Economic Geology

Elevated magmatic sulfur and chlorine contents in ore-forming magmas at the Red Chris porphyry Cu-Au deposit, Northern British Columbia, Canada --Manuscript Draft--

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Abstract:	The Red Chris porphyry Cu-Au deposit is located in the Stikinia island-arc terrane in northwest British Columbia. It is hosted by the Red Stock, which has four phases of porphyry intrusions: P1, P2E, P2L, and P3. New U-Pb dating of zircon shows that these intrusions were emplaced at 211.6 \pm 1.3 Ma (MSWD = 0.85), 206.0 \pm 1.2 Ma (MSWD = 1.5), 203.6 \pm 1.8 Ma (MSWD = 1.5), and 201.7 \pm 1.2 Ma (MSWD = 1.05), respectively. The ore-forming event at Red Chris was a short-lived event at 206.1 \pm 0.5 Ma (MSWD = 0.96; weighted average age of three Re-Os analyses), implying a duration of <1 m.y., as defined by the uncertainty range. This mineralization age coincides with the emplacement age of the P2E porphyry, and is consistent with crosscutting relationships that suggest P2E was the main syn-mineralization intrusion. Zircons from P1 to P3 porphyry rocks have consistently high EuN/EuN* ratios (mostly > 0.4), indicating that their associated magmas were moderately oxidized. The magmatic water contents estimated from plagioclase and amphibole compositions suggest H2O contents of ~5 wt. %. Taken together, the P1 to P3 porphyries are interpreted to be moderately oxidized and hydrous. The four phases of porphyries are differentiated by sulfur and chlorine contents. The SO3 contents of igneous apatite microphenocrysts from the mineralization-related P2 porphyries are higher (P2E: 0.30 \pm 0.13 wt. %, n = 34; P2L: 0.29 \pm 0.18 wt. %, n = 100) than those from the pre-mineralization P1 (0.11 \pm 0.03 wt. %, n = 34) and post-mineralization P3 porphyries (0.03 \pm 0.01 wt. %, n = 13). The chlorine contents in apatite grains from the P2E and P2L porphyries are 1.18 \pm 0.37 (n = 34) and 1.47 \pm 0.28 wt. % (n = 100), also higher than those from P1 (0.51 \pm 0.3 wt. % CI, n = 34) and P3 (0.02 \pm 0.02 wt. % CI, n = 17). These results imply that the sulfur and chlorine

contents of the P2E and P2L magmas were higher than in the P1 and P3 magmas, suggesting that elevated magmatic S-Cl contents in the P2 porphyries may have been important for ore-formation. Although the process that caused the increase in sulfur and chlorine is not clear, reverse zoning seen in plagioclase phenocrysts from the P2 porphyry, and the occurrence of more mafic compositions in P2L suggest that recharge of the deeper magma chamber by a relatively S-Cl-rich mafic magma may have triggered the ore-forming hydrothermal event.

Cover Letter to Editor

Dear Editor,

Thank you again for giving us the opportunity to improve our paper entitled "Elevated magmatic sulfur and chlorine contents in ore-forming magmas at the Red Chris porphyry Cu-Au deposit, Northern British Columbia, Canada" (ID: SEG-D-18-00013R1). Particularly, we appreciate the very careful review on this manuscript by the Associated Editor, David R. Cooke.

We agree with the comments by Dr. David Cooke and all the corresponding corrections have been done following his instructions. To facilitate review of our revised paper, the version with marked changes is given. We hope our manuscript could be accepted soon.

With thanks and regards,

Jing-Jing Zhu for the co-authors.

- 1 Elevated magmatic sulfur and chlorine contents in ore-forming magmas at the Red
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- 14 Abstract
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46 Introduction

Compared to the relatively long-lived magmatic systems represented by a composite batholith (up to 10 m.y.; Matzel et al., 2006; Walker et al., 2007; Harris et al., 2008; Schaltegger et al., 2009; Paterson et al., 2011; Rezeau et al., 2016), the life spans of porphyry magmatic-hydrothermal ore-forming events are much shorter, probably on the order of several hundred thousand years (up to 1 m.y.; Cathles et al., 1997; Hayba and Ingebritsen, 1997; Masterman et al., 2004; von Quadt et al., 2011; Weis et al., 2012; Chiaradia et al., 2013; Correa et al., 2016; Li et al., 2017). In many large magmatic systems, porphyry formation occurs at a relatively late stage in the system's evolution (Creasey, 1977; Titley and Beane, 1981; Candela, 1992; Richards, 2003; Rohrlach and Loucks, 2005; Yang et al., 2009; Correa et al., 2016). Although the multiple phases of porphyry stocks are commonly broadly cogenetic, they may be derived from packets of magma that evolved at different crustal levels over the history of the larger magmatic system (Annen et al., 2006). Understanding why ore-formation is only associated with a specific intrusive suite within these broader systems, and at discrete, commonly singular times, is a focus of this study. It is recognized that magmas with high sulfur (>1000 ppm), chlorine (>3000 ppm), and water (>4 wt. %) contents as well as relatively high oxidation states (higher than the fayalite-magnetite-quartz buffer, ΔFMQ , by 1–2 log fO_2 units) are fertile for the generation of magmatic-hydrothermal porphyry Cu deposits (Burnham, 1979; Candela, 1992; Richards, 2003, 2009, 2011, 2015; Wallace, 2005; Chambefort et al., 2008; Simon and Ripley, 2011; Chiaradia et al., 2012; Loucks, 2014; Hou et al., 2015; Lu et al., 2015, 2016; Chelle-Michou and Chiaradia, 2017). These ingredients are essential for the

transport of Cu (and Au) both in the magma (Zajacz et al., 2012) and in exsolved high temperature, SO₂-rich, saline magmatic fluids (Candela and Holland, 1984; Zajacz et al., 2011). However, whereas many magmas achieve such compositions, ore formation is a relatively rare and discrete event in such systems. Various mechanisms have been proposed that might trigger an ore forming event from a fertile magma source, including recharge of the magma chamber by hot, sulfur-rich melts (Hattori, 1993; Keith et al., 1997; Larocque et al., 2000; Hattori and Keith, 2001; Halter et al., 2005; Stern et al., 2007; Nadeau et al., 2010, 2016; Wilkinson, 2013; Tapster et al., 2016; Zhang and Audétat, 2017), fluxing by sulfur gases (Blundy et al., 2015), or increasing the water and oxidation state of magmas during long-term fractionation in deep magma chambers (Ballard et al., 2002; Wang et al., 2014a, 2014b; Dilles et al., 2015; Lu et al., 2015, 2016). The Red Chris porphyry Cu-Au deposit provides an opportunity to test these hypotheses, because ore-formation occurred at a discrete and relatively late stage in the ~10 m.y. history of the associated magmatic system. The Red Chris deposit is located in northwest British Columbia, and contains measured and indicated resources of 1,035 million metric tonnes of ore grading 0.35% Cu and 0.35 g/t Au (Gillstrom et al., 2012). Pre-, syn-, and post-mineralization porphyry intrusions have been recognized based on detailed core logging and petrographic work (Rees et al., 2015), with mineralization occurring in a singular episode during this magmatic history. In this paper, we present detailed geochronological and geochemical data for the porphyry phases at Red Chris, and show that they have similar bulk

compositions, including all being relatively hydrous and oxidized. However, the syn-

mineralization porphyry is characterized by plagioclase with reverse zoning and apatite with relatively high sulfur and chlorine contents. We interpret these data to indicate that high magmatic S and Cl contents, in addition to high H₂O contents and oxidation state, were critical for Cu-Au ore-formation at Red Chris. We speculate that there was an injection of relatively mafic (hotter, and more S- and Cl-rich) but cogenetic magma into the mid–upper crustal source magma chamber approximately coincident with emplacement of the syn-mineralization porphyry.

