27	-0.51	-1.18	0.89	0.16	-0.67	-0.33	0.12
Eu	-0.34	-0.53	-0.18	-0.38	0.41	-0.21	-0.19	-0.04	-0.15	-0.39
Tb	0.01	-0.07	0.03	0.14	-0.13	0.14	0.15	-0.03	0.08	0.04
Dy	0.54	-0.01	0.49	1.29	-0.46	0.95	1.19	-0.34	0.71	0.34
Но	0.01	-0.10	0.06	0.20	-0.19	0.19	0.17	-0.14	0.02	-0.04
Er	0.04	-0.26	0.13	0.60	-0.40	0.38	0.60	-0.36	0.16	-0.02
Tm	0.02	-0.02	0.05	0.09	-0.06	0.07	0.07	-0.02	0.01	0.02
Yb	0.06	-0.24	0.25	0.58	-0.67	0.42	0.45	-0.37	-0.24	-0.04
Lu	0.03	-0.08	0.01	0.08	-0.12	0.05	0.07	-0.07	-0.01	0.00
Ht	-0.44	0.54	2.45	0.15	0.48	0.88	1.01	0.62	1.12	0.34
Ta	0.06	0.10	0.14	0.15	0.07	0.13	0.12	0.08	0.04	0.04
Th	-0.14	3.58	2.77	2.78	1.01	3.39	1.93	1.12	0.33	0.37

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501224	H501225	H501226	H501227	H501228	H501229	H501230	H501231	H501232	H501233
Hole ID	BD10-50	BD10-56	BD10-56	BD10-56	BD10-56	BD10-56	BD10-49	BD10-49	BD10-49	BD10-49
Denth (m)	30.6	74	21.2	29.9	41.4	49.2	10.9	11.8	25	34.6
Lithology	FLT	FLT	FLT	FLT	FLT	FLT	FLT	FFL	FLT	FLT
Alteration	Otz-Ser	Otz-Ser	Chl	Chl	Chl-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiO	4 88	13.01	53.11	31.00	18 31	13.48	8.60	7.40	6.13	0.37
3102	-4.00	-13.01	-33.11	-31.09	18.51	13.46	8.00	7.40	-0.13	-9.37
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	2.23	-0.10	3.40	24.37	10.57	13.38	4.17	1.29	3.07	6.04
MnO	-0.01	-0.07	0.13	0.03	-0.03	-0.08	-0.07	-0.06	0.05	-0.02
MgO	0.16	0.07	15.14	11.09	-0.57	-1.78	-1.65	-1.06	1.14	-0.35
CaO	-0.09	-0.10	0.78	0.04	-0.09	-0.08	-0.10	-0.10	-0.03	-0.04
Na <sub>2</sub> O	-1.80	-1.78	-2.02	-1.99	-1.76	-1.76	-1.78	-1.80	-1.83	-1.79
K <sub>2</sub> O	0.45	0.45	-2.34	-2.20	0.55	0.95	0.92	0.73	0.29	0.56
TiO <sub>2</sub>	0.04	0.01	0.03	0.04	0.04	0.04	0.04	0.02	0.19	0.03
$P_2O_5$	0.02	0.01	0.06	0.01	0.00	0.00	0.00	0.00	0.05	0.08
Net Oxide Change	-3.88	-14.52	-37.93	0.30	27.01	24.16	10.14	6.41	-3.20	-4.87
Ba	-390.00	615.52	-1388.36	-1321.98	-72.13	-73.86	68.80	-70.43	-466.20	-113.04
Sr	-11.18	-4.07	2.88	-17.01	-10.99	-12.37	-10.57	-11.23	-11.01	-10.87
Y	2.31	-3.14	12.07	5.88	6.40	6.39	6.72	-0.23	-3.05	1.47
Sc	1.80	1.95	0.48	2.47	2.38	2.38	2.57	1.85	6.98	2.14
Zr	-6.32	-11.81	-10.77	10.83	-1.92	11.80	11.60	6.85	-27.15	5.74
Be	-0.02	-0.15	-0.67	-0.38	-0.31	-0.31	-0.43	-0.46	-0.50	-0.49
V	3.40	11.16	5.36	0.62	0.94	0.94	4.36	0.19	68.41	3.57
Hg	-9.54	6.78	-1.52	93.97	131.08	27.89	7.43	-9.31	-1.01	10.26
Cr 52	-0.07	-0.65	-1.54	1.10	1.67	1.67	10.34	0.34	21.42	7.76
Со	2.44	-0.73	-0.27	57.77	0.43	1.96	7.21	2.75	8.27	3.57
Ni	-0.08	-0.73	-1.73	1.23	1.88	1.88	0.72	0.38	15.38	0.06
Cu	-50.29	-63.88	-55.51	591.79	4282.46	3287.11	5525.18	185.41	100.64	310.33
Zn	-67.89	-58.70	160.30	7.92	55.82	-34.25	-31.27	-66.07	-55.97	-33.36
As	7.20	1.59	-0.83	55.25	63.72	42.46	4.04	3.30	17.93	25.91
Se 78	-0.22	-1.95	-4.61	49.23	5.02	5.01	1.91	1.03	-0.02	0.16
Br 79	-31.61	-41.63	-83.38	12.90	6.60	7.93	10.15	0.12	-17.81	-14.30
Mo	4.88	0.36	2.15	37.82	6.12	7.22	9.52	6.40	5.56	7.95
Ag 107	-0.01	-0.07	-0.17	0.12	0.19	0.19	1.11	0.04	0.00	0.01
Cd	-0.01	-0.07	-0.16	0.11	0.17	0.17	0.06	0.03	0.00	0.01
Sn	-1.43	-0.47	-3.14	0.55	0.61	1.17	1.83	0.92	1.50	3.42
Sb	0.84	0.12	5.26	10.46	6.43	1.28	0.06	-0.34	2.28	5.56
Te	-0.04	-0.31	-0.74	5.69	4.12	0.81	3.01	0.17	2.75	4.66
Ι	-0.30	-2.70	-6.39	4.56	6.95	6.94	2.65	1.42	-0.02	0.22
W	0.39	1.90	-0.55	2.82	4.60	3.73	3.52	2.43	5.00	2.17
Pb	-0.21	-1.85	-4.37	3.12	4.76	4.75	1.81	0.97	-0.02	49.43
Bi	0.49	-0.11	-0.02	17.96	11.08	6.38	11.70	3.63	3.39	8.17
Nb	1.31	1.20	1.44	9.59	2.16	0.79	-0.31	-0.84	-1.08	-0.77
La	-0.34	-0.36	0.84	5.09	-3.81	-1.27	-5.37	-6.54	-3.52	-6.19
Ce	-1.83	-2.47	0.21	9.76	-9.03	-3.31	-11.10	-14.59	-7.90	-14.22
Pr	-0.14	-0.20	0.03	1.30	-1.10	-0.39	-1.36	-1.73	-0.89	-1.73
Nd	-0.52	-0.28	0.72	6.58	-4.61	-0.95	-5.59	-7.47	-4.01	-7.07
Sm	0.14	0.12	0.22	1.68	-0.21	0.63	-0.76	-1.63	-0.79	-1.46
Eu	-0.08	0.12	0.31	-0.01	0.09	0.27	-0.31	-0.72	-0.74	0.18
Tb	0.07	-0.06	0.25	0.34	0.16	0.17	-0.15	-0.25	-0.22	-0.20
Dy	0.68	-0.33	1.92	2.64	1.38	1.19	-0.74	-1.49	-1.36	-1.07
Но	0.06	-0.14	0.39	0.50	0.12	0.02	-0.35	-0.36	-0.34	-0.28
Er	0.13	-0.34	1.03	0.08	0.16	-0.34	-0.91	-0.57	-0.81	0.03
Tm	0.04	-0.04	0.13	0.26	0.09	0.04	-0.17	-0.15	-0.12	-0.11
Yb	-0.06	-0.45	0.60	1.71	0.37	0.06	-1.15	-1.14	-0.94	-0.92
Lu	0.00	-0.04	0.10	0.27	0.18	0.10	-0.11	-0.18	-0.12	-0.14
Hf	0.51	0.35	-0.47	3.35	0.57	0.53	-0.43	-1.81	-1.54	-1.94
Та	0.05	0.03	0.03	0.29	0.07	0.04	-0.02	-0.05	-0.04	-0.06
Th	0.12	-0.08	0.39	4.58	0.24	-0.25	-1.20	-1.59	-1.16	-1.75

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

#### Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501234	H501235	H501236	H501237	, H501238	H501239	H501240	H501241	H501242	H501243
Hole ID	BD10-34	BD10-34	BD10-34	BD10-34	BD10-33	BD10-33	BD10-33	BD10-33	BD10-33	BD10-45
Depth (m)	11.5	26	33.6	18.6	9.5	18.3	27.1	36.3	46.2	81
Lithology	HEI	UPV	55.0 FLT	HEI	UEI	FI T	UPV	FIT		UEI
Alteration	Otz Ser	Otz Ser	Otz Ser	Ota Ser	Chl Ser	Otz Ser	Otz Ser	Otz Ser	Otz Ser	Otz Ser
Alteration	Qiz-Sei	Q12-301	Q12-Sei	2.42	20.00	Q12-3ei	Q12-3ei	Q12-301	Q12-Sei	Q12-361
SIO <sub>2</sub>	-16.28	-4.36	13.61	3.43	-29.60	2.18	-9.17	-15.45	19.03	-37.35
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	3.70	-0.38	1.05	1.98	1.12	0.93	0.87	3.74	3.20	1.71
MnO	0.01	-0.04	0.00	0.10	0.15	-0.02	0.03	0.00	0.01	0.03
MgO	-0.83	-0.88	-0.66	0.92	1.63	-0.56	-0.21	-0.77	0.55	0.13
CaO	-0.09	-0.09	0.29	-0.09	-0.03	-0.09	-0.01	0.27	-0.09	-0.10
Na <sub>2</sub> O	-1.84	-1.82	-1.82	-1.82	-1.89	-1.86	-1.86	-1.81	-1.83	-1.79
K <sub>2</sub> O	0.87	0.96	1.02	0.62	0.70	0.78	0.93	0.74	0.55	0.51
TiO <sub>2</sub>	0.01	0.01	0.04	0.01	0.00	0.05	0.00	0.15	0.02	0.00
P <sub>2</sub> O <sub>5</sub>	0.01	0.01	0.01	0.00	0.01	0.01	0.02	0.05	0.01	0.01
Net Oxide Change	-14 45	-6.59	13.54	5.14	-27 92	1 42	-9 41	-13.09	21.46	-36.84
Ba	-293.23	-281.50	-511.90	-539.30	-577.18	9.66	-119.33	1300.68	-509.14	195 42
Sr	-15 75	-10.84	-7.10	-12.36	-16.14	-13.81	-12.85	-5.68	-13 27	-9.03
Y	3.84	3 79	3 40	1.70	4.19	-1.13	34.17	-1.83	4 63	-5.27
Sc	-0.18	-0.70	2.60	-0.36	0.25	2.28	1.89	4 50	2.47	-0.27
Zr	4.05	1.97	9.70	-0.39	-6.65	-7.84	17.80	-28.06	3 36	-4.90
Be	-0.11	-0.07	0.15	0.07	-0.27	0.02	-0.09	-0.09	-0.38	-0.35
V	-0.27	4.01	5.52	0.18	-0.67	13.88	-0.21	52.80	0.50	-0.88
ν Hσ	764.50	8.46	120.93	141 17	-4.68	105 71	5 38	7.94	57.81	-6.81
Cr 52	4 23	5.90	0.65	5.87	-1.00	0.10	-0.38	20.60	1 10	-1.56
Cr 52	0.71	0.69	0.05	0.63	0.78	0.10	-0.58	20.00	0.55	-1.50
Ni	-0.71	-0.09	0.30	-0.05	-0.78	-0.05	-0.70	4.34	-0.35	-0.82
Cu	220.24	-0.35	1620 50	27.74	-1.54	101.65	-0.45	4.22	264.06	-1.70
Cu Zn	220.20	-74.00	1122.55	37.74	-00.07	2005.01	-04.04	40.17	169 52	-77.30
As	31.72	18.64	7.26	1 26	-32.40	4.32	0.01	-40.17	12.14	-45.04
A5 So 79	1 42	0.02	1.05	0.05	-0.02	4.32	1.14	1 25	2 20	-0.80
Dr 70	-1.43	-0.95	2.07	0.93	-3.38	28 70	-1.14	-1.23	1.29	-4.09
DI /9	2.07	79.24	-2.07	96.04	-00.89	-28.70	-00.17	-39.32	1.30	-03.07
MO A 107	2.97	/.84	0.85	1.54	0.77	0.01	4.13	2.31	1.84	1.62
Ag IU/	-0.05	-0.03	1.20	0.04	-0.13	2.20	-0.04	1.23	0.12	-0.18
Ca	32.22	0.63	3.97	4.38	-0.12	5.95	0.01	0.43	1.72	-0.16
Sn	-0.22	-1.27	1.31	-0.60	-2.15	4.71	-1.38	-0.36	-0.54	-2.31
50	2.09	17.21	1.51	0.74	0.26	7.22	0.38	1.06	3.47	-0.02
le	-0.23	-0.15	0.31	0.15	-0.58	2.33	-0.18	-0.20	0.53	-0.76
l	-1.99	-1.29	2.70	1.32	-4.96	0.44	-1.58	-1./3	4.56	-6.50
W	1.08	0.93	5.81	0.95	0.22	2.97	1.42	5.62	0.95	-0.59
Pb	49.67	-0.88	3/7.69	0.90	-3.39	2446.52	-1.08	44.39	181.75	-4.45
Bi	0.60	-0.11	1.51	-0.12	-0.11	0.13	0.18	1.40	1.08	-0.11
Nb	-0.57	1.32	1.97	-0.29	-1.62	3.65	0.74	1.25	2.56	-2.06
La	-3.82	-2.20	0.77	-2.88	-8.11	0.96	-3.91	-1.59	0.26	-8.72
Ce	-8.38	-5.40	2.00	-6.07	-18.22	1.61	-9.18	-3.81	0.93	-19.30
Pr	-0.92	-0.68	0.34	-0.67	-2.24	0.20	-1.08	-0.40	0.19	-2.39
Nd	-3.99	-2.67	2.69	-2.84	-8.77	0.73	-3.93	-0.93	1.40	-9.87
Sm	-0.78	-0.40	0.66	-0.37	-2.07	1.72	-1.06	0.03	0.40	-2.41
Eu	-0.41	-0.10	0.05	-0.14	-0.51	0.52	-0.13	-0.13	-0.07	-0.60
1b	-0.14	0.03	0.16	-0.07	-0.33	0.53	-0.16	0.25	0.18	-0.47
Dy	-0.63	0.79	1.58	-0.30	-2.01	3.75	-0.88	2.23	1.73	-2.93
Но	-0.17	0.18	0.39	-0.11	-0.43	0.55	-0.18	0.69	0.48	-0.64
Er	0.35	0.26	-0.05	0.08	-0.08	-0.03	2.71	-0.34	0.44	-0.55
Tm	-0.06	0.08	0.25	-0.05	-0.20	0.28	-0.06	0.43	0.21	-0.27
Yb	-0.55	0.48	1.27	-0.50	-1.48	1.83	-0.73	2.84	1.11	-2.17
Lu	-0.11	0.02	0.18	-0.08	-0.26	0.39	-0.12	0.45	0.15	-0.33
Hf	-0.92	-0.80	0.42	-0.85	-1.79	1.41	-0.24	0.25	0.36	-2.64
Та	-0.04	0.03	0.08	-0.01	-0.06	0.18	0.02	0.03	0.06	-0.09
Th	-0.98	0.11	1.02	-0.65	-1.89	1.25	-0.79	0.02	0.79	-2.50