Geological Setting

Red Chris is situated within the island-arc terrane of Stikinia in the Intermontane Belt of the Canadian Cordillera of British Columbia (Fig. 1; Monger and Irving, 1980; Nelson and Colpron, 2007; Nelson et al., 2013). Stikinia consists primarily of Mesozoic arcrelated igneous and sedimentary rocks, formed in response to subduction processes prior to accretion to the ancestral North American margin in the Middle Jurassic (Gabrielse, 1991; Ricketts et al., 1992; Lang et al., 1995; Mihalynuk et al., 2004; Nelson and Colpron, 2007; Logan and Mihalynuk, 2014). In northwestern Stikinia, arc assemblages comprise the Middle to Late Triassic Stuhini Group, unconformably overlain by Late Triassic and Early to Middle Jurassic volcanic and sedimentary rocks of the Hazelton Group (Fig. 2; Brown et al., 1991; Gabrielse, 1991; Marsden and Thorkelson, 1992; Dostal et al., 1999; Gagnon et al., 2012; Nelson et al., 2013; Logan and Mihalynuk, 2014). The Hazelton Group is overlain by sedimentary rocks of the syn- to post-accretion Middle Jurassic to Early Cretaceous Bowser Lake Group.

Several porphyry Cu±Au±Mo deposits occur in the region, hosted by Late Triassic to Early Jurassic arc-related plutons (Fig. 1), including Red Chris which formed in the Late Triassic Red Stock. Collectively, these regional intrusions have ages ranging from ~222 Ma to ~180 Ma, recording much of the pre-accretionary history of Stikinia (Lang et al., 1995; McMillan et al., 1995; Scott et al., 2008; Nelson et al., 2013; Logan and Mihalynuk, 2014). A significant percentage of the known porphyry Cu deposits, including Red Chris, formed during a 6-m.y. pulse of magmatism between 206 and 200 Ma, with compositions ranging from calc-alkaline to strongly alkaline (Lang et al., 1995; McMillan et al., 1995; Nelson et al., 2013; Bissig and Cooke, 2014; Logan and Mihalynuk, 2014; Micko et al., 2014; Pass et al., 2014).

Geology of the Red Chris Cu-Au deposit

The Red Chris Cu-Au deposit was discovered in the 1950s and explored intermittently in subsequent decades, with mining beginning in 2015 (Ash et al., 1995, 1996; Newell and Peatfield, 1995; Baker et al., 1997; Gillstrom et al., 2012; Rees et al., 2015). The deposit is hosted by the Red Stock, which is the largest of a suite of Late Triassic to Early Jurassic stocks and dikes that intrude the Stuhini Group in the district (Fig. 2; Friedman and Ash, 1997; Rees et al., 2015). The stock is tabular, elongate in an east to northeast direction, and approximately 6.5 km long by 300 to 1500 m wide (Fig. 2; Ash et al., 1995; Baker et al., 1997; Gillstrom et al., 2012). It has a steep northern contact against Stuhini Group country rocks, but its southeastern margin against Hazelton and Bowser Lake Group strata is poorly exposed, and has locally been truncated by the NE-

trending, steeply SE-dipping South Boundary fault (Fig. 2). This fault, and the East zone fault within the stock, probably reflect a long-lived and deep structure which guided the emplacement of the intrusions, mineralization, and subsequent deformation of the Red Stock.

The Red Stock is a composite intrusion consisting of several phases of porphyritic diorite to quartz monzonite. Mineralogically, the rocks consist of medium- to coarse-grained amphibole, plagioclase, and minor biotite phenocrysts, with K-feldspar, plagioclase, and quartz in the groundmass. Based on compositional and textural differences and crosscutting relationships, Rees et al. (2015) identified four distinct porphyry phases, P1, P2, P3, and P4. The P1 porphyry is a pre-mineralization leucodiorite which accounts for the main volume of the Red Stock, and is distinguished by sparse anhedral to subhedral amphibole (~10%) and abundant plagioclase (30–40 vol. %) phenocrysts with lengths up to 4 mm. The groundmass is composed of fine-grained plagioclase and minor quartz (Fig. 3A).

The P2 porphyry is a syn-mineralization quartz monzonite intrusion which is largely unexposed at surface but is observed in drill core to have intruded P1 at depth. It has been subdivided by Rees et al. (2015) into early (P2E), intermediate (P2I), and late (P2L) phases based on vein truncations and chilled margins. The P2 porphyries are generally characterized by tabular subhedral to euhedral amphibole (10–15 vol. %) and plagioclase (35–50 vol. %) phenocrysts. The amphibole crystals are mostly euhedral and larger than in P1, with lengths up to 10 mm. The groundmass comprises K-feldspar, plagioclase, and quartz. In this paper, the P2 subphases are simplified to an early stage (P2E; Fig. 3B) and late stage porphyry (P2L, probably corresponding to P2I and P2L of Rees et al., 2015;

Fig. 3C). The P2E porphyry has a crowded plagioclase texture with slightly smaller amphibole phenocrysts than P2L. In contrast, the P2L porphyry is relatively fresh and occurs only as small dikes (Fig. 4).

The post-mineralization P3 monzonite porphyry is much less abundant. It is texturally similar to the P2L porphyry with abundant amphibole phenocrysts (15–20 vol. %), but is distinguished by the absence of quartz in the groundmass, which is mainly composed of K-feldspar and plagioclase (Fig. 3D). Although no crosscutting relationships between the P3 and P2L porphyries have been found, the zircon U-Pb dating results (see below) confirm that P3 is younger. The P4 porphyry occurs as rare dikes and is typified by sparse fine-grained amphibole phenocrysts (Rees et al., 2015). No P4 samples were included in this study.

Several younger basaltic to andesitic dikes with sparse amphibole phenocrysts cut the Red Stock and the Stuhini Group host rocks (Figs. 3E and F). They postdate the porphyry-stage Cu-Au mineralization (Baker et al., 1997; Rees et al., 2015), but are mildly to strongly altered (Figs. 3E and F) and are crosscut by late quartz-calcite-pyrite veins.

Hydrothermal alteration

Alteration at Red Chris has been described previously by Baker et al. (1997), Gillstrom et al. (2012), Norris (2012) and Rees et al. (2015). The alteration assemblages are typical of calc-alkaline porphyry Cu systems (Lowell and Guilbert, 1970; Seedorff et al., 2005; Sillitoe, 2010), and consists of early stage potassic alteration, overprinted by chlorite–sericite, sericitic (phyllic), intermediate argillic, and minor late stage propylitic

alteration (see paragenetic diagram in Rees et al., 2015). Potassic alteration is expressed by replacement of amphibole phenocrysts by secondary biotite, plagioclase replaced or rimmed by secondary K-feldspar, and by K-feldspar veins (Figs. 3B and 5A–C). It is best preserved in the deeper levels of the deposit where it is spatially associated with the synmineralization P2E porphyry; however it locally extends into pre-mineralization P1 porphyry wall rocks (Fig. 4). The P2L and P3 porphyries were only weakly affected by potassic alteration (Rees et al., 2015).

Chlorite—sericite alteration is characterized by chlorite replacing secondary biotite (Fig. 5C) and sericite replacing feldspar (Fig. 5D). At shallower levels in the system, potassic alteration is completely overprinted by phyllic and intermediate argillic alteration (Gillstrom et al., 2012), characterized by sericite after plagioclase (phyllic; Fig. 5D), and illite and kaolinite (intermediate argillic; Norris, 2012). This lower temperature alteration overprint affects all the porphyry phases, but is less pervasive at depth.

Propylitic alteration at Red Chris is mainly observed as minor chlorite and epidote in the outer part of the Red Stock, and extends for 100 to 200 m into the Stuhini volcanic country rocks (Gillstrom et al., 2012; Norris, 2012; Rees et al., 2015).

Vein styles and mineralization

Detailed descriptions of vein styles at Red Chris have been given by Norris (2012) and Rees et al. (2015). A-type quartz veins (Gustafson and Hunt, 1975) and stockworks are associated with potassic alteration and host the bulk of the copper-gold mineralization. These veins are most intensely developed around the apex of the principal P2 porphyry body, but extend for hundreds of metres into the P1 wall rocks. Typical A veins contain

K-feldspar, biotite, chalcopyrite, bornite, and magnetite, with K-feldspar alteration halos (Figs. 5A–B and G). Copper sulfides also occur as disseminations in the host porphyry. Bornite is more abundant in the apex of P2E, and progressively decreases outwards where chalcopyrite is the dominant Cu-sulfide (Norris, 2012; Rees et al., 2015). At shallower depths in the preserved system, early bornite was sulfidized to chalcopyrite, and pyrite becomes increasingly dominant. Rees et al. (2015) delineated a high-sulfur contour (>4% S) in section above which total sulfide (dominantly pyrite) ranges from 4 to 10%. Microscopic native gold and electrum occur as inclusions in bornite (Rees et al., 2015). The grades of Cu and Au are positively correlated with quartz vein density (Gillstrom et al., 2012). In high-grade zones at depth (e.g., 4.12 % Cu and 8.83 g/t Au in hole 09-350 from 540 to 692.5 m down-hole depth), vein abundance exceeds 80 vol. % in sheeted arrays (Fig. 5G; Rees et al., 2015).

B- and D-type veins (Gustafson and Hunt, 1975) are relatively minor at Red Chris, and host only minor amounts of Cu sulfides. B quartz veins are characterized by relatively straight margins with sulfide centerlines (pyrite and minor chalcopyrite, and locally molybdenite; Fig. 5E). Pyritic D veins have variable widths (1 to 10 mm; Fig. 5F). Carbonate and minor chlorite veins cut all the earlier veins and are generally barren (Norris, 2012).

Sampling and Analytical Methods

Samples of the P1, P2E, P2L, and P3 porphyries were collected from drill core, and descriptions and locations are listed in Digital Appendix Table A1; sampled drill hole locations are also shown on Figure 2. Fourteen least-altered samples of the porphyry intrusions and three samples of late basaltic to andesitic dikes were selected for whole-

rock geochemical analysis. Eight of these samples were selected for determination of Nd-Sr isotopes. Three samples of quartz-carbonate-pyrite-molybdenite-chalcopyrite veins (Fig. 5H) were collected for Re-Os dating. Four samples of the P1 (RC13-40), P2E (RC13-107), P2L (RC13-33), and P3 (RC13-78) intrusions were selected for zircon U-Pb dating, Hf isotopic, and trace element analyses. Details of analytical methods are provided in Appendix 1.

Electron microprobe analyses

Primary igneous minerals such as plagioclase and amphibole in the porphyry rocks are widely altered to K-feldspar, sericite, and chlorite (Figs. 3A–B and 5C–D). However, a few least-altered samples of the P2E, P2L, and P3 porphyries contained unaltered plagioclase and amphibole grains. In addition, igneous apatite grains were typically preserved as inclusions within plagioclase and amphibole phenocrysts. Compared with hydrothermal apatite (acicular crystals intergrown with other hydrothermal minerals such as quartz, sericite, chlorite, and sulfides), igneous apatite grains typically showed stubby prismatic habits, as described by Richards et al. (2017). Detailed analytical methods are described in Appendix 1.

Re-Os molybdenite dating

Three samples of molybdenite from quartz-carbonate-pyrite-molybdenite-chalcopyrite veins were collected from the Gully zone (RC13-88 and RC13-103), and the East zone (RC13-82; Digital Appendix Table A1; Fig. 2). A molybdenite mineral separate was produced for each sample by metal-free crushing followed by gravity and magnetic

concentration methods. Dating was conducted at the Canadian Centre for Isotopic Microanalysis at the University of Alberta, Canada, using methods described in Appendix 1. **Geochronological Results** Zircon U-Pb ages of the Red Stock Zircon U-Pb results are presented in Digital Appendix Table A2 and illustrated in Figure 6; all ages are illustrated and reported with 2σ errors. All the zircons show oscillatory zoning under BSE imaging. Sample RC13-40 was collected from P1 leucodiorite porphyry (Digital Appendix Table A1). Analyzed zircon grains form a tightly clustered age population, mostly with low common lead contents. Except for one inherited or xenocryst zircon (apparent $^{206}\text{Pb}/^{238}\text{U}$ age = 261 ± 12 Ma), the twenty-seven grains yielded an intercept age of 211.6 ± 1.3 Ma (MSWD = 0.85; Fig. 6A), similar to the weighted mean $^{206}\text{Pb}/^{238}\text{U}$ age of 211.8 ± 1.3 Ma (MSWD = 0.8). Sample RC13-107 was collected from P2E quartz monzonite porphyry (Digital Appendix Table A1). One xenocrystic zircon grain yielded an older age (apparent $^{206}\text{Pb}/^{238}\text{U}$ age = 316 ± 10 Ma), but the other twenty-nine zircons contained low amounts of common lead and intersected the concordia line at 206.0 \pm 1.2 Ma (MSWD = 1.5; Fig. 6B), in good agreement with the weighted mean 206 Pb/ 238 U age of 205.9 \pm 1.5 Ma (MSWD = 1.5).Sample RC13-33 was collected from P2L quartz monzonite porphyry (Digital Appendix Table A1). Two xenocryst grains were found among thirty-one analyzed

zircons, with $^{206}\text{Pb}/^{238}\text{U}$ ages of 298 ± 16 Ma and 441 ± 13 Ma. The remaining twenty-