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501244	H501245	H501246	H501247	H501248	H501249	H501250	H501251	H501252	H501253
Hole ID	BD10-45	BD10-45	BD10-45	BD10-28	BD10-28	BD10-28	BD11-127	BD11-127	BD11-127	BD11-127
Depth (m)	18.4	31.4	46.7	6	20.4	34.3	6.7	16.4	21.1	37.9
Lithology	HFL	FLT	FBX	FLT	FLT	FLT	HBX	HBX	FLT	FLT
Alteration	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiO2	9 46	-3.22	-8 32	15.98	-11.50	7 32	913	-6.47	6.56	-2.95
AO.	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
A <sub>12</sub> O <sub>3</sub>	2.11	2.00	1.52	2.00	0.00	4.01	0.00	2.44	2.26	2.21
FeO Ma	3.11	5.84	1.55	2.99	0.27	4.01	0.60	2.44	2.30	2.21
Mil	-0.02	-0.03	0.00	0.13	0.02	0.25	0.04	0.01	0.02	0.08
MgO	-0.90	-1.09	0.03	2.26	-1.27	2.26	0.25	-0.21	-0.41	1.49
CaO	-0.09	-0.06	-0.11	-0.09	1.08	0.99	0.04	-0.10	-0.10	-0.08
Na <sub>2</sub> O	-1.76	-1.76	-1.82	-1.94	-1.90	-1.91	-1.74	-1.72	-1.72	-1.63
K <sub>2</sub> O	0.49	0.94	0.54	0.50	1.41	0.51	0.90	0.82	0.91	0.47
TiO <sub>2</sub>	0.00	0.03	0.02	0.03	0.03	0.13	0.00	0.00	0.04	0.04
$P_2O_5$	0.02	0.02	0.02	0.00	0.00	0.03	0.00	0.00	0.00	0.01
Net Oxide Change	10.32	-1.33	-8.10	19.86	-11.87	13.59	9.24	-5.22	7.67	-0.36
Ba	2659.51	-239.30	-501.74	-1017.34	-875.81	-922.31	-460.07	-514.62	-460.60	-694.04
Sr	-7.25	-7.65	-14.53	-17.09	-8.74	-6.71	-9.92	-8.39	-8.80	-4.69
Y	-9.51	-1.07	3.04	6.45	3.43	-6.07	0.54	4.75	-2.69	5.81
Sc	0.21	2.30	2.15	1.06	2.38	6.46	-0.11	0.69	2.10	3.24
Zr	0.19	-3.53	-1.12	3.65	5.29	-17.99	-2.21	5.68	5.75	0.64
Be	0.13	0.03	-0.07	0.23	-0.12	-0.41	0.10	-0.03	0.10	0.02
V	0.34	4.68	-0.16	0.57	-0.29	63.36	0.25	-0.07	9.60	5.65
Hg	-6.33	5.43	-9.66	5.20	-2.28	31.51	85.76	68.69	5.61	5.31
Cr 52	0.60	0.11	-0.29	1.02	-0.52	17.18	0.44	7.61	6.71	0.08
Co	0.51	1.57	-0.69	-0.56	-0.71	8.11	-0.62	0.06	3.13	0.61
Ni	0.67	0.13	-0.33	1.14	-0.58	10.56	0.49	-0.14	0.50	0.09
Cu	-72.75	1393.74	242.46	-71.87	-75.09	-72.36	23.73	339.91	45.48	-73.84
Zn	-51.97	-83.17	-86.60	10.61	-64.91	88.80	236.74	538.57	-18.86	15.51
As	1.61	5.46	2.45	3.20	6.36	9.26	3.16	5.70	9.31	2.44
Se 78	1.79	0.34	-0.88	3.05	-1.55	2.35	1.31	-0.37	1.34	0.25
Br 79	-23.99	-31.14	-31.29	138.52	55.47	105.03	74.95	83.49	93.79	68.26
Mo	1.80	6.39	3.35	15.21	3.83	7.20	3.67	7.20	2.16	3.66
Ag 107	0.07	0.01	-0.03	0.11	-0.06	0.09	0.05	-0.01	0.05	0.01
Cd	0.06	0.01	-0.03	0.10	-0.05	0.08	0.96	0.98	0.05	0.01
Sn	0.20	0.87	-2.22	-1.32	-1.85	-1.60	-2.40	-0.95	-1.19	-1.38
Sb	0.29	0.59	0.68	1.05	0.44	0.79	54.66	1.46	2.14	0.63
Те	0.29	0.05	-0.14	0.49	-0.25	0.38	0.21	-0.06	0.22	0.04
Ι	2.48	0.47	-1.22	4.23	-2.15	3.26	1.82	-0.52	1.86	0.34
W	0.72	1.39	2.32	0.09	0.68	0.93	0.62	1.36	1.99	1.34
Pb	1.70	12.27	-0.83	2.89	-1.47	2.23	1.25	18.38	1.27	0.23
Bi	0.85	3.81	0.11	1.71	0.50	1.62	0.37	0.30	2.11	0.30
Nb	5.57	0.52	1.55	3.03	-1.10	3.22	-1.22	-0.28	1.36	-0.39
La	5.65	-2.00	0.23	3.49	-5.51	5.42	-4.87	-4.44	0.45	-2.85
Ce	11.53	-5.65	-0.33	6.56	-12.92	11.75	-10.59	-9.53	-0.53	-5.95
Pr	1.53	-0.65	-0.02	0.63	-1.57	1.50	-1.24	-1.06	0.01	-0.67
Nd	7.25	-2.32	0.42	3.13	-6.55	6.67	-4.84	-4.21	0.49	-3.03
Sm	1.66	-0.72	-0.19	0.19	-1.62	1.53	-1.27	-1.05	-0.02	-0.67
Eu	-0.16	-0.79	-0.49	0.06	0.01	0.63	0.11	-0.71	-0.13	-0.37
Tb	0.42	-0.22	-0.09	0.35	-0.19	0.32	-0.26	-0.13	0.23	-0.08
Dy	3.23	-1.23	-0.51	2.67	-1.12	2.79	-1.30	-0.64	1.94	-0.45
Но	0.71	-0.30	-0.09	0.44	-0.24	0.52	-0.35	-0.28	0.29	-0.25
Er	-0.40	0.11	-0.13	0.56	0.27	-0.67	-0.39	0.32	-0.32	0.11
Tm	0.40	-0.13	-0.01	0.22	-0.13	0.30	-0.11	-0.10	0.12	-0.10
Yb	2.66	-0.91	-0.48	1.40	-1.18	2.04	-1.03	-0.79	0.75	-0.86
Lu	0.36	-0.17	-0.08	0.11	-0.21	0.28	-0.15	-0.15	0.14	-0.14
Hf	1.80	-0.43	-0.91	0.21	-2.64	0.40	-1.46	-1.27	-0.16	-1.15
Та	0.24	0.02	0.03	0.07	-0.06	0.09	-0.05	-0.02	0.04	-0.03
Th	3.52	-0.45	-0.02	1.44	-1.68	1.68	-1.27	-0.68	0.46	-0.61