nine zircon grains had low common lead contents and yielded an intercept age of 203.6 ± 1.8 Ma (MSWD = 1.5; Fig. 6C), similar to the weighted mean 206 Pb/ 238 U age of 203.3 \pm 1.5 Ma (MSWD = 1.04).Sample RC13-78 was collected from P3 monzonite porphyry (Digital Appendix Table A1). Thirty-two zircons with low common lead contents yielded an intercept age of 201.7 ± 1.2 Ma (MSWD = 1.05; Fig. 6D), and a weighted mean 206 Pb/ 238 U age of 201.6 ± 1.2 Ma (MSWD = 1.04). All of the zircons have magmatic textures, and the intercept ages above are interpreted to be the crystallization ages of the individual intrusions. The ages are consistent with crosscutting relationships described by Rees et al. (2015), and define a 10 m.y. span of magmatism from 211.6 \pm 1.3 Ma (P1) to 201.7 \pm 1.2 Ma (P3). The relative ages of the two samples of syn-mineralization P2 porphyry are consistent with crosscutting relationships, but the dates (P2E: 206.0 ± 1.2 Ma; P2L: 203.6 ± 1.8 Ma) overlap within the 2σ uncertainty. Hence the apparent 2.4 m.y. age difference is not statistically robust, and their true ages may in fact be closer as suggested by their close relationship with ore mineral paragenesis. Re-Os molybdenite ages The Re-Os model ages for three samples are shown in Table 1. They have relatively high rhenium contents ranging from 497.8 to 1771 ppm, with common ¹⁸⁷Os of 1078 to 3821 ppb. Although the three samples are from two different mineralization zones (i.e.,

East and Gully zones; Fig. 2), they yielded indistinguishable model ages within

uncertainty: 206.5 ± 0.8 Ma, 205.7 ± 0.9 Ma, and 205.9 ± 1.1 Ma (± 2 s.d.). The small

grain size of the molybdenite (< 1 mm) minimizes the risk of decoupling between Re and 187 Os in these samples (caused by diffusion: Selby and Creaser, 2004). The results yielded a weighted average age for all three samples of 206.1 ± 0.5 Ma (95% confidence level with MSWD = 0.96; Fig. 7). This age is consistent with the age of the synmineralization P2E porphyry (206.0 \pm 1.2 Ma).

Geochemical and Isotopic Results

Whole-rock major and trace elements

Whole-rock major and trace element compositions for fourteen samples of the Red Stock and three basaltic–andesitic dike samples are listed in Table 2. All the P1 to P3 porphyry intrusions have relatively homogeneous major element compositions, but have moderate to high loss-on-ignition values (LOI: 2.3 to 10.9%) reflecting varying degrees of potassic and sericitic alteration. On an LOI-free basis, they are mostly intermediate composition (SiO₂ = 56.71-63.16 wt. %; Digital Appendix Table A3), and straddle the boundary between granodiorite (diorite) and syenite on a Zr/Ti versus Nb/Y diagram (Fig. 8). The late basaltic-andesitic dikes have distinct compositions, and plot in the field of diorite and gabbro on Figure 8. On Harker-type diagrams of SiO₂ versus selected major elements, the alkali elements (K₂O and Na₂O; Figs. 9A–B) show significant scatter, likely due to hydrothermal alteration, as confirmed by a rough correlation between alkali contents and LOI. However, other elements such as Fe₂O₃, MgO, TiO₂, and Al₂O₃ show roughly correlated trends with SiO₂ (Figs. 9E–F), suggesting a broadly cogenetic suite, with the exception of the late basaltic to andesitic dikes, which do not plot on the same trends for K₂O and TiO₂.

On primitive mantle-normalized extended trace element and chondrite-normalized rare earth element (REE) diagrams (Figs. 10A–B), the porphyritic rocks show largely indistinguishable patterns consistent with typical subduction-related igneous rocks (Hawkesworth et al., 1993; Pearce, 1996), characterized by large-ion lithophile element (LILEs: Rb, Ba, Th, U, K) and light rare earth element enrichments (LREE), low concentrations of high field strength elements (HFSE: Nb, Ta, Zr, Hf, and Ti), relative depletions in compatible elements and middle to heavy rare earth elements (MREE, HREE; La/Yb = 8.9 ± 1.7 , n = 14), and flat to listric-shaped patterns from MREE to HREE. Such listric patterns likely reflect amphibole fractionation from hydrous magmas, and are an indication of magma fertility for porphyry ore formation (Richards and Kerrich, 2007). Most samples have flat or slightly positive Eu anomalies with Eu_N/Eu_N* of 1.1 ± 0.15 (n = 14; Fig. 10B), also likely reflecting amphibole fractionation and lack of plagioclase fractionation. Three samples of late basaltic to andesitic dikes show distinct trace element patterns, especially for HFSE and REE, which are enriched relative to the porphyries; they are clearly not cogenetic with the earlier porphyries. Excluding two samples with high Sr values that may reflect calcite veining, the porphyries have relatively high Sr/Y (53 \pm 23, n = 12) and V/Sc ratios (9.0 \pm 0.9, n = 14), which overlap the range of fertile rocks for porphyry formation (Richards and Kerrich, 2007; Loucks, 2013, 2014).

Whole-rock Nd-Sr isotopes

Nd and Sr isotopic compositions of the porphyritic rocks are very homogeneous (Table 4 and Fig. 11). They show a narrow range of initial Sr isotopic ratios from 0.7042 to 0.7047 (($^{87}\text{Sr}/^{86}\text{Sr}$)_t = 0.7044 ± 0.0002, n = 8), with relatively high $\epsilon_{Nd}(t)$ values from 2.4 to 3.6 (average = +3.1 ± 0.4, n = 8), consistent with a mantle source with minor crustal contamination. The single-stage Nd model ages (T_{DM1}) calculated following the methods of Goldstein et al. (1984) range from 0.74 to 0.88 Ga (average = 0.80 ± 0.05 Ga, n = 8), and are indistinguishable for the different porphyry phases. Two basaltic–andesitic dike samples show slightly lower Sr isotopic ratios (($^{87}\text{Sr}/^{86}\text{Sr}$)_t = 0.7042 and 0.7043) and higher $\epsilon_{Nd}(t)$ ratios (+3.5 and +3.9) than the porphyry rocks (Fig. 11).

Zircon Hf isotopes

Zircon Hf isotopic results are listed in Digital Appendix Table A4, and illustrated in Figure 12. The four zircon samples from porphyritic rocks show indistinguishable Hf isotopic compositions and single-stage Hf model ages of 375 ± 52 Ma (range = 264–527 Ma, n = 56), with relatively high $E_{Hf}(t)$ values of 12.0 ± 0.4 (weighted mean value, 95% confidence level; range = 8.1–14.8, n = 56). These data suggest that the porphyries shared a common primitive mantle source, consistent with their island arc origin.

Amphibole, Plagioclase, Zircon, and Apatite Compositions

The compositions of amphibole, plagioclase, zircon, and apatite from samples of leastaltered porphyritic rocks are listed in Digital Appendix Tables A5, A6, A7, and A8, respectively. Based on these analyses, water and sulfur contents as well as oxygen fugacity of the magma have been estimated qualitatively and quantitatively.