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Indel   BDII-127   BDII-128   BDII-128   BDII-116   BDII-116   BDII-168   BDII-168   BDII-120   BDII-120     Lthology   FL   HFL   HEL   FLT   FLT   FT   <	Sample ID	H501254	H501255	H501256	H501257	H501258	H501259	H501260	H501261	H501262	H501263
	Hole ID	BD11-127	BD11-128	BD11-128	BD11-128	BD11-116	BD11-116	BD11-116	BD11-116	BD11-116	BD11-120
	Denth (m)	46.5	12.6	15.5	26.5	13.6	16.8	29.2	38.1	49.7	94
	Lithology	FFL	HEL	FLT	FLT	FLT	FT	FLT	FT	FLT	HFL
Anchmin   Quebe   <	Alteration	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser
JAC   JAC <td>SiO</td> <td>5 22</td> <td>2 90</td> <td>7.16</td> <td>16.28</td> <td>10.27</td> <td>0.42</td> <td>11.09</td> <td>26.20</td> <td>1.52</td> <td>48.96</td>	SiO	5 22	2 90	7.16	16.28	10.27	0.42	11.09	26.20	1.52	48.96
Ag, by   0.00   0.01   0.00   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.01   0.00   0.02   0.00   0.00   0.00   0.00   0.00   0.00   0.00   0.00   0.00   0.00   0.00   <	3102	3.33	-2.90	7.10	10.28	-19.27	9.42	11.09	-20.20	-1.32	46.90
FeO   0.90   0.31   1.80   2.10   1.75   2.50   9.97   2.38   2.890   2.13     MgO   0.054   -1.17   0.03   -1.04   -0.02   -0.04   -0.06   -0.04   -0.05   -0.01   -0.06   -0.05   -0.01   -0.09   -0.09   -0.02   -0.01     NagO   -1.64   -1.81   -1.80   -1.81   -1.84   -1.84   -1.84   -1.82   -1.79   -1.82   -1.79   -1.82     KQO   0.03   0.06   0.03   0.05   0.00   0.02   0.01   0.00   0.02   0.00   0.01   0.03   0.021   0.03   0.021   0.06   0.01   0.	$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
MnO   0.02   0.02   0.09   -0.04   -0.04   -0.06   -0.06   -0.06   -0.13   -1.15     CaO   -0.05   0.15   0.39   -0.08   -0.12   -0.10   -0.09   -0.12   -0.10     Na <sub>0</sub> O   1.64   +1.81   -1.80   -1.80   -1.81   -1.81   -1.82     K <sub>2</sub> O   0.97   1.12   0.86   1.09   0.05   0.00   0.02   0.83   0.98   0.88     TiO <sub>2</sub> 0.01   0.04   0.03   0.05   0.00   0.02   0.01   0.00   0.02   0.01   0.00	FeO	0.90	0.31	1.80	2.10	1.75	2.50	9.97	2.83	28.90	2.13
MgO   -0.54   -1.17   0.93   -0.36   0.03   -0.36   -0.37   -0.36   0.036   -0.12   -0.01     Na <sub>0</sub> O   -1.64   -1.81   -1.80   -1.81   -1.84   -1.84   -1.84   -1.82   -1.79   -1.82     K <sub>0</sub> O   0.03   0.00   0.06   0.03   0.02   0.01   0.00   0.00     P <sub>0</sub> O   0.03   0.04   0.04   0.03   0.00   0.02   0.01   0.00   0.00     NetOxide Change   5.03   4.29   9.52   1.62   -19.48   1.63   -12.13   -12.95   20.21   285.02   24.49   48.97     Ba   -4.29.21   -76.180   -80.57   -702.62   713.91   625.96   20.21   -13.46   -12.80   -11.42     Y   3.37   1.33   2.19   0.06   -4.44   1.75   -78.5   -17.9   3.07     Be   0.06   -0.04   0.11   0.18   -0.43   -0.35 <t< td=""><td>MnO</td><td>0.02</td><td>-0.02</td><td>0.09</td><td>-0.04</td><td>-0.04</td><td>-0.02</td><td>-0.04</td><td>-0.06</td><td>-0.04</td><td>-0.03</td></t<>	MnO	0.02	-0.02	0.09	-0.04	-0.04	-0.02	-0.04	-0.06	-0.04	-0.03
	MgO	-0.54	-1.17	0.93	-1.34	-0.79	0.36	0.43	-0.81	-1.93	-1.15
Na <sub>0</sub> O-1.64-1.80-1.80-1.81-1.84-1.84-1.84-1.82-1.82-1.79-1.82K <sub>0</sub> O0.071.120.861.090.760.350.020.010.000.020.01P <sub>0</sub> O,0.010.040.040.030.000.010.000.020.000.00P <sub>0</sub> O,0.010.040.040.030.000.020.040.000.02NatOxide Change5.034.299.5215.26-19.4810.69200.21285.0224.4948.97Ba-429.21-76.180-80.57-702.62713.9162.59200.21285.0228.02-179.58Sr-6.11-9.52-30.8-1.1421.13-12.95-14.21-13.46-12.80-14.12Y3.371.332.190.06-4.441.15-78.55-19.2517.593.07Be0.06-0.040.110.18-0.18-0.13-0.22-0.23-0.24V0.15-0.1017.740.469.440.330.7441.751.331.28Hg-0.35-9.06-6.3725.21.531.4941.071.834.04Cr0.63-0.06-0.571.53-0.890.661.491.122.672.55Cu-0.53-0.569.4149.58-74.57-71.38-75.46-104.011.64	CaO	-0.05	0.15	0.39	-0.08	-0.12	-0.10	-0.09	-0.09	-0.12	-0.01
	Na <sub>2</sub> O	-1.64	-1.81	-1.80	-1.80	-1.81	-1.84	-1.84	-1.82	-1.79	-1.82
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	K <sub>2</sub> O	0.97	1.12	0.86	1.09	0.76	0.35	0.29	0.83	0.98	0.88
Pc0,   0.01   0.04   0.03   0.00   0.01   0.02   0.00   0.00     bet Oxide Change   5.03   4.29   9.52   16.26   -19.48   10.69   19.83   -25.20   24.80   44.89     Ba   -429.21   -761.80   -80.87   -702.62   713.91   625.96   200.21   285.02   382.02   -179.88     Sr   -6.11   -9.52   -3.08   -11.35   -12.13   -12.95   -17.41   -0.65   6.33     Sc   1.65   -0.40   4.47   1.84   2.34   0.19   0.38   3.97   3.80   0.58     Zr   1.173   6.71   7.50   1.57   0.18   -0.48   0.43   -0.33   0.74   41.75   1.33   1.22     Hg   -0.35   -9.60   -1.78   1.24   0.43   0.62   1.33   3.26   2.38   1.581     Co   -0.63   -0.60   1.58   1.00   0.50   -0.66   1.24 <td>TiO<sub>2</sub></td> <td>0.03</td> <td>0.00</td> <td>0.06</td> <td>0.03</td> <td>0.05</td> <td>0.00</td> <td>0.02</td> <td>0.11</td> <td>0.02</td> <td>0.01</td>	TiO <sub>2</sub>	0.03	0.00	0.06	0.03	0.05	0.00	0.02	0.11	0.02	0.01
Net Oxide Change 5.03 4.429 9.52 16.26 -19.48 10.69 9.23 -25.20 24.49 48.97   Sr -6.11 9.52 -3.08 -17.13 -12.13 -12.95 -14.21 -13.46 -12.80 -11.42   Y 3.37 1.33 2.19 0.06 -4.44 1.75 -1.74 -0.65 -6.33   Sc 1.67 0.78 6.71 7.50 13.59 0.63 -1.25 15.95 -19.25 17.59 3.07   Be 0.06 -0.04 0.11 0.18 0.01 0.38 0.32 0.22 -0.23 -0.24   V 0.15 -0.10 17.54 0.46 9.84 0.33 0.74 41.75 1.33 1.28   Ig 0.35 -9.60 0.87 234.17 80.97 16.30 8.77 59.42 11.92 84.84   Cr<2 5.57 -0.18 6.34 26.52 2.95 13.34 1.33 2.62 2.38 15.81   Cr 1.52 5.57.6	$P_2O_5$	0.01	0.04	0.04	0.03	0.00	0.01	0.00	0.02	0.00	0.00
Ba   429.21   -761.80   -80.37   -702.62   713.91   623.96   202.11   285.02   382.02   -179.80     Sr   -6.11   -9.52   -3.08   -11.35   -12.13   -12.95   -14.21   -14.20   -179.86     Y   3.37   1.33   2.19   0.06   -12.13   -12.95   -14.21   -17.44   -0.65   6.33     Sc   1.65   -0.40   4.47   1.84   2.24   0.19   0.38   3.97   3.80   0.58     Zr   11.78   6.71   7.50   3.071   4.13   3.262   -0.23   -0.24     V   0.15   -0.10   17.54   0.44   0.45   0.33   0.74   4.17.5   1.33   1.28     Hg   -0.35   -5.06   0.87   234.17   8.07   1.630   8.77   5.942   1.28   6.77   1.80   4.48   112.2   2.67     Co   0.63   -0.02   1.72.95   57.76   -5.98	Net Oxide Change	5.03	-4.29	9.52	16.26	-19.48	10.69	19.83	-25.20	24.49	48.97
Sr -6.11 -9.52 -3.08 -11.35 -12.13 -12.95 -14.21 -1.346 -12.80 -1.142   Y 3.37 1.33 2.19 0.06 -4.44 1.75 -7.85 -1.74 -0.65 6.33   Sc 1.65 -0.40 4.47 1.84 2.34 0.19 0.38 3.97 3.80 0.58   Zr 11.78 6.71 7.50 13.59 0.63 -1.25 15.95 -19.25 17.59 3.07   Be 0.06 -0.04 0.11 0.18 -0.18 0.33 0.74 41.75 1.33 1.28   Hg -0.35 -9.60 -0.87 234.17 80.97 16.30 8.77 59.42 1192.08 34.84   Cr52 5.57 -0.18 1.00 0.50 -0.66 1.49 11.21 2.67 2.55   Cu 390.96 -12.51 -72.95 -57.76 -28.71 40.18 153.80 644.48 112.6 1185.26   Zn -102.64 -102.58 5.00 <td>Ba</td> <td>-429.21</td> <td>-761.80</td> <td>-803.57</td> <td>-702.62</td> <td>713.91</td> <td>625.96</td> <td>200.21</td> <td>285.02</td> <td>382.02</td> <td>-179.58</td>	Ba	-429.21	-761.80	-803.57	-702.62	713.91	625.96	200.21	285.02	382.02	-179.58
Y $3.37$ $1.33$ $2.19$ $0.06$ $4.44$ $1.75$ $-7.85$ $-1.74$ $-0.65$ $6.33$ Sc $1.65$ $-0.40$ $4.47$ $1.84$ $2.34$ $0.19$ $0.38$ $3.97$ $3.80$ $0.58$ Zr $11.78$ $6.71$ $7.50$ $13.59$ $0.63$ $-1.25$ $15.95$ $19.22$ $-0.23$ $-0.24$ Be $0.06$ $-0.04$ $0.11$ $0.18$ $-0.43$ $0.43$ $0.35$ $-0.22$ $-0.23$ $-0.24$ Y $0.15$ $-0.10$ $17.54$ $0.46$ $9.84$ $0.33$ $0.74$ $41.75$ $1.33$ $1.28$ Hg $-0.55$ $-9.60$ $-0.87$ $234.17$ $80.97$ $16.30$ $8.77$ $59.42$ $1192.08$ $34.84$ Cr S2 $5.57$ $-0.18$ $6.34$ $26.52$ $2.95$ $13.34$ $1.33$ $32.62$ $2.38$ $15.81$ Co $-0.63$ $-0.06$ $1.58$ $1.00$ $0.50$ $-0.60$ $1.49$ $11.21$ $2.67$ $2.55$ Cu $30.906$ $-12.51$ $-72.95$ $5.77.6$ $-71.38$ $75.46$ $-104.01$ $-1.04$ As $2.23$ $6.25$ $5.00$ $9.42$ $2.14$ $4.16$ $17.08$ $3.69$ $242.04$ $37.92$ Se 78 $0.79$ $-0.53$ $1.51$ $24.57$ $1.57$ $3.98$ $2.290$ $7.13$ $6.82$ Br 79 $111.80$ $-54.16$ $-51.10$ $36.84$ $17.51$ $123.19$ $4.93$ $49.90$ </td <td>Sr</td> <td>-6.11</td> <td>-9.52</td> <td>-3.08</td> <td>-11.35</td> <td>-12.13</td> <td>-12.95</td> <td>-14.21</td> <td>-13.46</td> <td>-12.80</td> <td>-11.42</td>	Sr	-6.11	-9.52	-3.08	-11.35	-12.13	-12.95	-14.21	-13.46	-12.80	-11.42
Sc   1.65   -0.40   4.47   1.84   2.34   0.19   0.38   3.97   3.80   0.58     Zr   11.78   6.71   7.50   13.59   0.63   -1.25   -19.25   17.59   3.07     Be   0.06   -0.04   0.11   0.18   -0.43   -0.35   -0.22   -0.23   -0.24     V   0.15   -0.10   17.54   0.46   9.84   0.33   0.74   41.75   1.33   1.28     Cr 52   5.57   -0.18   6.34   26.52   2.95   13.34   1.33   32.62   2.38   15.81     Co   -0.63   -0.06   1.58   1.00   0.50   -0.66   1.49   11.21   2.67   2.55     Cu   390.96   -12.51   -72.95   5.77.6   -28.71   40.18   153.80   44.48   112.62   1185.26     Zr   6.23   6.05   9.42   20.14   41.16   17.08   36.95   24.244   37.92<	Y	3.37	1.33	2.19	0.06	-4.44	1.75	-7.85	-1.74	-0.65	6.33
Žr   11.78   6.71   7.50   13.59   0.63   -1.25   15.95   -19.25   17.59   3.07     Be   0.06   -0.04   0.11   0.18   -0.18   -0.35   -0.22   -0.23   -0.24     V   0.15   -0.10   17.54   0.46   9.84   0.33   0.74   41.75   1.33   1.28     Hg   -0.35   -9.60   -0.87   234.17   80.97   16.30   8.77   59.42   1192.08   34.84     Co   -0.63   -0.06   1.58   10.0   0.50   -6.00   1.24   6.77   18.20   -0.43     Ni   0.30   -0.20   0.57   15.82   -0.89   0.66   1.49   11.21   2.67   2.55     Cu   390.96   -12.51   -72.95   -57.76   2.81   1.40   17.68   36.95   242.04   1.97.99     Se 78   0.79   -0.53   1.51   2.45   -71.38   -75.46   -104.01	Sc	1.65	-0.40	4.47	1.84	2.34	0.19	0.38	3.97	3.80	0.58
Be   0.06   -0.04   0.11   0.18   -0.18   -0.43   -0.35   -0.22   -0.23   -0.24     V   0.15   -0.10   17.54   0.46   9.84   0.33   0.74   41.75   1133   1.28     Hg   -0.35   -9.60   -0.87   234.17   80.97   16.30   8.77   59.42   1192.08   34.84     Co   -0.63   -0.06   1.58   1.00   0.50   -0.60   1.24   6.77   18.20   -0.43     Si   0.30   -0.20   0.57   15.82   -0.89   0.66   1.49   11.21   2.67   2.55     Cu   390.96   -12.51   -72.95   -57.76   -2.871   40.16   17.08   36.95   242.04   35.92     Sa   2.23   6.25   5.00   9.42   20.14   4.16   17.08   36.95   242.04   35.92     Sa 78   0.79   11.18   54.16   51.10   36.44   17.51 <t< td=""><td>Zr</td><td>11.78</td><td>6.71</td><td>7.50</td><td>13.59</td><td>0.63</td><td>-1.25</td><td>15.95</td><td>-19.25</td><td>17.59</td><td>3.07</td></t<>	Zr	11.78	6.71	7.50	13.59	0.63	-1.25	15.95	-19.25	17.59	3.07
$ \begin{array}{c c c c c c c c c c c c c c c c c c c $	Be	0.06	-0.04	0.11	0.18	-0.18	-0.43	-0.35	-0.22	-0.23	-0.24
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	V	0.15	-0.10	17.54	0.46	9.84	0.33	0.74	41.75	1.33	1.28
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Hg	-0.35	-9.60	-0.87	234.17	80.97	16.30	8.77	59.42	1192.08	34.84
$\begin{array}{cccccccccccccccccccccccccccccccccccc$	Cr 52	5.57	-0.18	6.34	26.52	2.95	13.34	1.33	32.62	2.38	15.81
Ni $0.30$ $-0.20$ $0.57$ $15.82$ $-0.89$ $0.66$ $1.49$ $11.21$ $2.67$ $2.55$ Cu $390.96$ $-12.51$ $-72.95$ $-57.76$ $-28.71$ $40.18$ $153.80$ $644.48$ $112.62$ $1185.26$ Zn $-102.64$ $-102.58$ $-50.69$ $1414.95$ $-59.82$ $-7.45$ $-71.38$ $-75.46$ $-104.01$ $-1.04$ As $2.23$ $6.25$ $5.00$ $9.42$ $20.14$ $4.16$ $17.08$ $36.95$ $242.04$ $37.92$ Se 78 $0.79$ $-0.53$ $1.51$ $2.45$ $-2.37$ $1.76$ $3.98$ $-2.99$ $7.13$ $6.82$ Br 79 $111.80$ $-54.16$ $-51.10$ $36.84$ $17.51$ $123.19$ $166.12$ $22.03$ $196.48$ $252.15$ Mo $4.61$ $1.76$ $2.82$ $12.40$ $7.82$ $4.40$ $27.81$ $4.93$ $49.90$ $11.15$ Ag 107 $0.03$ $-0.02$ $0.05$ $4.97$ $-0.08$ $0.66$ $0.13$ $0.30$ $0.24$ $0.23$ Sn $-1.61$ $-2.25$ $-1.45$ $-1.87$ $-0.82$ $0.56$ $0.13$ $-1.71$ $1.61$ $0.69$ Sb $3.21$ $0.68$ $0.76$ $0.96$ $10.10$ $1.53$ $1.48$ $8.33$ $299.83$ $16.09$ Te $0.13$ $-0.09$ $0.24$ $0.39$ $-0.28$ $0.64$ $-0.48$ $27.10$ $1.10$ I $1.10$ $0.74$ $2.99$ $3.12$ $2$	Со	-0.63	-0.06	1.58	1.00	0.50	-0.60	1.24	6.77	18.20	-0.43
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	Ni	0.30	-0.20	0.57	15.82	-0.89	0.66	1.49	11.21	2.67	2.55
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Cu	390.96	-12.51	-72.95	-57.76	-28.71	40.18	153 80	644 48	112.62	1185.26
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Zn	-102.64	-102.58	-50.69	1414.95	-59.82	-7.45	-71.38	-75.46	-104.01	-1.04
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	As	2.23	6.25	5.00	9.42	20.14	4.16	17.08	36.95	242.04	37.92
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	Se 78	0.79	-0.53	1.51	2.45	-2.37	1.76	3.98	-2.99	7.13	6.82
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	Br 79	111.80	-54.16	-51.10	36.84	17.51	123.19	166.12	22.03	196.48	252.15
Ag 1070.03-0.020.060.09-0.090.070.150.909.010.26Cd0.03-0.020.054.97-0.080.060.130.300.240.23Sn-1.61-2.25-1.45-1.87-0.820.560.13-1.711.610.69Sb3.210.680.760.9610.101.531.488.33299.8316.09Te0.13-0.090.240.39-0.380.280.64-0.4827.101.10I1.10-0.742.093.40-3.282.445.51-4.149.889.45W7.211.291.021.202.680.511.345.043.224.13Pb0.75-0.501.432.32-2.2435.983.7713.1856.016.47Bi-0.020.230.301.262.090.162.890.456.4442.62Nb0.775.311.991.50-1.139.097.210.0310.364.36La-1.42-5.672.67-2.51-8.004.120.70-7.3813.06-0.67Ce-4.20-11.094.42-5.91-1.7448.731.38-15.9228.09-2.10Pr-0.35-1.270.45-0.71-2.181.180.29-1.973.61-0.31Sm0.00	Mo	4 61	1.76	2.82	12 40	7 82	4 40	27.81	4 93	49.90	11.15
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Ag 107	0.03	-0.02	0.06	0.09	-0.09	0.07	0.15	0.90	9.01	0.26
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Cd	0.03	-0.02	0.05	4.97	-0.08	0.06	0.13	0.30	0.24	0.23
Sb $3.21$ $0.68$ $0.76$ $0.96$ $10.10$ $1.53$ $1.48$ $8.33$ $299.83$ $16.09$ Te $0.13$ $-0.09$ $0.24$ $0.39$ $-0.38$ $0.28$ $0.64$ $-0.48$ $27.10$ $1.10$ I $1.10$ $-0.74$ $2.09$ $3.40$ $-3.28$ $2.44$ $5.51$ $-4.14$ $9.88$ $9.45$ W $7.21$ $1.29$ $1.02$ $1.20$ $2.68$ $0.51$ $1.34$ $5.04$ $3.22$ $4.13$ Pb $0.75$ $-0.50$ $1.43$ $2.32$ $-2.24$ $35.98$ $3.77$ $13.18$ $56.01$ $6.47$ Bi $-0.02$ $0.23$ $0.30$ $1.26$ $2.09$ $0.16$ $2.89$ $0.45$ $64.44$ $2.62$ Nb $0.77$ $5.31$ $1.99$ $1.50$ $-1.13$ $9.09$ $7.21$ $0.03$ $10.36$ $4.36$ La $-1.42$ $-5.67$ $2.67$ $-2.51$ $-8.00$ $4.12$ $0.70$ $-7.38$ $13.06$ $-0.67$ Ce $-4.20$ $-11.09$ $4.42$ $-5.91$ $-17.44$ $8.73$ $1.38$ $-15.92$ $28.09$ $-2.10$ Pr $-0.35$ $-1.27$ $0.45$ $-0.71$ $-2.18$ $1.18$ $0.29$ $-1.97$ $3.61$ $-0.31$ Nd $-1.16$ $-6.66$ $1.89$ $-2.33$ $-8.88$ $4.60$ $1.25$ $-8.86$ $15.54$ $-1.93$ Sm $0.00$ $-0.44$ $-0.62$ $-0.84$ $-2.43$ $0.98$ $0.27$	Sn	-1.61	-2.25	-1.45	-1.87	-0.82	0.56	0.13	-1.71	1.61	0.69
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Sb	3 21	0.68	0.76	0.96	10.10	1.53	1 48	8.33	299.83	16.09
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Те	0.13	-0.09	0.24	0.39	-0.38	0.28	0.64	-0.48	27.10	1 10
W7.211.291.021.202.680.511.345.043.224.13Pb0.75-0.501.432.32-2.2435.983.7713.1856.016.47Bi-0.020.230.301.262.090.162.890.4564.442.62Nb0.775.311.991.50-1.139.097.210.0310.364.36La-1.42-5.672.67-2.51-8.004.120.70-7.3813.06-0.67Ce-4.20-11.094.42-5.91-17.448.731.38-15.9228.09-2.10Pr-0.35-1.270.45-0.71-2.181.180.29-1.973.61-0.31Nd-1.16-6.061.89-2.33-8.884.601.25-8.8615.54-1.93Sm0.00-1.04-0.62-0.84-2.430.980.27-2.174.28-0.37Eu-0.18-0.31-0.34-0.18-0.470.160.07-0.441.140.17Tb0.20-0.190.080.04-0.390.100.06-0.490.920.02Dy1.44-0.900.960.39-2.381.160.68-3.236.460.43Ho0.16-0.170.07-0.01-0.590.170.08-0.751.23-0.01Er0.08 <td>I</td> <td>1.10</td> <td>-0.74</td> <td>2.09</td> <td>3 40</td> <td>-3.28</td> <td>2 44</td> <td>5.51</td> <td>-4.14</td> <td>9.88</td> <td>9.45</td>	I	1.10	-0.74	2.09	3 40	-3.28	2 44	5.51	-4.14	9.88	9.45
Pb $0.75$ $0.02$ $0.143$ $2.32$ $-2.24$ $35.98$ $3.77$ $13.18$ $56.01$ $6.47$ Bi $-0.02$ $0.23$ $0.30$ $1.26$ $2.09$ $0.16$ $2.89$ $0.45$ $64.44$ $2.62$ Nb $0.77$ $5.31$ $1.99$ $1.50$ $-1.13$ $9.09$ $7.21$ $0.03$ $10.36$ $4.36$ La $-1.42$ $-5.67$ $2.67$ $-2.51$ $-8.00$ $4.12$ $0.70$ $-7.38$ $13.06$ $-0.67$ Ce $-4.20$ $-11.09$ $4.42$ $-5.91$ $-17.44$ $8.73$ $1.38$ $-15.92$ $28.09$ $-2.10$ Pr $-0.35$ $-1.27$ $0.45$ $-0.71$ $-2.18$ $1.18$ $0.29$ $-1.97$ $3.61$ $-0.31$ Nd $-1.16$ $-6.06$ $1.89$ $-2.33$ $-8.88$ $4.60$ $1.25$ $-8.86$ $15.54$ $-1.93$ Sm $0.00$ $-1.04$ $-0.62$ $-0.84$ $-2.43$ $0.98$ $0.27$ $-2.17$ $4.28$ $-0.37$ Eu $-0.18$ $-0.31$ $-0.34$ $-0.18$ $-0.47$ $0.16$ $0.07$ $-0.44$ $1.14$ $0.17$ Tb $0.20$ $-0.19$ $0.08$ $0.04$ $-0.39$ $0.10$ $0.06$ $-0.49$ $0.92$ $0.02$ Dy $1.44$ $-0.90$ $0.96$ $0.39$ $-2.38$ $1.16$ $0.68$ $-3.23$ $6.46$ $0.43$ Ho $0.16$ $-0.17$ $0.07$ $-0.18$ $-0.22$ $-0.23$ <	W	7.21	1.29	1.02	1 20	2.68	0.51	1 34	5.04	3 22	4 13
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Ph	0.75	-0.50	1 43	2 32	-2.24	35.98	3 77	13.18	56.01	6.47
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	Bi	-0.02	0.23	0.30	1.26	2.09	0.16	2.89	0.45	64 44	2.62
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Nh	0.77	5 31	1 99	1 50	-1.13	9.09	7 21	0.03	10.36	4 36
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	La	-1.42	-5.67	2.67	-2.51	-8.00	4.12	0.70	-7.38	13.06	-0.67
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Ce	-4.20	-11.09	4 42	-5.91	-17 44	8.73	1.38	-15.92	28.09	-2.10
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Pr	-0.35	-1 27	0.45	-0.71	-2.18	1.18	0.29	-1.97	3.61	-0.31
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Nd	-1.16	-6.06	1.89	-2.33	-8.88	4.60	1.25	-8.86	15.54	-1.93
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	Sm	0.00	-1.04	-0.62	-0.84	-2.43	0.98	0.27	-2.17	4 28	-0.37
Tb   0.20   -0.19   0.08   0.04   -0.39   0.10   0.06   -0.49   0.92   0.02     Dy   1.44   -0.90   0.96   0.39   -2.38   1.16   0.66   -3.23   6.46   0.43     Ho   0.16   -0.17   0.07   -0.01   -0.59   0.17   0.08   -0.75   1.23   -0.01     Er   0.08   -0.26   -0.22   -0.48   -0.77   -0.31   -0.80   -0.20   0.09   0.17     Tm   0.09   -0.06   0.07   0.02   -0.23   0.15   0.10   -0.35   0.65   0.02     Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Eu	-0.18	-0.31	-0.34	-0.18	-0.47	0.16	0.07	-0.44	1.14	0.17
Dy   1.44   -0.90   0.96   0.39   -2.38   1.16   0.68   -3.23   6.46   0.43     Ho   0.16   -0.17   0.07   -0.01   -0.59   0.17   0.08   -0.75   1.23   -0.01     Er   0.08   -0.26   -0.22   -0.48   -0.77   -0.31   -0.80   -0.20   0.09   0.17     Tm   0.09   -0.06   0.07   0.02   -0.23   0.15   0.10   -0.35   0.65   0.02     Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.01   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Th	0.20	-0.19	0.08	0.04	-0.39	0.10	0.06	-0.49	0.92	0.02
Ho   0.16   -0.17   0.07   -0.01   -0.59   0.17   0.08   -0.75   1.23   -0.01     Er   0.08   -0.26   -0.22   -0.48   -0.77   -0.31   -0.80   -0.20   0.09   0.17     Tm   0.09   -0.06   0.07   0.02   -0.23   0.15   0.10   -0.35   0.65   0.02     Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.01   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Dv	1 44	-0.90	0.96	0.39	-2.38	1.16	0.68	-3.23	6.46	0.43
Er   0.08   -0.26   -0.22   -0.48   -0.77   -0.31   -0.80   -0.20   0.09   0.17     Tm   0.09   -0.06   0.07   0.02   -0.23   0.15   0.10   -0.35   0.65   0.02     Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.01   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Ho	0.16	-0.17	0.07	-0.01	-0.59	0.17	0.08	-0.75	1.23	-0.01
Tm   0.00   -0.06   0.07   0.02   -0.23   0.15   0.10   -0.35   0.65   0.02     Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.01   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Er	0.08	-0.26	-0.22	-0.48	-0.77	-0.31	-0.80	-0.20	0.09	0.17
Yb   0.39   -0.64   0.26   -0.05   -1.78   0.64   0.47   -2.53   4.13   0.07     Lu   0.03   -0.11   -0.01   -0.08   -0.32   0.09   0.08   -0.40   0.67   0.00	Tm	0.09	-0.06	0.07	0.02	-0.23	0.15	0.10	-0.35	0.65	0.02
Lu 0.03 -0.11 -0.01 -0.08 -0.32 0.09 0.08 -0.40 0.67 0.00	Yb	0.39	-0.64	0.26	-0.05	-1.78	0.64	0.47	-2.53	4 13	0.02
	In	0.03	-0.11	-0.01	-0.08	-0.32	0.09	0.08	-0.40	0.67	0.00
Hf $-0.53$ $-0.72$ $-0.24$ $-1.00$ $-1.80$ $1.96$ $1.34$ $-1.65$ $4.64$ $1.03$	Lu Hf	-0.53	-0.72	-0.24	-1 00	-1.80	1.96	1 34	-1.65	4 64	1.03
$T_a = 0.01 - 0.08 - 0.05 - 0.02 - 0.07 - 0.10 - 1.09 - 1.09 - 1.09 - 1.09 - 0.01 - 0.08 - 0.32 - 0.04$	 Тя	0.01	0.08	0.05	0.02	-0.07	0.21	0.14	-0.08	0.32	0.04
The $-0.25$ 1.66 0.60 $-0.19$ $-2.04$ 4.62 3.15 $-0.82$ 530 0.23	Th	-0.25	1.66	0.60	-0.19	-2.04	4.62	3.15	-0.82	5.30	0.23

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia

FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501264	H501265	H501266	H501267	H501268	H501269	H501270	H501271	H501272	H501273
Hole ID	BD11-120	BD11-120	BD10-88	BD10-7	BD10-7	BD10-7	BD10-22	BD10-22	BD10-22	BD10-22
Denth (m)	34 7	42.2	33.5	11.2	17.7	27.8	4.6	13.9	17.9	23.1
Lithology	HBX	HEL	FLT	FLT	FLT	FLT	HBX	FLT	FFL	FLT
Alteration	Otz-Ser	Chl-Ser	Chl	Otz-Ser	Chl	Chl	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiO	11 70	14.24	43.07	10.75	27.03	60.10	17.45	6.07	10.11	2 14
3102	11.79	-14.24	-43.07	-10.73	-27.93	-00.10	-17.43	-0.97	-10.11	-2.14
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	3.70	3.23	10.95	2.92	34.45	12.33	2.66	1.54	0.43	1.95
MnO	0.03	0.06	0.04	-0.09	0.34	0.01	0.08	0.00	0.13	-0.02
MgO	-0.21	0.68	6.68	-1.57	11.63	12.03	4.97	-0.79	0.17	-1.27
CaO	0.09	0.55	-0.08	-0.14	0.12	-0.12	-0.11	-0.11	-0.11	-0.01
Na <sub>2</sub> O	-1.81	-1.80	-1.79	-1.88	-2.02	-2.03	-1.95	-1.89	-1.87	0.14
K <sub>2</sub> O	0.77	0.58	-1.37	1.06	-2.30	-2.36	-0.46	1.04	0.90	0.20
TiO <sub>2</sub>	0.06	0.06	0.06	0.02	0.01	0.03	0.03	0.03	0.04	0.02
$P_2O_5$	0.00	0.02	0.01	0.01	0.13	0.01	0.02	0.02	0.00	0.00
Net Oxide Change	14.43	-10.87	-28.56	-10.41	14.44	-40.20	-12.20	-7.12	-10.43	-1.15
Ва	-115.74	-292.25	-1134.48	209.97	-1387.16	-1402.23	366.68	1350.95	490.91	-85.48
Sr	-10.19	-4.45	3.54	-14.60	-19.02	-21.28	-18.38	-14.45	-12.10	4.21
Y	5.89	-0.21	-4 40	8.18	-1.78	8 61	1.70	-2.18	4 20	-0.70
Sc	2.99	2.01	0.71	2.03	3.40	5.15	0.85	2.27	2.60	2.10
Zr	10.92	8 32	6 59	12.26	-8.02	24 51	1 53	7.96	6.19	-2.87
Be	-0.41	-0.54	-0.59	-0.54	-0.26	-0.64	-0.10	-0.06	-0.10	0.01
V	11.67	3.97	9.03	-0.19	1.22	-0.70	3.83	18 27	3.80	4 56
Ησ	9.26	8 32	-4 59	714 19	41.60	1 70	14 23	1272 11	73 50	1389 14
Cr 52	0.81	7.41	7 37	28.06	43.21	16.94	4.58	-0.25	-0.45	0.04
Ci 52	0.01	3.40	12.38	0.32	510.87	28 50	9.17	0.84	0.71	2.62
Ni	11.32	-0.38	-0.88	14.68	24.17	9.04	-0.48	-0.28	-0.71	0.04
Cu	779 66	-0.58	-0.88	50.52	24.17	120.86	100.48	-0.28	-0.50	700.50
Cu Zn	57 77	321.20	-/3.00	5677 54	2/140.09	129.80	121.01	2013.02	414.95	6627.82
As	110.20	-57.59	7.65	15 14	110 72	142.57	5.06	11.86	1 40	44.18
A5 So 79	2.42	1.02	2.25	14.52	68.06	4.48	1.27	0.74	-1.40	44.18
Se /8 D= 70	2.42	-1.02	-2.55	14.52	08.90	-3.72	-1.27	-0.74	-1.34	0.11
DI /9	4.52	70.08	14.55	42.97	0.95	-23.03	42.30	44.07	25.55	6.49
NIO A. 107	42.80	7.78	14.85	8.15	9.85	10.34	4.91	2.33	2.31	0.48
Ag 107	0.09	-0.04	-0.09	-0.04	3.18	0.24	0.44	7.20	-0.05	0.00
Ca	0.08	-0.03	-0.08	24.79	0.22	-0.13	-0.04	28.97	0.69	22.24
Sn	0.21	-1.4/	-2.73	9.01	5.29	-4.14	0.32	9.76	-1.15	11.97
50	4.00	1.32	-0.11	3.82	1.40	0.21	2.98	11.56	0.93	9.72
le	0.39	-0.16	-0.38	-0.16	21.99	21.03	9.08	-0.12	-0.22	0.02
l	3.35	-1.41	-3.26	-1.39	9.04	-5.16	-1.//	-1.03	-1.85	0.15
W	4.16	3.97	9.89	3.35	9.25	5.55	5.80	1.78	0.73	0.20
Pb	2.29	-0.97	-2.23	40.98	6.19	-3.53	13.06	/16.08	-1.27	1106.88
Bi	11.32	2.72	3.50	18.16	24.42	25.56	13.01	12.59	-0.09	1.79
Nb	2.23	-0.80	2.11	-0.01	0.56	2.22	5.24	4.48	4.82	1.86
La	-6.37	-5.06	-3.01	-1.42	2.77	6.92	0.33	0.33	0.45	-2.02
Ce	-13.11	-11.26	-7.89	-3.22	6.47	11.85	0.22	0.51	0.89	-4.84
Pr	-1.67	-1.32	-1.06	-0.38	1.05	1.43	-0.07	0.00	0.07	-0.55
Nd	-6.79	-5.06	-4.76	-0.55	5.22	6.89	-0.38	-0.33	0.18	-1.76
Sm	-1.46	-1.26	-1.09	-0.67	1.21	0.84	-0.10	0.03	0.01	0.03
Eu	-0.30	-0.20	-0.39	-0.24	-0.13	-0.48	-0.16	0.71	0.23	0.12
Tb	-0.22	-0.22	-0.17	0.12	0.31	0.24	0.05	-0.11	0.01	-0.10
Dy	-1.09	-1.17	-0.76	0.68	1.72	1.50	0.50	-0.53	0.44	-0.34
Но	-0.29	-0.31	-0.23	0.06	0.10	0.19	0.05	-0.15	0.00	-0.18
Er	0.06	0.01	-0.51	0.27	0.17	0.62	0.13	-0.31	0.12	-0.39
Tm	-0.11	-0.12	-0.07	0.04	0.02	0.08	0.03	-0.04	0.02	-0.04
Yb	-0.97	-0.91	-0.58	0.00	-0.21	0.51	0.15	-0.40	0.04	-0.44
Lu	-0.14	-0.19	-0.10	-0.06	-0.07	0.01	0.03	-0.06	0.03	-0.07
Hf	-1.04	-1.65	-0.37	-1.05	-1.25	-0.41	0.46	0.12	0.43	0.30
Та	0.03	-0.06	0.02	-0.04	-0.01	0.00	0.08	0.07	0.06	0.04
Th	-0.42	-1.66	0.28	-0.45	-0.74	0.22	1.28	1.01	0.86	0.37

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

#### Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501274	H501275	H501276	H501277	H501278*	H501279	H501280	H501281	H501282	H501283
Hole ID	BD10-9	BD10-9	BD10-9	BD10-9	BD10-11	BD10-11	BD10-11	BD10-11	BD10-21	BD10-21
Denth (m)	4.1	13.5	20.4	30.1	4	9.8	17.1	28.2	24	15.9
Lithology	UDV	FI T	EDV	FRV	UEI	UEI	FIT	EDV	2.4 HEI	FIT
Alteration	Chl Sar	Ota Sar	Ota Sar	Ota Sar	Ota Sar	Ota Sar	Ota Sar	Ota Sar	Ota Sar	Chl Sor
SiO	12.99	2.60	Q12-301	0.07	0.00	12.57	27.24	12.16	22.06	8.02
SIO <sub>2</sub>	-13.88	3.60	-8.16	-0.07	0.00	-13.57	37.24	-13.16	-32.96	-8.92
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	2.72	-0.13	0.40	0.16	0.00	0.48	0.93	-0.84	-0.52	23.30
MnO	0.11	-0.08	0.04	0.04	0.00	0.03	0.15	-0.01	-0.09	0.13
MgO	4.88	-1.46	-0.77	-0.01	0.00	0.15	-0.81	-0.13	-1.20	3.39
CaO	-0.11	-0.11	-0.01	-0.10	0.00	-0.13	-0.09	-0.09	0.25	-0.03
Na <sub>2</sub> O	-1.96	-1.93	-1.92	-1.77	0.00	-1.94	-1.92	-1.92	0.04	-1.89
K <sub>2</sub> O	-0.20	1.21	1.39	0.97	0.00	0.91	1.38	1.12	-0.13	-0.32
TiO <sub>2</sub>	0.03	0.03	0.02	0.02	0.00	0.01	0.02	0.01	0.02	0.05
$P_2O_5$	0.01	0.01	0.01	0.00	0.00	0.01	0.00	0.00	0.02	0.01
Net Oxide Change	-8.41	1.14	-8.99	-0.77	0.00	-14.05	36.90	-15.00	-34.59	15.72
Ва	57.59	429.47	284.87	-354.52	0.00	305.04	855.14	-221.37	-912.25	-94.24
Sr	-17.28	-13.92	-12.86	-14.05	0.00	-16.81	-15.11	-13.58	67.55	-15.05
Y	6 27	0.40	7.62	3.74	0.00	1.62	0.97	-3.65	8.37	-0.07
Sc	0.37	2.12	1.88	1.93	0.00	0.37	1.03	0.94	2.33	1.12
Zr	-3 53	6.52	7.65	9.05	0.00	5 39	7 99	-8.66	13.98	-1.81
Be	-0.06	0.01	-0.09	-0.01	0.00	-0.14	-0.31	0.68	0.30	-0.30
V	3.16	22 75	5 73	-0.02	0.00	-0.34	0.95	8.44	2.04	26.69
ν Hσ	55.90	279.93	966.06	590.27	0.00	70.97	871.48	174.83	-1.62	5618 41
Cr 52	6 29	0.05	-0.38	-0.03	0.00	-0.60	17.46	-0.70	-1.56	1 74
Ci 52	11.08	0.65	-0.58	-0.05	0.00	-0.00	0.49	-0.70	0.82	111 20
N	0.28	-0.00	-0.13	-0.00	0.00	0.69	1.90	-0.73	-0.82	105
NI Cu	-0.28	206.92	-0.45	-0.03	0.00	-0.08	1.09	-0.79	-1.70	1.95
Cu 7.	85.02	300.83	2018 74	1475.50	0.00	247.95	407.18	-1.99	-77.29	23103.40
	422.03	15 20	4 910.74	14/3.30	0.00	12 72	2703.47	4.02	-/2.01	140.00
AS S. 79	4.30	13.69	4.65	0.98	0.00	12.72	5.05	4.05	-1.00	27.80
Se /8	-0.76	0.14	-1.15	-0.08	0.00	-1.81	5.05	-2.11	-4.09	37.89
BI /9	-27.00	-17.45	-31.20	-12.17	0.00	20.32	138.90	28.11	-37.08	113.40
M0	10.84	1.43	2.34	2.18	0.00	1.76	3.75	2.76	-0.18	14.86
Ag 107	1.74	0.01	-0.04	0.00	0.00	0.73	0.19	-0.08	-0.18	41.68
Ca	-0.03	4.53	11.28	5.72	0.00	0.96	11./1	1.22	-0.16	155.63
Sn	-0.64	10.07	27.66	0.55	0.00	4.30	33.33	4.29	-2.92	33.59
Sb	1.6/	6.30	3.03	0.68	0.00	1.63	5.00	0.23	-0.55	60.45
Te	-0.12	0.02	-0.18	-0.01	0.00	-0.29	0.81	-0.34	-0.75	8.18
l	-1.05	0.19	-1.59	-0.11	0.00	-2.51	7.00	-2.93	-6.50	7.22
W	4.45	2.27	0.18	0.54	0.00	1.71	1.55	1.80	0.10	1.97
Pb	423.44	15.88	-1.09	-0.08	0.00	285.99	95.96	-2.01	-4.45	8399.42
Bi	4.72	0.61	0.52	0.54	0.00	1.47	0.65	1.09	-0.16	57.62
Nb	1.31	0.22	0.51	0.32	0.00	0.05	-0.12	-0.30	3.25	2.59
La	0.00	0.12	1.60	0.42	0.00	-1.95	0.53	-8.30	-2.43	-0.53
Ce	-0.86	-0.48	2.68	0.26	0.00	-4.42	0.78	-17.97	-4.60	-2.56
Pr	-0.06	-0.01	0.30	0.02	0.00	-0.49	0.16	-2.17	-0.49	-0.33
Nd	0.27	0.79	1.80	0.78	0.00	-1.87	0.39	-9.18	-2.03	-1.52
Sm	-0.17	0.04	0.26	0.17	0.00	-0.47	0.17	-2.06	-0.33	-0.25
Eu	-0.28	0.31	0.19	0.07	0.00	-0.01	-0.03	-0.16	0.13	0.49
Tb	0.09	-0.04	0.22	0.10	0.00	-0.02	-0.02	-0.21	0.13	-0.01
Dy	1.01	-0.10	1.74	0.71	0.00	-0.05	0.13	-0.87	1.20	0.04
Но	0.16	-0.05	0.29	0.09	0.00	-0.05	-0.01	-0.20	0.17	-0.02
Er	0.44	-0.16	0.84	0.12	0.00	-0.01	-0.01	-0.59	0.72	0.00
Tm	0.07	0.01	0.12	0.04	0.00	0.00	0.05	-0.08	0.12	0.00
Yb	0.15	-0.07	0.72	0.12	0.00	-0.07	0.11	-0.58	0.73	-0.27
Lu	0.01	-0.05	0.07	-0.01	0.00	-0.01	0.02	-0.13	0.08	-0.02
Hf	-0.08	-0.17	-0.07	0.07	0.00	-0.63	-0.29	-0.73	1.11	1.01
Та	0.00	0.00	0.01	0.03	0.00	-0.01	0.00	-0.03	0.09	0.01
Th	-0.21	-0.02	0.21	0.05	0.00	-0.31	0.12	-1.01	0.80	1.75

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

#### Alteration

Qtz-Ser= Quartz-sericite Chl-Ser= Chlorite-sericite Chl= Chlorite \* Inicates least altered sample