Amphibole compositions

Amphibole compositions can be used to estimate magmatic oxidation state, crystallization temperatures and pressures, as well as water contents (e.g., Ridolfi et al., 2010; Zhang et al., 2012; Wang et al., 2014b). However, fresh amphibole grains were only observed in the P2E and P2L porphyries. All of the analyzed grains are calcic amphiboles, and two generations of amphibole were recognized in both P2E and P2L. Early stage grains occur as phenocrysts and are characterized by higher Al contents (range = 9.35-12.86 and 8.79-13.46 wt. % Al₂O₃ for P2E and P2L, respectively; Digital Appendix Table A5); they are mostly classified as magnesio-ferri-hornblende and magnesio-hastingsite (including potassic-magnesio-hastingsite; Fig. 13A). The late stage grains were developed in the groundmass or as recrystallized phenocrysts, typified by lower Al concentrations (range = 1.62-4.76 and 4.33-7.53 wt. % Al₂O₃ for P2E and P2L, respectively; Digital Appendix Table A5). They are classified as magnesio-ferrihornblende or rare actinolite (Fig. 13A). The early stage amphibole grains are intergrown with plagioclase phenocrysts (Fig. 14), and igneous apatite crystals occur as inclusions within amphibole (Fig. 14B). This indicates that these three minerals are broadly coeval. The crystallization temperature, magmatic water contents, and oxygen fugacities were estimated for the two stages of amphiboles using the spreadsheet of Ridolfi et al. (2010), and the crystallization pressures were calculated following the equation of Mutch et al. (2016). The Al-in-hornblende geobarometer used is applicable to granitoids that contain amphibole, plagioclase, quartz, and alkali feldspar, broadly consistent with the mineral assemblage within P2 porphyry rocks. The calculated results are listed in Digital Appendix Table A5 and illustrated in Figure 13 (B–C).

Amphibole phenocrysts from the P2E and P2L porphyry samples have similar compositions, with calculated crystallization pressures from 3.5 to 6.7 kbar (average = 4.6 ± 0.8 kbar, n = 51), temperatures from 855 to 983 °C (average = 900 ± 30 °C, n = 51), Δ FMQ values from 0.5 to 1.8 (average = 1.1 ± 0.3, n = 51), and H₂O contents in melts from 4.0 to 6.1 wt. % (average = 5.1 ± 0.4 wt. %, n = 51). Late stage (low-Al) amphiboles in P2E porphyry samples yielded crystallization pressures from 0.6 to 1.4 kbar (average = 1.1 ± 0.3 kbar, n = 12), temperatures from 637° to 774 °C (average = 719 \pm 34 °C, n =12), Δ FMQ values from 2.6 to 3.3 (average = 3 \pm 0.2, n = 12), and magmatic water contents from 3.6 to 5.1 wt. % (average = 4 ± 0.4 Ma, n = 12). Low-Al amphibole grains from P2L samples yielded slightly higher crystallization pressures (2.0 \pm 0.4 MPa; range = 1.3–2.6 kbar, n = 8) and temperatures (800 \pm 24 °C; range = 751-800 °C, n = 8), slightly lower Δ FMQ values (2.4 ± 0.3 ; range = 1.1-2.9, n = 8), but similar H₂O contents $(4.0 \pm 0.2; \text{ range} = 3.7-4.4, \text{ n} = 8)$ to those calculated for P2E. There are clear trends of increasing oxidation state and decreasing crystallization pressures, temperatures, and magmatic water contents from early to late stage amphiboles in both P2E and P2L porphyries (Figs. 13B–D). These trends are consistent with the high-Al amphibole phenocrysts having crystallized at depth before final crystallization of the magma (and low-Al amphibole) after emplacement at shallow levels (e.g., Rutherford and Devine, 2003). The decreasing magmatic water contents might be attributed to degassing during ascent and crystallization, which can also result in a small increase in oxygen fugacity (Mathez, 1984; Candela, 1986; Burgisser and Scaillet, 2007; Zimmer et

al., 2010; Bell and Simon, 2011; Dilles et al., 2015).

Plagioclase compositions

Plagioclase phenocrysts from all P1 porphyry samples studied have been altered, but partially unaltered phenocrysts were found in some P2 and P3 porphyry samples. Analyses with $K_2O > 1$ wt. % are interpreted to reflect potassic alteration and have been excluded. Plagioclase grains from the P2E porphyry (sample RC13-13) are classified as oligoclase (\geq An₁₇) to labradorite (\leq An₇₀), with an average anorthite (An) content of 44 \pm 12 mol % (n = 28). Nearly all these crystals exhibited reverse zoning in An content, with compositional ranges up to ~16 mol \%, and FeO contents that correlate positively with X_{An} values (Digital Appendix Table A6; Fig. 14A). Plagioclase grains from the P2L porphyry (samples RC13-26 and RC13-33) have compositions similar to P2E (An mol % = 48 ± 5 , n = 38), and also show reverse zoning with amplitudes up to ~19 mol % An that correlate with FeO contents (Fig. 14B). Such reverse zoning with positive An-FeO correlations were observed both in partially altered (i.e., sericite alteration; Fig. 14A) and relatively fresh plagioclase crystals (Fig. 14B), indicating that it was not likely to have been generated by hydrothermal alterations. In contrast, plagioclase grains from the P3 porphyry (RC13-78) display relatively uniform An compositions (An mol $\% = 51 \pm 5$, n = 37), with no clear zoning in X_{An} and FeO (Fig. 14C). Plagioclase compositions (An proportions) are very sensitive to dissolved water content and temperature of the melt, and thus can be used as a hygrometer if corrected for temperature (Mathez, 1973; Lange, 2009; Waters and Lange, 2015). We used apatite saturation temperatures (AST) following the formula of Piccoli and Candela (1994) to calculate magma temperatures for P2 and P3 samples (Table 4; Digital Appendix Table

A3). The whole-rock SiO₂ and P₂O₅ values were assumed to approximate the melt contents at the time of crystallization, although this assumption can be challenged (Piccoli and Candela, 1994, 2002). Nevertheless, the calculated temperatures are reasonable for magmas of intermediate, hydrous composition, and are also in agreement with the estimated crystallization temperatures of early stage amphibole (Digital Appendix Table A5): 932 °C for P2E (RC13-13), 892 °C (RC13-26) and 905 °C (RC13-33) for P2L, and 928 °C for P3 (RC13-78; Table 4).

The updated spreadsheet of Waters and Lange (2015) has been used to calculate magmatic water contents (Digital Appendix Table A9). The calculated values are relatively uniform and indistinguishable between the different porphyry phases: 5.2 ± 0.2 (n = 28) for P2E, 5.5 ± 0.2 (n = 38) for P2L, and 5.1 ± 0.1 wt. % (n = 37) for P3. For the P2E and P2L porphyries, the magmatic water contents calculated from plagioclase and early stage amphibole compositions agree well with each other (Digital Appendix Tables A5 and A8).

Zircon trace element compositions

Twenty trace element spot analyses were obtained for zircons from samples of P1 to P3 porphyries (Digital Appendix Table A7; Fig. 15). It is common to encounter small mineral inclusions, especially of apatite, titanite, in zircon during LA-ICP-MS analyses (e.g., Lu et al., 2016). We have taken Ca > 200 ppm or La > 0.3 ppm as an indication of apatite contamination, and Ti > 20 ppm to reflect titanite contamination, and such data were excluded (Digital Appendix Table A7). The remaining analyses have low LREE and elevated HREE contents, with small negative Eu and strongly positive Ce anomalies

(Fig. 15). These REE patterns and total contents ($\Sigma REE = 351-1125 \text{ ppm}$), as well as Th/U ratios above 0.1 (0.26–0.63) (Digital Appendix Table A7) are typical of igneous zircons from relatively oxidized magmas (Hoskin and Schaltegger, 2003). Zircon crystals from the P1 to P3 porphyries show slightly different calculated Ti-inzircon temperatures and Eu anomalies $(Eu_N/Eu_N^* = Eu_N/(Sm_N \times Gd_N)^{0.5})$; Fig. 16; Digital Appendix Table A7). Titanium-in-zircon temperatures were calculated using the equation of Ferry and Watson (2007), where it is assumed that $\log a_{SiO2} = 1$ because of the existence of quartz in the host porphyries, and $\log a_{TiO2} = 0.7$ due to the presence of titanite (a common accessory mineral in these rocks). Zircon crystals from the P2E porphyry (sample RC13-107) have the highest calculated temperatures (average of 760 \pm 29 °C; range = 723° – 819° C, n = 14) with relatively uniform Eu_N/Eu_N* values from 0.5 to 0.7 (average = 0.6 ± 0.1 ; n = 14). Zircon crystals from the P3 porphyry (sample RC13-78) show the lowest calculated temperatures (average = 726 ± 24 °C; range = 693°-758°C, n = 18) and lowest Eu_N/Eu_N* ratios (0.5 \pm 0.1; range = 0.3–0.6, n = 18). Zircon crystals from the P1 and P2L porphyries have intermediate calculated temperatures (P1: 736 ± $26 \,^{\circ}\text{C}$, range = 699° – $810 \,^{\circ}\text{C}$, n = 16; P2L: $730 \pm 24 \,^{\circ}\text{C}$, range = 705° – 795°C , n = 17), and a similar wide range of Eu_N/Eu_N* values (P1: 0.3–1.0, average = 0.6 ± 0.2 ; P2L: 0.3– 0.9, average = 0.6 ± 0.1). Trace element compositions in zircon are sensitive to magmatic water content and oxygen fugacity, and have been used to distinguish between fertile and infertile suites in porphyry Cu ± Au ± Mo systems (Ballard et al., 2002; Liang et al., 2006; Qiu et al., 2014; Wang et al., 2014b; Dilles et al., 2015; Shen et al., 2015; Lu et al., 2016; Xu et al., 2016;

Loader et al., 2017). In hydrous magmas, early plagioclase crystallization is suppressed

(Moore and Carmichael, 1998), leading to no pronounced depletion of Eu in melts (Eu²⁺ substitutes for Ca²⁺ in plagioclase). On the other hand, in oxidized magmas, Eu is predominantly present as Eu³⁺ and is partitioned into zircon along with other REE (Ballard et al., 2002; Hoskin and Schaltegger, 2003; Trail et al., 2011; Dilles et al., 2015). Therefore, the small negative Eu anomalies in zircon samples (as indicated by Eu_N/Eu_N* ratios >0.3 and mostly > 0.4; Fig. 16A), and slightly positive anomalies in whole rocks (Fig. 10) indicate relatively high magmatic water contents and/or oxidation states (Ballard et al., 2002; Dilles et al., 2015; Lu et al., 2016). Ballard et al. (2002) defined the zircon Ce⁴⁺/Ce³⁺ ratio as an indicator of magmatic oxidation state. However, Dilles et al. (2015) and Lu et al. (2016) argue that these values are difficult to estimate accurately because the abundances of the adjacent elements, La and Pr, which are used as a baseline to calculate the magnitude of the Ce anomaly, are low and close to the analytical detection limit. Confirming this reservation, Ce⁴⁺/Ce³⁺ ratios and fO_2 values calculated for our samples using the equation of Trail et al. (2011) yielded unrealistic and widely variable Δ FMQ values (-9 to +5). Apatite compositions The SO₃, Cl, and F analyses of igneous apatites are listed in Digital Appendix Table A8 and illustrated in Figures 17 and 18. The results show that igneous apatites from the P2 porphyries have higher sulfur and chlorine concentrations than those from P1 and P3 rocks. The high SO₃ contents in apatites from the P2 porphyries (P2E: 0.11 to 0.8 wt. %

 SO_3 , average = 0.30 ± 0.13 wt. %, n = 34; P2L: 0.07 to 1.2 wt. % SO_3 , average = 0.29 ±

0.18 wt. %, n = 100) are similar to values from global porphyry Cu deposits (Streck and

Dilles, 1998; Imai, 2002; Li et al., 2012; Pan et al., 2016; Richards et al., 2017). In comparison, apatite crystals from the P1 porphyry (RC13-39) have lower sulfur contents $(0.05 \text{ to } 0.19 \text{ wt. } \% \text{ SO}_3; \text{ average} = 0.11 \pm 0.03 \text{ wt. } \%, \text{ n} = 34), \text{ whereas those from the P3}$ porphyry (RC13-78) have the lowest sulfur concentrations (from 0.05 wt. % to below the analytical detection limit of SO₃; average of analyses above limit of detection = $0.03 \pm$ 0.01, n = 13). Although the ranges for P1 and P2L porphyries overlap at their outer limits, a t-test shows that the two populations are significantly different (t = 5.87, p = 0). Similarly, the Cl contents in apatite crystals from P1 and P3 porphyries are significantly lower (0.51 \pm 0.3 wt. %, n = 34, and 0.1 \pm 0.06 wt. %, n = 17, respectively) compared to values from the P2 porphyries (P2E: 1.18 ± 0.37 wt. %, n = 34; P2L: $1.47 \pm$ 0.28 wt. %, n = 100). There is a rough positive correlation between sulfur and chlorine contents of apatites from the four porphyries (Fig. 18A), with relatively constant molar S/Cl ratios (P1: 0.13 ± 0.08 , n = 34; P2E: 0.13 ± 0.08 , n = 34; P2L: 0.10 ± 0.14 , n =100; P3: 0.16 ± 0.09 , n =13; Table 4). Detailed analyses also show that single apatite crystals are commonly zoned in sulfur, with decreasing core-to-rim SO₃ contents (and small decreases in Cl; Fig. 17C), possibly reflecting progressive degassing of SO₂ (and Cl) from the melt during crystallization (Richards et al., 2017). Fluorine contents in apatites from the P1 and P3 porphyries are similar (P1: $3.64 \pm$ 0.48 wt. %, n = 34; P3: 3.6 ± 0.4 , n = 17) and distinctly higher than in the P2E (2.33 \pm 0.29 wt. %, n = 34) and P2L (2.32 \pm 0.23 wt. %, n = 100) porphyries. Some hydrothermal apatite grains were also analyzed for comparison with igneous grains, and they show wide variations in sulfur and chlorine content (Digital Appendix

Table A8), as observed by other researchers (e.g., Li et al., 2012; Richards et al., 2017).