Sample ID	H501284	H501285	H501286	H501287	H501288	H501289	H501290	H501291	H501292	H501293
Hole ID	BD10-21	BD10-21	BD10-26	BD10-26	BD10-26	BD10-26	BD10-26	BD11-146	BD11-146	BD11-146
Depth (m)	30.1	31.1	9.2	18.5	20.6	29.1	37	6.2	11.8	14.9
Lithology	FBX	FBX	FLT	FLT	FLT	FLT	FFL	FLT	FLT	FLT
Alteration	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	41.27	-22.38	27.19	-23.28	-37.87	-1.12	-1.68	12.67	0.12	1.91
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	0.08	-0.03	0.67	-0.54	-1.95	6.93	1.25	0.43	0.67	20.76
MnO	0.25	0.02	-0.09	-0.09	-0.09	-0.07	-0.07	-0.08	-0.02	0.00
MgO	0.77	1.09	-1.44	-1.27	-2.02	-2.07	-0.55	-0.97	-1.09	-0.51
CaO	1.77	0.31	-0.13	-0.13	-0.14	-0.12	-0.12	-0.13	-0.12	-0.14
Na <sub>2</sub> O	-0.92	-1.69	-1.90	-1.91	-1.90	-1.90	-1.91	-1.94	-1.93	-1.93
K <sub>2</sub> O	0.39	0.62	1.36	1.21	1.28	1.16	0.75	1.13	1.25	1.17
TiO <sub>2</sub>	0.02	0.02	0.05	0.02	0.03	0.02	0.02	0.03	0.04	0.02
$P_2O_5$	0.02	0.01	0.02	0.01	-0.01	0.01	0.02	0.00	-0.01	0.00
Net Oxide Change	43.64	-22.03	25.72	-25.97	-42.68	2.84	-2.29	11.15	-1.10	21.28
Ва	-427.86	-283.37	420.94	61.18	33.25	211.04	-327.32	72.66	58.63	97.65
Sr	28.30	-2.52	-14.42	-16.86	-15.36	-16.51	-16.06	-16.42	-17.02	-16.32
Y	4.11	5.42	-3.61	10.98	-29.03	74.06	5.53	-3.29	-1.16	-16.27
Sc	1.50	2.47	2.64	3.96	-3.36	4.27	1.88	1.16	1.95	7.05
Zr	3.03	12.43	40.80	4.20	7.82	18.55	16.31	4.52	9.33	25.45
Be	-0.28	-0.22	-0.37	-0.27	-0.45	-0.45	-0.51	0.12	0.00	-0.29
V	1.09	-0.55	31.63	2.64	1.38	5.18	-0.03	28.74	3.47	24.49
Hg	347.29	64.38	258.49	651.66	12.92	132.89	-5.07	570.44	328.52	1990.79
Cr 52	47.96	22.22	30.89	19.71	10.46	12.48	23.44	0.52	-0.02	1.87
Co	1.39	-0.27	0.10	-0.40	-0.86	5.53	0.16	-0.61	-0.66	-0.47
Ni	21.82	10.67	18.07	10.35	4.19	0.49	13.24	0.58	-0.02	2.10
Cu	37.95	-58.12	291.44	-47.13	-78.18	635.12	-62.27	538.23	8.93	168.49
Zn	1084.09	212.77	1031.41	2897.10	-/6.52	207.98	-20.42	1/96.50	/93.56	2398.16
As	-0.97	9.50	3.72	8.93	-1.68	5/.11	69.33	55.20	89.45	682.11
Se /8	5.84	-2.95	3.52	-3.54	-5.96	1.30	-0.14	1.55	-0.06	5.01
DI /9	10.81	7.34	7 92	-14.90	-07.94	1 00	7.17	/1./4	30.71	2.61
M0 Ag 107	0.22	7.50	0.13	7.58	4.98	4.00	0.01	4.51	0.56	0.21
Ag 107	2.55	-0.11	3 54	9.46	-0.20	0.83	-0.01	7.76	2.38	7.66
Sn	-0.81	-1.67	1.20	83 14	0.09	5.02	10.53	5.86	6.66	8 64
Sh	1 70	0.89	3 34	6.84	-0.29	13 79	4 30	6.10	5.67	37.29
Те	0.94	-0.47	0.57	-0.57	-0.96	0.21	-0.02	0.25	-0.01	0.90
I	8.09	-4.08	4.88	-4.90	-8.26	1.81	-0.19	2.14	-0.08	7.78
W	1.60	1.37	3.72	1.22	2.38	5.95	2.40	2.75	3.42	3.11
Pb	5.53	-2.79	321.95	982.12	-5.65	79.40	-0.13	405.80	539.93	70.26
Bi	0.76	-0.09	4.12	1.43	-0.17	0.55	4.30	2.32	-0.09	-0.03
Nb	0.50	0.34	1.41	-0.29	-1.14	3.02	0.38	-0.02	-0.01	0.46
La	-1.15	-3.99	5.07	0.91	-19.70	8.14	-1.32	1.64	1.74	-3.02
Ce	-2.62	-8.43	8.73	2.36	-42.15	13.19	-3.18	1.63	3.39	-7.46
Pr	-0.22	-1.07	1.07	0.37	-5.30	1.51	-0.43	0.20	0.40	-1.04
Nd	-0.10	-3.66	5.43	1.91	-22.30	6.41	-1.31	1.11	1.56	-4.35
Sm	0.10	-0.80	0.78	0.51	-5.28	1.99	-0.13	0.04	0.11	-1.59
Eu	0.03	0.05	0.67	-0.19	-1.15	1.19	-0.23	0.48	0.25	-0.25
Тb	0.07	0.04	-0.13	0.11	-0.91	0.89	-0.03	-0.07	-0.05	-0.46
Dy	0.71	0.41	-0.88	0.84	-5.61	8.03	0.11	-0.56	-0.05	-2.84
п0 Ба	0.13	0.10	-0.19	0.17	-1.18	1.86	0.00	-0.19	0.00	-0.63
EI Tm	0.27	0.28	-0.51	0.49	-5.41	0.49	0.05	-0.59	0.27	-1./6
Vh	0.05	0.04	-0.01	0.09	-0.48	1.09	-0.12	-0.07	0.10	-0.23
In	0.50	0.00	0.05	0.21	-0.49	1.32	-0.13	-0.00	0.00	-0.27
Hf	0.00	-0.57	0.13	-1.09	0.49	8.45	-0.05	-0.12	-0.25	0.27
 Та	0.02	0.01	0.05	-0.01	-0.03	0.06	0.03	0.00	0.00	-0.01
Th	-0.11	-0.21	0.65	-0.81	-0.75	1.24	-0.17	-0.01	0.23	-0.43

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

#### Alteration

Sample ID	H501294	H501295	H501296	H501297	H501298	H501299	H501300	H501301	H501302	H501303
Hole ID	BD11-143	BD11-143	BD11-143	BD11-143	BD11-134	BD11-134	BD11-134	BD11-134	BD11-154	BD11-154
Depth (m)	4.3	7.9	16.7	28.1	10.9	21.5	33.1	43	10.9	26.1
Lithology	FLT	FLT	FLT	FFL	FLT	FLT	FLT	FBX	FLT	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	43.22	44.07	45.23	19.87	-1.90	-19.08	19.97	13.00	-4.31	13.00
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	8.98	6.29	4.48	-0.24	12.71	8.83	7.55	9.61	1.50	3.56
MnO	-0.09	-0.06	-0.07	-0.09	-0.08	0.10	-0.09	-0.06	-0.09	-0.09
MgO	-1.93	1.01	0.07	-1.00	-0.77	10.67	-2.04	-0.62	-2.10	-1.97
CaO	-0.12	-0.13	-0.12	-0.13	-0.13	1.36	-0.13	-0.01	-0.14	-0.11
Na <sub>2</sub> O	-1.86	-1.91	-1.89	-1.89	-1.87	-2.01	-1.85	-1.88	-1.84	-1.82
K <sub>2</sub> O	1.02	0.12	0.49	0.90	0.86	-2.20	1.14	0.81	1.18	1.10
TiO <sub>2</sub>	0.05	0.01	0.04	0.03	0.03	0.05	0.04	0.03	0.04	0.04
$P_2O_5$	0.00	0.01	0.00	0.01	0.00	0.01	0.00	0.00	-0.01	0.00
Net Oxide Change	49.27	49.41	48.21	17.46	8.85	-2.27	24.59	20.88	-5.77	13.70
Ва	-303.42	-598.38	-487.31	-387.06	-72.06	-1348.32	-399.79	-458.03	134.26	-209.68
Sr	-14.09	-15.77	-15.89	-14.97	-12.28	13.17	-12.75	-10.26	-10.51	-10.27
Y	3.73	6.15	11.39	3.20	4.76	10.02	5.93	6.67	-23.68	-16.72
Sc	2.66	2.46	3.74	2.89	3.37	3.85	3.21	3.05	-2.34	-0.62
Zr	21.16	15.37	24.48	14.41	13.03	16.98	14.89	18.11	8.17	17.98
Be	-0.21	-0.22	-0.24	-0.41	-0.39	-0.47	-0.34	-0.35	-0.52	-0.41
V	1.46	1.39	1.32	0.43	0.54	3.89	0.80	0.76	-0.11	4.54
Hg	6.99	5.13	1.74	-9.07	24.47	5.05	11.78	27.15	-3.38	0.90
Cr 52	42.44	2.48	47.60	31.56	6.60	0.29	1.43	18.28	7.68	9.64
Co	29.95	18.29	20.71	1.81	8.69	3.31	12.45	2.53	0.94	6.83
Ni	20.79	2.78	24.90	18.26	1.08	0.33	1.61	1.52	-0.21	0.86
Cu	4705.37	1417.01	1669.24	-27.85	771.02	65.72	-53.10	871.24	-74.41	-72.40
Zn	-17.09	-22.83	-15.27	-70.39	-11.84	220.62	-115.09	-82.30	-119.38	-111.77
As	26.90	22.99	49.20	0.19	77.78	4.53	43.37	227.01	15.07	33.23
Se /8	/./8	1.44	/.03	2.30	2.88	0.88	4.29	4.07	-0.57	2.30
BI /9	118.05	5.42	181.03	/1.40	140.17	91.85	174.05	152.55	5 40	105.50
M0 Ag 107	0.20	0.28	20.40	0.09	4.90	0.03	0.16	0.15	0.57	4.67
Cd	0.29	0.28	0.20	0.09	0.11	0.03	0.10	0.15	-0.02	0.09
Sn	0.20	0.25	0.24	-0.71	1 73	-2.25	-0.40	0.14	-0.64	-0.03
Sh	1.81	1.16	1 29	0.14	1.56	0.96	1.97	13 59	0.41	0.58
Те	10.38	4 58	1.13	0.37	12.50	0.14	0.69	0.66	-0.09	0.37
I	10.78	10.30	9.75	3.18	3.99	1.21	5.95	5.64	-0.79	3.19
W	9.87	7.96	6.42	6.36	6.52	6.90	3.97	1.49	3.10	3.69
Pb	7.37	7.05	6.67	2.18	76.13	36.52	4.07	29.21	15.27	2.18
Bi	15.38	7.84	6.01	1.83	21.25	1.52	5.08	9.20	3.48	3.60
Nb	1.37	2.56	1.96	0.75	2.95	4.68	1.78	2.44	1.94	2.70
La	-4.22	-0.39	-0.43	-1.54	-4.04	1.00	-0.68	-0.83	-13.16	-16.25
Ce	-9.41	0.79	-1.63	-5.59	-8.01	2.38	-1.42	-1.97	-31.50	-35.18
Pr	-1.23	0.19	-0.19	-0.71	-0.92	0.36	-0.09	-0.26	-4.15	-4.45
Nd	-5.66	0.91	-0.87	-1.77	-3.55	1.40	-0.40	-1.16	-17.95	-18.86
Sm	-1.37	0.18	-0.01	-0.01	-0.48	0.87	0.16	0.06	-4.55	-4.47
Eu	-0.03	0.34	0.15	0.01	0.04	0.39	-0.08	0.42	-0.86	-0.85
Tb	-0.09	0.07	0.21	0.06	0.02	0.24	-0.02	0.04	-0.75	-0.65
Dy	0.15	0.26	1.83	0.25	0.61	1.37	0.14	0.65	-4.45	-3.61
Ho	0.09	0.01	0.35	0.02	0.00	0.27	-0.08	0.07	-0.92	-0.73
Er Ter	0.28	-0.06	1.04	0.21	0.22	0.81	-0.35	0.06	-2.40	-1.86
1m Vh	0.03	0.01	0.18	0.02	0.06	0.12	-0.02	0.04	-0.30	-0.22
10	0.22	-0.13	0.//	0.10	0.10	0.78	-0.29	-0.04	-1.0/	-1.32
LU Uf	0.04	-0.02	0.08	0.07	0.07	0.14	-0.05	0.02	-0.28	-0.20
111 To	0.50	1.04	-0.42	0.35	1.55	1.91	0.57	1.05	-0.33	0.47
Th	0.47	1.56	0.41	0.09	0.83	1.10	0.16	0.57	0.14	0.67

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

#### Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H201304	U501305	H201306	U501307	U501308	U501300	H501310	H501311	U501312	U501313
Sample ID	DD11 154	DD11 159	DD11 159	DD11 159	DD11 159	DD11 126	DD11 126	DD11 126	DD11 105	DD11 105
	BD11-154	BD11-138	BD11-158	BD11-158	BD11-158	BD11-130	BD11-130	BD11-130	BD11-195	BD11-195
Depth (m)	30.5	5.8	14	25.4	39.8	10.7	23.4	31.9	7.9	9.5
Lithology	FLT	FLT	FLT	FLT	FLT	FLT	FLT	FLT	FLT	FFL
Alteration	Qtz-Ser	Chl	Chl	Chl-Ser	Qtz-Ser	Qtz-Ser	Chl	Qtz-Ser	Qtz-Ser	Chl-Ser
SiO <sub>2</sub>	-4.30	-58.79	-10.86	14.21	-1.59	46.88	-61.01	9.62	12.52	-6.42
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	0.57	9.95	9.13	1.47	1.87	10.80	6.26	5.10	5.36	4.00
MnO	-0.09	-0.03	0.19	-0.06	-0.05	-0.09	-0.02	-0.09	-0.09	-0.03
MgO	-1.96	14.74	20.58	3.74	1.45	-1.40	10.91	-1.62	-1.86	1.92
CaO	-0.13	0.00	9.45	-0.07	-0.10	-0.11	-0.14	-0.12	-0.12	-0.10
Na <sub>2</sub> O	-1.84	-2.01	-2.01	-1.96	-1.95	-1.86	-2.03	-1.88	-1.90	-1.94
K-0	1.18	-2.34	-2.33	-0.16	0.33	0.99	-2.36	1.06	1.28	0.18
TiO <sub>2</sub>	0.03	0.03	0.14	0.02	0.03	0.04	0.04	0.03	0.02	0.05
P.O.	-0.01	0.00	0.11	0.00	0.01	0.00	0.00	0.00	0.00	0.02
Nat Oxida Changa	-0.01	38 44	24.38	17.20	0.01	55 25	48.35	12 10	15.22	2 22
Re Oxide Change	-0.55	1204 41	1287.07	667.40	738.80	964.98	1400.47	12.10	256.34	-2.52
Da Sr	-90.44	-1394.41	-1387.97	17.25	-/38.80	10.28	-1400.47	-139.92	-230.34	-015.89
SI V	-9.70	-19.14	22.00	-17.23	-10.91	-10.38	-21.42	-12.02	-13.97	-10.92
1 So	-13.01	-22.33	33.09	1.97	4.60	6.61	-10.02	9.03	2.06	2.20
7-	1.67	1.09	4.09	1.07	0.82	0.01	-0.13	2.89	2.00	2.20
Zr D-	1.07	-1.98	0.22	0.97	9.82	0.17	13.08	7.30	9.30	0.40
DC V	-0.33	-0.04	-0.55	-0.41	-0.49	-0.17	-0.71	-0.41	-0.40	-0.49
V U-	4.08	-0./1	0.84	4.02	2.39	1.05	10.55	0.45	0.51	0.04
пg Cr 52	-5.54	1/9/.21	1 40	4.01	-4.87	182.29	-10.35	5.24	42.27	1.52
CI 52	6.92	-1.27	1.49	0.83	0.08	20.25	-1.0/	5.15	0.57	0.07
C0	0.85	1 42	23.70	0.07	0.00	3.01	18.78	5.15	-0.57	-0.05
NI Cu	-0.50	-1.42	221.02	0.95	50.02	2976.90	-2.10	0.80	2074.56	0.08
Cu Zn	-/4.5/	6075.24	766 51	59.59	02.14	2670.80	-24.09	24.40	64 51	22.03
As	-120.34	72.84	21.68	5.63	92.14	339.09	25.06	-00.81	82.07	20.52
AS S., 79	4.51	2.04	51.00	3.03	43.41	333.80	23.90	9.08	2.75	7.10
Br 70	-0.80	-5.80	4.47	2.49	0.23	0.02	-3.02	2.50	2.73	68.24
DI /9	40.40	0.14	10.20	2 42	2.24	260.40	2.50	2.00	90.30	2.14
M0 A ~ 107	0.24	-0.14	0.17	5.42	5.24	1 1 9	2.30	5.09	0.45	0.01
Ag 107	-0.03	3.02	1.40	0.09	0.01	0.20	0.11	0.09	0.10	0.01
Cu Sn	-0.03	25.42	5.10	12.10	3.70	20.08	-0.19	10.01	0.09	1.77
Sh	0.73	4.30	0.62	0.50	1.52	29.98	2.19	0.24	-0.30	-1.//
50 To	0.72	6.05	-0.02	0.39	0.04	20.29	-0.19	0.34	4.20	2.90
IC	-0.13	5.03	6.20	2.45	0.04	1.42	-0.90	0.57	2.91	0.04
I W	-1.11	-3.27	1.24	5.45	1.99	2.08	-1.19	3.16	2.01	0.51
W Dh	0.76	2.61	1.24	0.08	0.24	2.08	5.00	2.10	10.16	0.21
Di Di	-0.70	-5.01	0.81	2.50	1.77	39.05	-5.55	2.18	2 24	1.25
DI Nib	1.27	0.12	2.04	0.93	0.15	0.20	1.02	2.72	2.34	2.20
IND	2.41	-0.15	2.94	-0.14	0.13	-0.39	6.26	-0.00	2.85	5.80
La Ca	-2.41	-18.30	41.70	-0.27	-0.33	50.92	-0.30	4.55	-1.19	-0.52
Dr	-9.51	-39.40	10.80	-1.48	-1.50	7.10	-15.55	1.20	-3.08	-2.04
ri Nd	-1.34	-4.92	10.89	-0.21	-0.11	21.66	-2.08	6.85	-0.28	-0.24
Sm	-7.48	-20.30	40.55	-0.70	-0.18	7 70	-0.05	1.80	-1.01	-0.33
5m En	-2.40	-4.//	9.93	-0.22	-0.10	1.70	-2.43	1.60	-0.32	-0.20
Eu	-0.34	-1.06	0.41	0.58	0.23	1.03	-0.55	0.06	0.28	0.18
10 Dec	-0.42	-0.78	1.22	-0.02	0.03	1.09	-0.38	1.02	0.00	0.20
Цо	-2.39	-4.39	0.31	0.00	0.54	0.07	-1.99	1.02	0.00	1.02
Fr	-0.59	2 40	2.72	0.02	0.00	1.00	0.70	0.00	0.00	0.40
Tm	-1.30	-2.09	5.25 0.49	-0.01	0.10	2.00	-0.78	0.25	0.52	0.02
1111 Vb	-0.21	-0.58	0.48	0.01	0.05	0.38	-0.05	0.02	0.08	0.15
10	-1.24	-2.45	2.92	-0.01	0.09	2.31	-0.54	-0.1/	0.14	0.62
LU Uf	-0.19	-0.35	0.40	0.00	0.00	0.51	-0.03	-0.01	0.01	0.07
пі Т.	-0.06	-0.29	2.51	-0.45	-0.97	-0.52	0.80	-0.39	0.15	-0./1
1a Th	0.04	-0.03	0.12	0.00	0.01	-0.02	0.05	0.00	0.06	0.11
111	0.40	-1.14	3.11	-0.10	0.03	-0.38	1.21	-0.54	0.75	0.94