The sulfur partition coefficient between apatite and melt is jointly controlled by temperature, oxygen fugacity, and the S content in the silicate melt (Peng et al., 1997; Parat and Holtz, 2005; Parat et al., 2011; Webster and Piccoli, 2015; Konecke et al., 2017a, b), and no formula currently exists to accurately calculate magmatic sulfur concentrations from apatite SO₃ contents. However, two semi-quantitative formulae can be used to estimate relative magmatic S content (Peng et al., 1997; Parat et al., 2011). Using the equation of Peng et al. (1997), the magmas related to the P2E and P2L porphyry intrusions are calculated to have sulfur concentrations of 0.03 ± 0.01 wt. % (n = 34) and 0.02 ± 0.01 wt. % (n = 100), significantly higher than the corresponding S contents for P1 (0.010 \pm 0.003 wt. % S, n = 34) and P3 magmas (0.003 \pm 0.001 wt. % S, n = 13; Table 4). The results calculated by the method of Parat et al. (2011) are more variable, and yielded lower absolute values than those from Peng et al. (1997). However, they also support the conclusion that the P2E and P2L magmas had significantly higher sulfur contents than the P1 and P3 magmas (Table 4). The chlorine partition coefficient between apatite and melt is a complex function of magma composition and Cl concentration in the melt (Zhu and Sverjensky, 1991; Piccolli and Candela, 1994; Mathez and Webster, 2005; Webster et al., 2009; Chelle-Michou and Chiaradia, 2017). Mathez and Webster (2005) proposed a value of 0.8 as the partition coefficient (mass ratio) between apatite and basaltic melt (51.1 wt. % SiO₂). Based on this semi-quantitative equation (and recognizing that these magmas were not basaltic), we estimate Cl concentrations in the P2E and P2L magmas of 1.48 ± 0.46 wt.% (n = 34) and 1.83 ± 0.34 wt.% (n = 100), respectively, significantly higher than for P1 (0.63 \pm 0.38 wt.%, n = 34) and P3 (0.12 ± 0.07 wt.%; Table 4).

Discussion

Magmatic duration and timing of ore formation at Red Chris

Four phases of the Red Stock, P1, P2E, P2L, and P3, have been dated at 211.6 ± 1.3 Ma, 206.0 ± 1.2 Ma, 203.6 ± 1.8 Ma, and 201.7 ± 1.2 Ma, respectively, spanning a period of ~ 10 m.y. (Fig. 6; Digital Appendix Table A2). These ages are consistent with crosscutting relationships described by Rees et al. (2015). Re-Os dating of molybdenite intergrown with chalcopyrite in quartz veins yielded a weighted mean age of 206.1 ± 0.5 Ma (Fig. 7), reflecting the timing of mineralization and in good agreement with the emplacement age of the mineralized P2E quartz monzonite porphyry. Ore formation occurred at a relatively late stage in the ~ 10 m.y. evolution of the Red Stock, and over a relatively brief period time (<1 m.y., as defined by the uncertainty on the average of three Re-Os analyses). This relatively late and short duration of mineralization is similar to the timing observed in many other porphyry deposits (e.g., Cathles et al., 1997; Masterman et al., 2004; von Quadt et al., 2011; Chiaradia et al., 2013; Correa et al., 2016; Li et al., 2017).

The age of mineralization at ~206 Ma is contemporaneous with the major pulse of mineralization in the Stikine and Quesnel terranes, representing a particularly fertile preaccretionary magmatic event (Nelson and Colpron, 2007; Nelson et al., 2013; Logan and Mihalynuk, 2014).

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Whole-rock major and trace element compositions of P1 to P3 porphyry rocks are almost identical to each other, and show characteristics of subduction-related magmatic rocks with high LILE and low HFSE concentrations (Figs. 9 and 10; Hawkesworth et al., 1993; Pearce, 1996). The high Sr/Y ratios (53 \pm 23, n = 12) and slightly positive Eu anomalies (Eu_n/Eu_n* = 1.1 ± 0.15 ; n = 14) indicate the suppression of plagioclase crystallization and early crystallization of amphibole from hydrous and oxidized magmas (Richards and Kerrich, 2007; Loucks, 2014). Rees et al. (2015) concluded from their analysis of petrochemistry that the Red Stock evolved from subalkaline (P1, P2) to marginally alkaline (P3), and classified the overall system as high-K calc-alkalic. All the porphyritic rocks from P1 to P3 show mantle-like, relatively low (87Sr/86Sr)_t (0.7044 ± 0.0002) and high $\mathcal{E}_{Nd}(t)$ (+3.1 ± 0.4) values, and similar single-stage Nd model ages (0.80 \pm 0.05 Ga). Zircon crystals from these rocks yield positive zircon $\mathcal{E}_{Hf}(t)$ values (+8.1 to +14.8) and single-stage Hf model ages of 375 ± 52 Ma (Figs. 11 and 12). These data indicate minimal involvement of ancient crustal components in the petrogenesis of these magmas, as observed for other Mesozoic igneous rocks in the Stikine terrane, and are consistent with an oceanic island arc setting (Samson et al., 1989; Logan and Mihalynuk, 2014). Late basaltic to andesitic dikes have similar Nd–Sr isotopic compositions to the porphyritic rocks, but show distinct trace element and REE patterns, and lower Sr/Y and La/Yb ratios, suggesting that they are not cogenetic.

Oxygen fugacity and sulfur-chlorine-water contents of the porphyry magmas

Magmatic oxidation state, and sulfur, chlorine, and water contents are fundamental factors for the formation of porphyry Cu deposits (e.g., Burnham, 1979; Candela, 1992; Richards, 2003; Loucks, 2014). It is difficult to measure the original magmatic oxidation state in altered plutonic rocks, but normalized Ce and Eu anomalies in zircons can be used to provide qualitative estimates (Ballard et al., 2002; Dilles et al., 2015; Lu et al., 2016). Zircon grains from the P1 to P3 porphyries display large positive Ce and small negative Eu anomalies (Eu_N/Eu_N* mostly > 0.4; Figs. 15 and 16), similar to zircons from other mineralized porphyry systems worldwide (Wang et al., 2014b; Dilles et al., 2015; Lu et al., 2016). These data are also consistent with estimates of magmatic oxidation state from early stage amphibole phenocrysts from the P2E and P2L porphyries, which yielded Δ FMQ values of 1.5 \pm 0.2 (n = 7) and 1.1 \pm 0.3 (n = 44). Consequently, we conclude that the four phases of magma in the Red Stock were all at least moderately oxidized. It is similarly difficult to estimate original magmatic water contents from altered plutonic rocks. However, the presence of amphibole phenocrysts in the P1 to P3 porphyries suggest that the magmas contained at least 4 wt. % H₂O (Burnham, 1979, Naney, 1983; Merzbacher and Eggler, 1984; Rutherford and Devine, 1988; Ridolfi et al., 2010). This observation is supported by: (1) high whole-rock Sr/Y ratios (53 \pm 23, n = 12), which reflect the suppression of plagioclase relative to amphibole crystallization under hydrous conditions (Moore and Carmichael, 1998; Richards and Kerrich, 2007); (2) high 10000×(Eu_N/Eu_N*)/Y (>1) and low Dy/Yb (<0.3) ratios in zircons (Digital Appendix Table A7; Lu et al., 2016); (3) high whole-rock V/Sc ratios (9.0 \pm 0.9, n = 14) (Loucks, 2014); and (4) estimates of magmatic water content from plagioclase phenocryst compositions, which indicate values of ~ 5 wt.% H₂O.