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

#### Alteration

Sample ID	H201314	H501315	H501316	H501317	U501318	H501310	H501320	H501326	U501327	U501328
Hole ID	PD11 105	PD11 105	RD10.6	RD10.6	RD10.6	RD10.6	RD10.6	RD10 113	PD10 113	RD10 113
Depth (m)	16.8	22.8	4.6	7 3	13.7	17.0	32.1	5	10	28.2
Lithology	10.8 ELT	25.8 EEI	4.0	7.5 ELT	13.7 ELT	17.5 EDV	52.1 EDV	UEI	ELT	28.5 ELT
Alteration	FLI Ota Sar	Ota Sar	Ota Sar	ChlSar	FL1 Ota Sar	FBA Ota Sar	FBA Ota Sar	Ota Sar	Chl	FLI Ota Sar
Alteration	QIZ-Sei	Q12-Sei	Q12-Sei	CIII-Sei	QIZ-Sei	QIZ-Sei	Q12-Sei	0.20	10.((	QIZ-Sei
S1O <sub>2</sub>	8.07	-2.58	14.94	-6.59	-5.87	5.96	-2.91	9.29	-18.66	-9.84
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	2.90	1.69	0.23	6.00	1.24	0.74	-0.12	-0.61	8.61	-0.19
MnO	-0.08	-0.05	0.02	0.07	0.05	0.05	-0.02	-0.08	0.19	0.06
MgO	-0.83	1.04	1.38	0.79	-0.64	-0.87	-0.36	0.04	9.63	0.06
CaO	-0.12	-0.11	-0.11	-0.12	-0.11	0.08	-0.03	-0.13	-0.12	0.16
Na <sub>2</sub> O	-1.92	-1.94	-1.93	-1.94	-1.91	-1.31	-1.82	-1.93	-2.00	-1.43
K <sub>2</sub> O	0.91	0.59	0.61	0.38	1.08	0.97	1.09	0.89	-1.84	0.66
TiO <sub>2</sub>	0.01	0.03	0.01	0.01	0.03	0.03	0.01	0.00	0.03	0.02
P <sub>2</sub> O <sub>5</sub>	0.00	0.01	0.00	0.00	0.01	0.01	0.01	0.02	0.01	0.02
Net Oxide Change	8.94	-1.33	15.15	-1.41	-6.11	5.66	-4.15	7.49	-4.15	-10.49
Ba	-360.02	-585.02	987.29	449.61	381.41	440.12	-283.17	1243.60	-1048.67	-124.46
Sr	-15.29	-15.98	-16.23	-16 77	-15 34	-6.04	-13 38	-16.66	-20.96	-5.92
Y	1.26	5.16	6.00	-1.35	0.00	-3.89	-12.20	0.54	-0.35	-4.16
Sc	2.30	3.05	0.38	1.51	2.37	0.64	0.53	0.69	0.45	1.61
Zr	-2.96	12 57	1 53	-19.49	1.50	1.83	-0.45	-6.06	-2.94	-1 19
Be	-0.44	-0.50	0.15	0.05	-0.05	0.06	-0.04	0.07	-0.48	-0.11
V	0.30	0.01	0.38	20.51	9.87	0.16	4.20	0.17	12.12	10.90
ν Hσ	26.03	16.10	0.69	1640.60	327.62	321.02	510.69	20.06	106.04	342.60
Cr 52	0.53	0.02	0.68	8 86	-0.22	0.28	6.27	0.31	0.20	-0.48
Ci 52	0.55	0.62	0.11	18.40	-0.22	0.63	0.68	0.63	30.26	1 2 2
Ni	-0.01	-0.00	0.11	0.23	0.03	-0.03	-0.08	-0.05	0.20	0.53
Cu	141.00	72.08	2.51	2272.02	120.50	202.08	-0.21 516.61	164.05	2000 16	-0.33
Cu 7n	60.65	-73.98	-2.51	5611 59	129.39	1201.25	1475.62	168.02	2090.10	1542.92
As	-09.05	-32.32	0.10	100 20	51.06	24.27	11 47	2 26	112 20	20.08
AS S = 79	1.59	2.10	0.19	190.20	51.00	24.27	0.57	2.20	0.50	29.98
Br 70	1.30	52.72	2.05	102 47	-0.03	110.20	-0.37	105.80	0.39	-1.43
DI /9	95.64	32.12	0.09	103.47	100.55	2.22	2.47	105.80	94.92	2.15
NIO A = 107	3.37	3.30	0.08	0.72	4.91	2.33	3.47	1.08	7.90	2.15
Ag IU/	0.06	0.00	1.00	9.75	0.72	0.03	-0.02	0.03	4.10	-0.05
Ca	0.03	0.00	0.07	15.75	3.90	4.60	3.21	0.03	5.12	4.00
Sn	-0.53	-2.36	9.32	8.48	14.12	13.53	3.57	5.80	12.14	13.67
50 T	3.30	1.39	1.44	25.92	11.0/	3.09	0.08	5.40	10.28	2.23
le	0.25	0.01	0.33	0.10	-0.10	0.14	-0.09	0.15	2.67	-0.23
1 W	2.19	0.07	2.85	0.85	-0.90	1.18	-0.79	1.27	0.82	-1.98
W	2.97	1.36	0.81	2.15	1.49	1.61	2.16	1.23	2.32	0.20
Pb	1.50	0.05	135.97	13/9./1	34.90	17.27	-0.54	81.77	191.24	116.94
BI	1.70	0.10	1.94	9.51	1.36	1.48	0.24	0.34	15.04	0.29
Nb	1.02	1.66	-0.20	-0.80	0.08	0.13	-0.30	-0.69	0.73	0.41
La	0.34	-0.52	0.19	-3.13	-0.21	-2.64	-1.74	-0.48	2.26	1.39
Ce	-1.52	-1.40	-1.06	-7.83	-0.93	-5.63	-3.96	-2.13	3.36	2.70
Pr	-0.19	-0.13	-0.01	-0.94	-0.07	-0.69	-0.43	-0.14	0.50	0.39
Nd	-0.38	-0.36	-0.02	-3.8/	-0.26	-2.81	-1.96	-0.07	1.91	1.99
Sm	-0.22	-0.06	0.10	-0.86	-0.14	-0.65	-0.73	-0.05	0.27	0.20
Eu	0.09	0.38	0.17	0.28	0.21	-0.07	-0.23	0.00	0.37	0.35
Ib	-0.08	0.09	0.04	-0.13	-0.03	-0.15	-0.28	-0.02	0.03	-0.04
Dy	-0.34	0.81	0.68	-0.65	0.09	-0.63	-1.83	0.06	0.41	-0.25
Но	-0.09	0.19	0.09	-0.17	-0.02	-0.19	-0.46	0.01	0.05	-0.10
Er	-0.15	0.52	0.42	-0.43	-0.03	-0.58	-1.23	0.10	0.15	-0.20
Tm	-0.02	0.10	0.08	-0.07	0.02	-0.07	-0.16	0.01	0.03	0.00
Yb	-0.30	0.60	0.28	-0.57	-0.04	-0.54	-1.17	-0.02	0.03	-0.11
Lu	-0.05	0.03	0.08	-0.09	0.00	-0.07	-0.15	-0.02	-0.01	-0.02
Hf	0.32	0.59	0.27	-0.76	-0.71	-0.41	-0.92	-0.49	-0.69	0.05
Та	0.02	0.05	0.02	-0.03	0.01	0.01	-0.01	-0.03	0.01	0.01
Th	0.27	0.86	-0.24	-1.22	-0.47	-0.50	-0.79	-0.37	0.08	0.06

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapili tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow

INT= Intrusion

Alteration

Sample ID	H501329	H501332	H501333	H501334	H501335	H501336	H501337	H501338	H501339	H501340
Hole ID	BD10-113	BD10-3	BD10-3	BD10-3	BD10-3	BD10-3	BD10-18	BD10-18	BD10-18	BD10-18
Depth (m)	40.8	6.6	8.2	16.3	30.7	37.9	4.3	9.9	19.7	27.2
Lithology	FLT	HFL	FLT	FLT	FLT	FLT	FLT	FFL	FBX	FFL
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl	Chl-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	13.75	-2.42	10.47	-11.12	-29.46	-53.20	6.31	14.70	15.16	15.71
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	-0.16	0.01	5.89	2.28	7.98	8.12	14.05	3.19	2.63	3.85
MnO	0.21	-0.06	-0.08	-0.08	-0.03	0.81	-0.02	-0.08	-0.09	-0.07
MgO	0.11	-1.36	-1.69	-1.80	-0.88	13.23	0.16	-1.42	-1.96	-1.29
CaO	0.24	-0.14	-0.13	-0.14	-0.13	1.35	-0.10	-0.02	-0.13	-0.12
Na <sub>2</sub> O	-1.90	-1.87	-1.88	-1.88	-1.88	-2.03	-1.94	-1.92	-1.92	-1.92
K <sub>2</sub> O	1.15	1.19	1.24	1.29	1.13	-2.33	0.90	1.29	1.33	1.08
TiO <sub>2</sub>	0.04	0.02	0.05	0.03	0.02	0.00	0.03	0.02	0.03	0.04
$P_2O_5$	0.05	-0.01	0.00	-0.01	-0.01	0.01	0.02	0.01	0.00	0.00
Net Oxide Change	13.49	-4.64	13.87	-11.42	-23.25	-34.04	19.42	15.76	15.05	17.28
Ва	-41.16	1241.52	1631.50	828.11	1.96	-1396.89	-262.79	-174.52	-309.99	-447.42
Sr	-2.73	-14.37	-14.81	-14.73	-13.53	-6.44	-18.00	-16.05	-16.12	-14.73
Y	5.20	2.96	-4.27	3.71	2.52	2.48	9.05	-5.69	11.60	2.19
Sc	2.47	2.40	1.98	1.81	3.56	7.04	3.35	1.89	2.95	2.11
Zr	21.60	1.88	-4.88	5.83	2.44	-16.85	15.17	9.63	10.35	11.26
Be	0.13	-0.05	-0.40	-0.09	-0.15	-0.63	-0.33	-0.41	-0.41	-0.39
V	0.33	-0.12	4.69	2.04	-0.38	5.65	4.17	0.47	0.44	0.53
Hg	206.74	446.75	1736.49	996.57	454.88	32.45	29.38	9.41	-1.41	1.32
Cr 52	0.59	-0.21	6.26	-0.41	-0.68	-1.15	1.49	8.71	0.79	7.29
Co N:	-0.07	-0.68	-0.5/	-0.70	-0.27	16.68	-0.51	1.01	-0.58	-0.56
NI Cu	0.07	-0.23	11.48	-0.46	-0.76	-1.50	1.07	0.95	0.88	1.00
Cu Zn	540.30	1/3.41	-0.04	20.30	55.15 1015 78	30.74 7000.10	4807.80	2 25	070.02	200.71
As	22.76	8 79	125.90	13 23	97.05	-0 20	153.14	68.83	61 11	115 23
Se 78	1 78	-0.62	2.64	-1.22	-2.04	-3.46	4 47	2.53	2.36	2.82
Br 79	130.58	46.72	108 10	31.87	-5.52	-16.84	163.93	151.94	93 47	142.05
Мо	2.20	1.83	3.36	2.38	5.17	3.31	52.18	4.44	2.55	3.11
Ag 107	0.07	2.31	1.44	-0.05	0.36	-0.13	0.17	0.09	1.19	0.11
Cd	1.18	10.29	37.18	20.98	3.91	6.62	0.15	0.09	0.08	0.10
Sn	5.29	6.48	1.25	-0.52	37.32	-1.45	7.71	6.86	4.90	6.24
Sb	1.67	5.43	8.83	5.30	23.46	69.16	2.69	2.09	0.99	2.44
Te	0.29	-0.10	0.42	-0.20	-0.33	-0.56	0.72	0.41	20.07	0.45
Ι	2.47	-0.86	3.66	-1.69	-2.82	-4.80	6.19	3.51	3.27	3.91
W	0.04	2.35	4.44	3.93	0.59	9.32	3.93	2.88	4.83	6.11
Pb	1.69	2982.42	350.30	18.79	80.04	109.57	51.07	31.55	2.24	24.96
Bi	-0.12	-0.10	1.42	0.00	16.63	1.37	15.42	5.62	30.40	8.05
Nb	0.83	1.48	3.62	2.45	2.79	0.95	0.68	-0.17	0.33	-0.11
La	1.39	-3.12	0.33	-1.55	-2.77	1.90	9.65	-1.76	-1.08	0.26
Ce	2.53	-7.52	-0.84	-3.97	-5.89	4.32	20.11	-4.08	-2.21	0.32
Pr	0.38	-0.93	0.04	-0.48	-0.71	0.62	2.68	-0.41	-0.22	0.17
ING See	1.82	-3.03	0.82	-1.42	-2.08	3.09	12.34	-1.49	-0.42	0.51
Sm	0.47	-0.93	-0.10	-0.55	-0.55	0.89	2.78	-0.55	0.43	0.09
Th	0.17	-0.10	-0.09	0.00	-0.03	-0.30	0.33	-0.18	-0.07	-0.10
Dv	1 38	-0.28	-0.50	0.46	0.13	1 23	1.93	-0.84	1.91	-0.12
Но	0.29	0.01	-0.10	0.10	0.06	0.19	0.36	-0.20	0.38	-0.01
Er	0.80	0.07	-0.25	0.29	0.39	0.40	0.94	-0.64	1.06	-0.10
Tm	0.14	0.04	-0.01	0.09	0.08	0.05	0.16	-0.06	0.16	0.01
Yb	0.72	0.17	-0.29	0.31	0.37	0.00	0.78	-0.42	0.73	-0.02
Lu	0.06	0.02	-0.06	0.05	0.07	0.02	0.10	-0.04	0.11	0.01
Hf	0.30	0.78	-0.13	-0.13	-0.01	0.12	0.17	-0.81	-0.85	-1.36
Та	0.01	0.06	0.04	0.02	0.05	-0.01	0.02	0.00	0.01	0.00
Th	0.34	0.43	0.39	0.43	0.23	-0.15	0.04	-0.32	0.85	-0.25

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion Alteration

Sample ID	H501341	H501342	H501343	H501344	H501345	H501346	H501347	H501348	H501349	H501350
Hole ID	BD10-18	BD11-190	BD11-190	BD11-190	BD11-189	BD11-189	BD11-189	BD11-189	BD11-189	BD10-106
Depth (m)	31.3	9.8	24.9	39.3	11.3	19.5	20.7	23.7	38.2	96
Lithology	FBX	FLT	FLT	FLT	FLT	FFL	FLT	INT	INT	FLT
Alteration	Chl	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiO	6.88	1 22	0.96	2 2 5 61	0.51	7.60	22.50	5.06	12.08	10.83
3102	-0.88	1.33	0.90	2.23	0.31	-7.09	-33.23	3.00	12.98	10.85
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	5.68	1.15	5.72	2.47	4.30	8.83	0.52	4.47	1.39	2.23
MnO	0.17	-0.06	-0.05	0.02	0.01	0.28	-0.04	0.09	-0.05	0.06
MgO	7.47	-0.97	-1.28	0.38	-0.04	1.89	-1.05	0.74	-0.96	3.10
CaO	-0.09	-0.11	-0.12	-0.09	-0.11	-0.05	-0.11	-0.10	-0.12	-0.05
Na <sub>2</sub> O	-2.00	-1.87	-1.84	-1.86	-1.85	-1.83	-1.82	-1.86	-1.83	-1.84
K <sub>2</sub> O	-1.37	0.69	0.83	0.35	0.42	0.13	0.90	0.19	0.68	0.06
TiO <sub>2</sub>	0.03	0.03	0.03	0.05	0.04	0.11	0.03	0.03	0.02	0.07
$P_2O_5$	0.03	0.01	0.00	0.02	0.02	0.01	0.02	0.01	0.00	0.05
Net Oxide Change	3.05	0.19	4.24	3.59	3.30	1.68	-34.78	8.63	12.12	14.51
Ва	-1114.43	-129.03	-298.48	-355.88	-203.96	-394.71	-62.93	-375.37	77.89	-420.42
Sr	-19.83	-12.88	-12.10	-12.45	-14.47	-10.95	-11.48	-12.95	-14.05	-12.66
Y	5.58	4.58	4.99	0.24	5.12	1.89	0.44	9.63	-0.41	-5.15
Sc	0.87	2.16	2.10	1.67	2.91	3.26	2.49	3.57	2.49	2.84
Zr	15.92	0.94	10.05	4.55	13.17	48.38	12.78	26.92	5.97	-3.75
Be	-0.46	-0.49	-0.45	-0.47	-0.46	-0.45	-0.34	-0.43	-0.43	-0.42
V	2.94	3.58	0.25	7.05	0.19	4.13	-0.86	0.33	0.34	22.01
Hg	14.09	-9 47	1.20	-3.52	-9.31	115.08	-6.74	-6.34	-9.16	-3.83
Cr 52	0.39	0.06	12.37	12.20	32.90	36.42	12.88	26.23	25.62	35.11
Co	-0.62	0.15	4 25	0.00	5 57	21.66	0.01	4 86	0.94	0.84
Ni	9.32	0.07	0.50	0.30	24 73	17.14	4 99	12.10	12.56	19 99
Cu	229.42	893.49	778 87	320.76	-61.32	1329.95	-77.21	-51.26	-72 75	272.28
Zn	301.76	-63.09	33.04	-68 74	-3.84	40.35	-86.24	-29.07	-36.15	17.20
As	30.91	21.50	13.80	0.85	15.92	21.30	0.40	3 20	-1 21	0.32
Se 78	1 17	0.18	1 33	0.81	1.01	1.40	-4.57	1.75	1.21	2 23
Br 79	98.66	77 25	38.87	60.37	55.41	18 55	-35.66	54 72	67.96	65 55
Mo	1 35	2.96	6.08	5 3 5	11 60	17.48	-55.00	8 50	7.99	12.80
Ag 107	4.55	2.90	0.08	0.03	0.04	0.05	0.17	0.07	0.07	0.08
Ag 107	0.04	0.01	0.03	0.03	0.04	0.05	-0.17	0.07	0.07	0.08
Cu En	2.05	5.05	0.04	1.05	0.05	0.05	-0.15	1.22	1.75	0.08
Sh	2.95	0.45	2.20	-1.05	-0.03	-0.99	-2.74	-1.55	1.73	-2.27
50 T-	4.15	0.43	5.59	0.33	0.02	5.90	0.30	0.28	0.02	0.42
le	0.19	0.03	0.21	0.13	0.10	0.23	-0.74	0.28	0.29	0.36
1 W	1.01	0.25	1.84	1.12	1.41	1.94	-0.34	2.42	2.50	5.10
W	1.22	2.39	0.00	2.82	2.66	14.23	2.74	4.09	3.90	6.70
P0	1.10	0.17	1.26	0.77	0.96	19.01	-4.34	1.66	1./1	2.12
BI	1.//	1.11	8.17	0.57	0.85	7.71	-0.10	0.05	0.37	0.77
ND	1.10	-0.22	2.13	2.76	0.96	3.91	0.41	1.41	0.67	-0.07
La	-1.30	-1.16	-2.11	-2.34	-1.13	-2.43	0.16	-2.33	-2.53	-1.42
Ce D	-3.43	-3.39	-4.79	-5.83	-3.24	-9.16	-1.38	-6.21	-6.31	-5.31
Pr	-0.36	-0.32	-0.53	-0.67	-0.44	-1.27	-0.23	-0.77	-0.72	-0.82
Nd	-0.87	-0.74	-1.88	-2.80	-0.98	-5.38	0.33	-2.28	-2.69	-2./1
Sm	-0.08	-0.11	-0.33	-0.53	0.09	-0.81	0.52	0.00	-0.49	-0.41
Eu	0.34	-0.29	-0.19	-0.37	-0.08	-0.38	-0.11	-0.33	-0.42	-0.28
Ib	0.13	0.03	0.06	-0.05	0.04	-0.06	-0.04	0.11	-0.09	-0.09
Dy	1.22	0.55	0.71	0.00	0.58	0.13	-0.34	1.14	-0.23	-0.58
Но	0.26	0.09	0.08	-0.07	0.05	0.02	-0.15	0.19	-0.11	-0.21
Er	0.77	0.30	0.18	-0.12	0.15	0.07	-0.54	0.53	-0.39	-0.65
Im	0.12	0.06	0.05	-0.04	0.04	0.05	-0.06	0.08	-0.03	-0.08
Yb	0.60	0.11	-0.01	-0.42	-0.10	0.55	-0.39	0.43	-0.59	-0.67
Lu	0.10	0.03	0.01	-0.03	0.07	0.17	-0.03	0.12	0.00	-0.05
Hf	0.04	-0.42	-0.82	0.17	-0.31	1.37	-0.52	0.12	0.01	-0.23
Та	0.02	0.00	0.02	0.05	0.05	0.08	0.00	0.04	0.03	0.00
Th	0.14	-0.40	-0.09	-0.09	-0.29	1.12	-0.31	-0.05	-0.35	-0.96

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

#### Alteration

Sample ID	H501351	H501352	H501353	H501354	H501355	H501356	H501357	H501358	H501359	H501360
Hole ID	BD10-106									
Depth (m)	13.9	17.1	18.4	21.8	29.4	36.9	40.5	44.3	51.2	68.5
Lithology	FBX	FT	FLT	FBX	FLT	FLT	FT	FLT	FBX	FFL
Alteration	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Chl-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser	Otz-Ser
SiOa	0.18	-3.68	1 33	-2.46	-9.96	-6.57	-10.61	2.99	9.92	8 51
A 0	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
A <u>12</u> O3	0.00	2.26	2.22	1.92	3.02	0.00	1.94	1.41	0.68	0.00
reo Mao	0.93	2.20	2.22	1.85	3.02	0.97	1.84	1.41	-0.68	0.38
MINO	0.08	-0.04	0.03	-0.04	0.09	0.02	0.01	-0.03	0.01	0.02
MgO	4.97	-0.42	0.56	-0.55	2.28	0.35	1.53	-0.24	-0.07	-0.80
CaO	0.30	-0.10	-0.01	-0.11	-0.10	-0.10	-0.08	-0.11	-0.11	-0.10
Na <sub>2</sub> O	-1.88	-1.80	-1.83	-1.81	-1.80	-1.80	-1.80	-1.81	-1.82	-1.80
K <sub>2</sub> O	-0.43	0.76	0.45	0.71	0.23	0.65	0.40	0.66	0.71	1.06
11O <sub>2</sub>	0.04	0.05	0.05	0.06	0.06	0.05	0.13	0.07	0.05	0.05
$P_2O_5$	0.02	0.02	0.02	0.03	0.01	0.02	0.04	0.02	0.02	0.02
Net Oxide Change	4.21	-2.96	2.82	-2.35	-6.17	-6.40	-8.54	2.96	8.03	7.33
Ba	-590.05	-240.29	-165.27	-300.77	-498.59	-331.77	-453.49	-276.60	-243.20	-282.43
Sr	-5.18	-12.07	-12.55	-12.05	-9.41	-11.60	-10.80	-10.52	-12.34	-11.23
Y	5.16	0.72	6.17	5.77	1.68	2.61	1.12	3.79	-2.50	4.08
Sc	2.62	1.92	1.56	1.94	2.59	2.29	4.93	3.57	2.88	2.93
Zr	13.75	-1.02	9.42	10.19	16.74	11.99	-19.97	3.21	2.08	12.26
Be	-0.47	-0.50	-0.47	-0.50	-0.52	-0.53	-0.07	0.04	0.07	0.08
V	4.86	7.43	6.95	6.45	9.12	-0.14	48.82	16.29	14.68	3.96
Hg	-4.64	2.89	-2.55	-5.04	-9.58	-9.64	-2.67	-9.39	-9.32	-0.15
Cr 52	26.21	-0.03	28.04	-0.02	-0.14	27.89	53.01	16.86	0.33	0.34
Co	0.51	0.18	-0.64	0.46	-0.67	-0.68	2.42	0.26	-0.63	0.88
Ni	16.99	-0.04	0.25	-0.03	-0.16	11.38	22.56	0.22	0.37	0.39
Cu	-42.58	239.13	1870.85	-29.02	-74.30	652.00	-62.98	-73.60	-52.08	-24.01
Zn	25.26	-70.66	-71.41	-94.50	-44.05	-2.97	-8.71	-58.73	-75.08	-91.14
As	-0.20	10.41	2.81	12.26	2.80	0.00	2.59	0.35	-1.26	1.03
Se 78	0.69	-0.09	0.67	-0.07	-0.42	-0.72	-0.89	0.58	0.98	1.03
Br 79	48.92	47.02	78.55	75.15	34.45	49.53	0.30	71.97	80.27	81.67
Mo	7.85	4.17	8.72	5.33	3.38	8.53	8.20	3.11	3.28	12.55
Ag 107	0.03	0.00	0.03	0.00	-0.02	-0.03	-0.03	0.02	0.04	0.04
Cd	0.02	0.00	0.02	0.00	-0.01	-0.02	-0.03	0.02	0.03	0.03
Sn	-2.95	-1.85	-1.73	-2.28	-2.08	-2.01	-2.79	-0.18	-1.19	-2.40
Sb	0.00	1.66	0.47	0.78	0.03	-0.02	1.05	0.59	0.61	0.30
Te	0.11	-0.02	0.11	-0.01	-0.07	-0.12	-0.14	0.09	0.16	0.17
Ι	0.95	-0.13	0.93	-0.10	-0.58	-1.00	-1.24	0.81	1.36	1.43
W	5.49	1.48	2.49	2.02	0.82	3.02	6.22	6.37	5.29	3.20
Pb	0.65	-0.09	0.64	-0.07	-0.40	-0.69	-0.85	0.55	0.93	0.98
Bi	0.05	2.74	3.97	2.33	0.19	0.41	0.68	0.15	0.24	2.30
Nb	1.14	2.35	3.66	2.37	3.69	1.19	-0.53	2.50	2.84	2.37
La	-0.50	-3.00	-3.85	0.50	-0.75	-2.42	-3.51	-3.86	0.36	-0.70
Ce	-1.58	-5.74	-6.78	-1.40	-1.93	-5.18	-7.15	-7.39	0.75	-2.20
Pr	-0.29	-0.81	-0.82	-0.46	-0.30	-0.66	-0.68	-0.86	0.06	-0.18
Nd	-0.67	-3.20	-3.19	-2.68	-0.54	-1.75	-1.87	-3.51	-0.26	-0.54
Sm	-0.14	-0.94	-0.73	-0.80	-0.23	-0.18	-0.12	-0.82	-0.13	-0.29
Eu	-0.21	-0.30	-0.33	-0.32	-0.22	-0.31	-0.37	-0.30	-0.31	-0.38
Tb	0.02	-0.14	0.05	-0.01	-0.02	-0.05	-0.09	-0.04	-0.16	0.00
Dy	0.30	-0.62	0.92	0.40	0.10	0.04	-0.36	-0.07	-1.19	0.40
Но	0.07	-0.20	0.14	0.06	0.04	0.02	-0.11	-0.03	-0.28	-0.02
Er	0.23	-0.43	0.26	0.11	0.22	0.26	-0.27	-0.22	-0.68	-0.10
Tm	0.06	-0.06	0.06	0.02	0.04	0.03	-0.05	-0.03	-0.08	-0.01
Yb	0.10	-0.82	0.11	-0.09	0.09	0.14	-0.67	-0.33	-0.46	-0.18
Lu	0.03	-0.10	0.03	-0.03	0.01	0.04	-0.10	-0.11	-0.04	-0.03
Hf	0.55	-0.65	0.66	-0.06	0.78	0.24	-0.94	0.10	1.34	1.43
Та	0.02	0.06	0.12	0.07	0.10	0.02	-0.03	0.07	0.10	0.12
Th	-0.10	0.57	1.01	0.50	1.27	-0.14	-1.21	0.18	0.54	1.37

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow

INT= Intrusion

Alteration

Sample ID	H501361	H501363	H501364	H501365	H501366	H501367	H501368	H501369	H501370	H501371
Hole ID	BD11-129	BD10-9	BG10-7	BG10-7	BG10-7	BG10-7	BG10-7	BG10-7	BD11-183	BD11-183
Depth (m)	28.3	10.1	6.5	18.7	31.9	47.2	55.2	63	8.2	21.1
Lithology	FFL	HBX	HFL	FLT	FLT	FLT	FBX	FLT	HFL	HBX
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser	Chl-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	30.17	-18.65	34.81	-15.84	-8.48	4.09	-12.00	-21.21	59.03	-28.03
$A_{12}O_3$	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	6.66	1.62	0.97	0.49	7.26	3.33	1.96	3.58	8.96	3.75
MnO	-0.09	0.07	-0.08	-0.08	-0.06	-0.08	0.04	0.13	-0.07	-0.04
MgO	-2.15	0.21	-0.75	-1.13	-0.11	-1.52	0.63	1.64	-1.89	-0.93
CaO	-0.14	-0.13	-0.12	-0.13	-0.11	-0.12	-0.10	-0.10	-0.13	-0.11
Na <sub>2</sub> O	-1.77	-1.94	-1.86	-1.86	-1.87	-1.86	-1.88	-1.90	-1.84	-1.85
K <sub>2</sub> O	0.95	0.57	0.73	0.84	0.56	1.08	0.55	0.27	1.09	0.84
TiO <sub>2</sub>	0.01	0.02	0.02	0.03	0.05	0.02	0.04	0.04	0.01	0.04
P <sub>2</sub> O <sub>5</sub>	0.00	0.02	0.02	0.02	0.01	0.01	0.02	0.03	0.01	0.02
Net Oxide Change	33 64	-18 22	33.74	-17.65	-2.76	4 94	-10.75	-17.51	65.17	-26.31
Ba	206.84	199.23	401.17	198.96	-126.85	-177.01	-455.73	-584.35	927.07	66.45
Sr	-12.18	-17.00	-15.26	-13.72	-14.69	-13.34	-14.70	-15.11	-15.03	-15.03
Y	3.09	1.87	-1.28	0.76	2.80	1.06	2.96	5.75	18.46	2.82
Sc	2.63	2.51	0.78	2.42	3.58	1.91	1.86	2.05	-1.29	1.62
Zr	0.91	8.47	3.18	1.61	-4.97	3.32	7.33	-2.99	8.23	6.78
Be	-0.30	0.67	-0.33	-0.17	-0.48	-0.46	-0.09	-0.14	-0.13	-0.23
V	5.92	3.34	6.93	4.95	13.17	4.00	-0.22	-0.35	7.95	1.37
Hg	66.57	153.14	-1.22	-1.23	92.50	0.99	-9.72	0.05	-1.55	-4.26
Cr 52	21.59	-0.74	13.87	-0.76	0.20	13.67	19.59	-0.62	3.30	23.27
Co	12.19	-0.73	1.46	-0.74	7.62	-0.62	-0.70	-0.17	28.54	10.20
Ni	2.01	-0.83	1.74	-0.86	0.22	0.41	8.90	-0.70	3.71	8.80
Cu	948.34	-75.56	-70.76	-75.62	-35.64	-73.24	759.22	429.38	-67.08	-65.93
Zn	-102.52	600.25	-80.85	-115.49	-99.40	-122.56	-15.87	140.23	-94.64	-38.53
As	15.75	-0.55	1.40	4.78	123.96	12.86	13.50	56.95	42.54	9.95
Se 78	5.38	-2.22	4.63	-2.29	0.60	1.10	-1.17	-1.86	9.90	-3.01
Br 79	138.30	4.37	206.52	19.17	62.43	73.52	7.42	35.17	200.14	-8.82
Mo	10.51	1.40	4.75	2.65	5.47	6.63	6.44	2.46	13.55	46.67
Ag 107	0.20	-0.08	0.17	-0.09	0.02	0.04	-0.04	-0.07	0.37	-0.11
Cd	0.18	1.60	0.16	-0.08	0.02	0.04	-0.04	-0.06	0.33	-0.10
Sn	-0.83	-0.02	-0.88	-2.43	-1.76	-1.46	-1.88	12.01	2.23	-1.98
50 T.	1.80	0.86	1.66	1.80	6.47	1.83	0.41	13.91	2.38	1.23
Ie	0.87	-0.50	6.42	-0.57	5.15	0.18	-0.19	-0.30	1.39	-0.49
I W	7.40	-3.07	0.42	-3.18	0.85	1.55	-1.03	-2.38	13.71	-4.17
Db	5.00	2.09	4 30	2.17	2.55	1.05	1.24	1.76	0.38	2.72
Bi	11.01	0.04	2.10	-2.17	7 99	2 75	2 90	-1.70	4.87	-2.80
Nh	0.80	3.04	2.10	1.70	3.02	1.92	0.46	-0.02	2 36	1.76
La	-2.31	-1 47	-0.87	-2.13	-1 41	-3 34	-4 36	-3.98	-3.83	-3 64
Ce	-6.29	-3 49	-2.16	-5.21	-3.63	-8.20	-10.06	-8.68	-8.67	-8.32
Pr	-0.75	-0.37	-0.22	-0.60	-0.33	-0.98	-1.28	-1.06	-0.98	-0.96
Nd	-2.92	-1.26	-0.21	-1.65	-1.15	-3.74	-5.19	-3.71	-3.08	-3.98
Sm	-0.63	-0.35	0.01	-0.48	-0.17	-0.85	-0.86	-0.83	0.00	-0.80
Eu	-0.01	0.45	-0.14	0.17	0.10	0.76	-0.20	-0.18	-0.27	-0.44
Tb	-0.09	0.02	-0.06	-0.02	0.00	-0.09	-0.07	0.04	0.41	-0.08
Dy	-0.42	0.24	0.08	0.00	0.20	-0.30	-0.24	0.66	2.72	-0.15
Но	-0.12	0.00	-0.03	-0.01	0.04	-0.08	-0.03	0.10	0.49	-0.02
Er	-0.66	0.05	-0.21	-0.14	0.12	-0.11	-0.11	0.37	1.30	-0.11
Tm	-0.08	0.00	-0.05	-0.01	0.01	-0.04	-0.01	0.06	0.17	-0.03
Yb	-0.70	-0.12	-0.25	-0.43	-0.02	-0.22	-0.31	0.29	0.88	-0.31
Lu	-0.11	0.04	-0.07	-0.05	-0.02	-0.06	0.03	0.04	0.13	-0.04
Hf	0.43	0.60	-0.07	-0.25	0.20	-0.77	-0.86	-0.95	0.51	-0.42
Та	0.03	0.09	0.09	0.04	0.05	0.06	0.00	-0.02	0.05	0.00
Th	1.14	1.41	1.17	0.80	1.11	0.51	-0.57	-0.74	1.51	-0.53

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow

INT= Intrusion

Alteration

Sample ID	H501372	H501373	H501374	H501375	H501376	H501377	H501378	H501379
Hole ID	BD11-183	BD11-183	BD11-183	BG10-2	BG10-2	BG10-2	BG10-2	BG10-2
Depth (m)	34.3	52	68	10.1	20.4	27.6	31.9	44.3
Lithology	HFL	HFL	HFL	HFL	HFL	FLT	FBX	FLT
Alteration	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser	Qtz-Ser
SiO <sub>2</sub>	4.83	26.12	-44.87	7.84	1.32	-4.55	11.58	-6.10
A <sub>12</sub> O <sub>3</sub>	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
FeO	2.28	2.68	-0.42	1.64	0.81	4.45	1.02	1.79
MnO	-0.04	-0.06	-0.06	0.00	-0.03	0.12	0.07	0.01
MgO	-0.94	-1.56	-0.99	-0.29	-0.99	0.35	-0.10	-0.14
CaO	-0.11	-0.07	-0.10	-0.12	-0.10	1.12	1.08	-0.11
Na <sub>2</sub> O	-1.86	-1.84	-1.86	-1.88	-1.86	-1.88	-1.88	-1.85
K <sub>2</sub> O	0.82	0.99	0.92	0.57	0.94	0.66	0.85	0.42
TiO <sub>2</sub>	0.01	0.05	0.05	0.01	0.01	0.06	0.03	0.02
P <sub>2</sub> O <sub>5</sub>	0.00	0.00	0.01	0.01	0.02	0.02	0.01	0.01
Net Oxide Change	5.00	26.29	-47.31	7 77	0.13	0.34	12.66	-5 94
Ba	-118.42	30.07	-14.53	-367.50	-277.49	-438.67	-463.38	-449.50
Sr	-15.56	-10.39	-15.21	-16.53	-14.93	3.15	7.56	-11.46
Y	3.91	-0.31	-7.12	8.05	3.39	5.01	-0.35	5.09
Sc	-0.34	2.90	2.02	0.94	1.10	3.62	2.50	2.46
Zr	11.64	5.90	7.10	5.99	3.38	4.85	14.14	12.19
Be	-0.46	0.29	0.05	-0.45	0.01	-0.48	0.14	-0.04
V	6.08	23.30	6.91	4.06	0.02	11.12	5.46	5.17
Hg	-9.32	-4.26	25.63	59.12	-2.92	-0.47	-9.16	-9.60
Cr 52	0.32	32.86	-2.12	0.42	23.23	0.21	8.78	-0.18
Co	2.87	3.30	-0.87	2.99	1.95	12.13	0.55	4.45
Ni	0.36	1.45	-2.39	0.47	11.02	0.24	0.68	-0.21
Cu	-73.33	-71.30	-78.47	223.10	3.10	6.81	-72.73	-74.40
Zn	-92.74	-100.29	-113.36	320.14	-0.41	-36.79	-20.69	110.62
As	3.77	8.87	8.33	0.25	2.78	44.68	2.32	-0.25
Se 78	0.97	3.87	-6.37	1.26	0.13	0.64	1.83	-0.55
Br 79	61.56	134.03	-65.41	74.47	20.23	99.83	40.06	32.85
Mo	9.36	10.11	5.61	1.66	11.16	6.04	4.26	2.80
Ag 107	0.04	0.14	-0.24	0.05	0.00	0.02	0.07	-0.02
Cd	0.03	0.13	-0.21	1.75	0.00	0.02	0.06	-0.02
Sn	-1.08	-1.24	-3.38	-1.78	-2.26	7.26	-0.88	-0.81
Sb	0.80	3.90	5.54	1.43	0.76	1.97	1.18	0.41
le	0.16	0.62	-1.03	0.20	0.02	0.10	0.29	-0.09
1 W	1.35	5.36	-8.83	1.74	0.17	0.89	2.53	-0.//
W DL	2.71	7.50	2.12	5.95	2.55	0.40	0.54	1.06
FU Di	0.92	2.07	-0.04	0.16	0.12	3.26	0.44	-0.33
Nh	2.11	2.54	1.78	1 00	0.00	0.27	3.06	2.60
Ind	-1.68	-7.29	-4.35	-1.94	-1.85	3.14	-2.13	-1.00
Ce	-4 45	-15.88	-8.81	-4.62	-5.33	6.19	-5.27	-2.86
Pr	-0.51	-2.06	-1.06	-0.58	-0.71	0.87	-0.69	-0.43
Nd	-1.70	-8.91	-3.88	-2.37	-2.18	4 00	-2.65	-1 01
Sm	-0.34	-2.05	-1.00	-0.46	-0.37	1.29	-0.74	-0.25
Eu	-0.41	-0.61	-0.02	-0.23	-0.26	0.43	-0.08	-0.09
Tb	-0.01	-0.25	-0.21	-0.06	-0.08	0.27	-0.13	0.02
Dy	-0.01	-0.85	-1.28	-0.05	-0.20	1.78	-0.51	0.43
Но	-0.03	-0.15	-0.28	0.04	-0.04	0.29	-0.06	0.15
Er	0.01	-0.38	-0.88	0.27	0.05	0.68	-0.06	0.43
Tm	0.00	-0.05	-0.10	0.03	0.02	0.13	0.00	0.08
Yb	-0.17	-0.39	-0.96	0.07	0.13	0.55	-0.08	0.27
Lu	-0.01	-0.07	-0.15	-0.02	0.04	0.09	0.00	0.08
Hf	0.21	1.07	1.08	-0.03	-0.30	0.49	2.35	0.21
Та	0.09	0.09	0.04	0.08	-0.01	0.02	0.13	0.10
Th	0.99	0.79	0.39	0.73	-0.39	-0.02	1.01	0.79

Appendix C: Table C.1.2 Mass changes for samples at the Boundary deposit

## Units

HFL= Hanging wall flow HBX= Hanging wall breccia FLT= Footwall lapilli tuff FT= Footwall tuff FBX= Footwall breccia FFL= Footwall flow INT= Intrusion

## Alteration



Appendix C.2: Mass Changes by Alteration Assemblage and Lithology

\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.



\*Whiskers extend to 5<sup>th</sup> and 95<sup>th</sup> percentile; diamonds represent the mean.

## **Appendix D: Principal Component Analysis**

## **D.1-Supplementary Principal Component Analysis Methods**

A principal component analysis (PCA) was carried out on the geochemical dataset presented in Appendix B. A PCA is useful in determining geochemical trends in large data sets by grouping elements that exhibit similar behavior together. Prior to running the analysis, data had to be pretreated. Any element that contained more than 10% of values below the limit of detection had to be removed. Once these elements were removed the data underwent a pair of statistical tests known as the Kaiser-Meyer-Olkin (KMO) and Bartlett's tests. These tests are used to determine if the data require further treatment (e.g., log center transformation). The results of these tests indicated that the data do not require further treatment. One final round of data reduction was performed prior the analysis: a correlation matrix was created using the remaining elements and any elements that had low correlation coefficients (<0.5) with all other elements were removed. The elements that remained after reduction and that were used in the analysis are as follows: SiO<sub>2</sub>, Al<sub>2</sub>O<sub>3</sub>, FeO, MnO, MgO, CaO, Na<sub>2</sub>O, K<sub>2</sub>O, TiO<sub>2</sub>, P<sub>2</sub>O<sub>5</sub>, LOI, Ba, Sr, Y, Sc, V, Hg, Cr, Co, Cu, Zn, As, Br, Mo, Sn, Sb, W, Pb, Bi, and Lu. Previous PCA attempts included all rare earth elements and high field strength elements; however, these elements are essentially immobile and the results were heavily skewed towards these group of elements. In order to focus the analysis on the mobile elements, only one high field strength (Y) and one rare earth element (Lu) was included.

The actual analysis was performed using the IBM SPSS Statistics<sup>TM</sup> software. The conditions for the principal component analysis are as follows: the analysis used a correlation coefficient matrix, KMO and Bartlett's test of sphericity, utilized the varimax rotation option, and suppressed any coefficients below 0.3.

The results of the analysis are found in Table D.1.1. For a complete explanation of principal component analyses please refer to "Statistics and Data Analysis in Geology" by J.C. Davis and "Statistical Data Analysis Explained: Applied Environmental Statistics with R" by C. Reimann, P. Filzmoser, R. Garrett, and R. Dutter.

Table D1.1 KMO and Bartlett's Test							
Kaiser-Meyer-Olkin Measure of Samplin	g Adequacy.	0.394					
Bartlett's Test of Sphericity	Approx. Chi-Square	9149.84					
	df	435					
	Sig.	0.00					

# Table D1.2 Total Variance Explained

<i></i>	Initial Eigenvalues			Extracti	on Sums of Squ	ared Loadings	Rotation Sums of Squared			
Component	Total	0/ of Variance	Cumulative 0/	Total	0/ of Variance	Cumulative 0/	Total	Loading	S Cumulative 0/	
1	10181			Total			T0(a)	% of variance	Cumulative %	
I	5.947	19.824	19.824	5.947	19.824	19.824	5.222	17.408	17.408	
2	5.264	17.547	37.372	5.264	17.547	37.372	4.569	15.230	32.638	
3	3.094	10.315	47.687	3.094	10.315	47.687	2.952	9.839	42.477	
4	2.185	7.283	54.970	2.185	7.283	54.970	2.718	9.059	51.536	
5	1.766	5.887	60.857	1.766	5.887	60.857	2.082	6.939	58.475	
6	1.458	4.859	65.716	1.458	4.859	65.716	1.800	6.001	64.476	
7	1.284	4.281	69.997	1.284	4.281	69.997	1.541	5.136	69.612	
8	1.088	3.628	73.625	1.088	3.628	73.625	1.204	4.013	73.625	
9	.944	3.148	76.773							
10	.877	2.923	79.697							
11	.739	2.465	82.162							
12	.716	2.386	84.548							
13	.662	2.207	86.754							
14	.621	2.069	88.823							
15	.526	1.753	90.576							
16	.455	1.518	92.094							
17	.438	1.461	93.555							
18	.367	1.222	94.777							
19	.352	1.174	95.952							
20	.281	.935	96.887							
21	.246	.821	97.708							
22	.198	.661	98.368							
23	.158	.528	98.896							
24	.113	.377	99.273							
25	.082	.274	99.547							
26	.055	.183	99.729							
27	.052	.174	99.903							
28	.025	.083	99.986							
29	.004	.014	100.000							
30	.000	.000	100.000							

		Component										
	1	2	3	4	5	6	7	8				
FeO	.895											
SiO2	840			307								
LOI	.831			.347								
Co	.693											
Мо	.669											
Cu	.669											
K2O	608	.369		541								
W	.578				.306							
Bi	.567							.314				
Al2O3		.911										
Sc		.864			.321							
Lu		.836										
Y		.834										
TiO2		.718			.592							
P2O5		.411		.401	.402							
Zn			.930									
Hg			.917									
Pb			.876									
Sn			.455									
MnO				.698								
CaO				.683		.465						
MgO	.483	.371		.682								
Ва	450	.348		495								
v					.857							
Cr					.741							
Sr				.303		.884						
Na2O						.808						
Sb							.770					
As	.331						.660					
Br								825				

**Table D1.3 Rotated Component Matrix** 

Extraction Method: Principal Component Analysis.

a. Rotation converged in 12 iterations.

Rotation Method: Varimax with Kaiser Normalization.

## **Appendix E: Scanning Electron Microscope Analysis**

## E.1-Supplementary Scanning Electron Microscope Analysis Methods

Scanning electron microscope (SEM) analyses were used to document finer scale textures, trace minerals, and semi-quantitative chemical information of the sulfide phases and textural relationships. SEM analyses were undertaken at Memorial University's CREAIT-Micro-Analysis Facility, Inco Innovation Centre (MAF-IIC) using the FEI MLA 650F SEM. Analyses took place on polished thin section (25x45 mm) at a constant 25 kV and variable working distance (14.0-15.5 mm). Characteristic mineral phases and textures were identified using backscatter electron imaging, whereas semi-quantitative data were collected using point analyses and element maps. The SEM analyses were critical in identifying all the mineral phases present within the Boundary deposit. The following set of images represents typical mineral phases from each alteration assemblage.











18/03/2014 det HV WD Hf 3:33:06 PM BSED 25.00 kV 15.3 mm 448 mag 🗆

100 µm

E.1.2 SEM Image Compilation: H501332



# E.1.2 SEM Image Compilation: H501332 Cont.








E.1.2 SEM Image Compilation: H501206 Cont.



## E.1.2 SEM Image Compilation: H501206 Cont.











# E.1.2 SEM Image Compilation: H501306 Cont.



E.1.2 SEM Image Compilation: H501306 Cont.





Appendix F.1.1 Chapter 2 Figure 2.19A and 2.19B

Figure 2.19 Series of 3D models of the Boundary deposit. A) Model showing key lithological contacts, major sulfide zones, and drill holes used to determine alteration extents. Ore zones are defined by assays from additional drill holes not shown. B) Model showing CCPI values in relationship to major ore zones.



Appendix F.1.2 Chapter 2 Figure 2.19C and Figure 2.19D

Figure 2.19 Series of 3D models of the Boundary deposit. C) Model showing AlOH wavelengths. Note that wave-lengths are significantly longer in the South zone. Long wavelengths in the North zone only occur on the western end of the deposit. D) Model showing FeOH absorption depth. The higher values occur within the same samples of the high CCPI samples from Figure 19B.



Figure 2.19 Series of 3D models of the Boundary deposit. E) Alteration model of the Boundary deposit based on mass balance changes. The major chlorite gains correlate with the high CCPI values (2.19B) and FeOH absorption values (2.19D).