The compositions of igneous apatite inclusions in plagioclase and amphibole phenocrysts have been used to estimate magmatic sulfur and chlorine contents. The results suggest that apatite microphenocrysts from the P2E and P2L porphyries had higher S and Cl contents than for P1 and P3 (Fig. 18). The SO₃ contents in apatite are controlled by both magmatic sulfur concentration and oxidation state (Peng et al., 1997; Parat and Holtz, 2005; Webster and Piccoli, 2015). Sulfur occurs in the apatite structure mainly as S⁶⁺ and its content will increase in relatively oxidized environments (Boyce et al., 2010; Parat et al., 2011; Konecke et al., 2017b). The magmas associated with the four porphyry phases at Red Chris were all moderately oxidized, and therefore the differences in apatite sulfur content likely mainly reflect variations in the sulfur fugacity in the associated melts. The data indicate that the P2E and P2L porphyry magmas contained significantly higher S contents than the P1 and P3 magmas (Fig. 18). Models used to estimate magmatic sulfur content in equilibrium with apatite (Peng et al., 1997) are not strictly accurate because of uncertainties in the details of exchange reactions (Streck and Dilles, 1998; Mao et al., 2016) but are expected to be correct in relative terms. Our data yield higher values for the P2E and P2L porphyries (0.02–0.03 wt. % S) compared to estimates for P1 (0.01 wt. % S) and P3 (0.003 wt. % S). Consequently, we conclude that the syn-mineral porphyry (P2E and P2L) magmas were more S-rich than the premineralization P1 and especially the post-mineralization P3 porphyry magmas. Similarly, the magmatic Cl contents calculated from apatite compositions in the P2E and P2L porphyries are also higher than in the P1 and P3 porphyries, whereas F contents are lower (Table 4). Volatile species such as S, Cl, and F are differentially affected by degassing during ascent and crystallization of magmas: S and Cl will be preferentially

lost to the vapor phase relative to F (Webster et al., 2014; Stock et al., 2016), leading to high F/Cl and F/S ratios in degassed magmas. The data shown in Figure 18 suggest that the P2E and P2L were S-Cl-rich, whereas the P1 and P3 were S-Cl-poor but enriched in F. These differences could reflect either loss of volatiles from the source magma chamber (presumed to be an underlying batholith) prior to emplacement of the P1 and P3 magmas but not in the case of P2, or conversely addition of a volatile phase or a volatilerich magma to the magma chamber immediately prior to emplacement of P2. Volatile loss is expected to be a general condition of upper crustal magma emplacement, and this may well be a factor contributing to the low Cl/F ratios in the late P3 magmas, as well as the negatively correlated trend for F and Cl in P1. The P2 magma was emplaced after P1, and so might be expected to be similarly degassed if this represents a comagmatic sequence. However, the 6–8 m.y. age difference between P1 (211.6 \pm 1.3 Ma) and P2 (P2E: 206.0 ± 1.2 Ma; P2L: 203.6 ± 1.8 Ma) indicates that P1 and P2 are not directly comagmatic. One explanation is that the P2 event represents a pulse of S-Cl-rich magma injected at a relatively late stage into the mid-crustal magma reservoir, leading to a pulse of fluid exsolution and injection of P2 magma into the shallower level Red Stock, where syn-P2, quartz-vein hosted mineralization formed the Red Chris Cu-Au deposit. The post-mineralization P3 (201.7 \pm 1.2 Ma) and P4 porphyries may represent the final stages of intrusive activity emanating from the now degassed underlying magma chamber. Although both P2E and P2L porphyries are S-Cl-rich, apatites from P2L are marginally the most enriched. Samples of P2L are also the most mafic (SiO₂-poor) in the suite, and it is therefore tempting to speculate that the source magma chamber was recharged with more mafic, S-Cl-H₂O-rich magma during the P2 stage. The solubility of

S and Cl is higher in oxidized mafic magmas compared to felsic melts (> 1000 ppm S and > 3000 ppm Cl; Carmichael and Ghiorso, 1986; Webster, 1997; Webster et al., 1999; Hattori and Keith, 2001; Jugo et al., 2005; Jugo, 2009; Chelle-Michou and Chiaradia, 2017). Therefore, a key step in porphyry ore formation at Red Chris may have been recharge of the batholithic system by a less evolved, although broadly cogenetic, magma (cf. Hattori and Keith, 2001; Steinberger et al., 2013; Large et al., 2018). Support for this recharge model is provided by high magmatic temperatures calculated from zircon grains from some P2 samples (Fig. 16), and reverse zoning observed in plagioclase phenocrysts from the P2E and P2L porphyries, but which is not observed in P3 (Fig. 14C). In detail, zircons from the P2E porphyry show the highest Tiin-zircon temperatures and Eu_N/Eu_N* values, and P3 the lowest values. This might reflect an influx of higher temperature melt during the evolution of the P2E magma, and greater degrees of plagioclase fractionation from the late P3 magma. The compositional ranges of up to ~19 mol % An for plagioclase are higher than the upper limit caused by chemical diffusion (~10 mol % An; Pearce and Kolisnik, 1990). Furthermore, these reverse zoning patterns are matched by FeO contents in the phenocrysts (Figs. 14A and 14B), which suggests that the zonation reflects changes in magma composition (as opposed to simply changes in pressure, temperature, or water content; Ginibre and Wörner, 2007; Lange, 2009; Ustunisik et al., 2014; Waters and Lange, 2015). We therefore interpret these

more evolved resident magma.

Conclusions

changes to reflect late-stage (rim) growth from a more mafic magma that mixed into a

Zircon U-Pb ages indicate that the Red Stock was emplaced over a period of ~10 m.y. $(211.6 \pm 1.3 \text{ Ma}, 206.0 \pm 1.2 \text{ Ma}, 203.6 \pm 1.8 \text{ Ma}, \text{ and } 201.7 \pm 1.2 \text{ Ma} \text{ for the P1}, P2E, P2L, and P3 porphyries, respectively). The ore-forming event at Red Chris occurred relatively late in this magmatic history, synchronous with P2E at <math>206.1 \pm 0.5 \text{ Ma}$, and over a short period of <1 m.y. (as constrained by the error on the average of three Re-Os analyses). The four phases of porphyry and their related magmas were chemically fairly homogeneous and isotopically primitive, consistent with an island arc origin. The magmas were all moderately oxidized and hydrous (~5 wt. H₂O), but the P2 magmas had distinctly higher S and Cl contents. Combined with evidence for reverse zonation in plagioclase phenocrysts from P2 porphyries, and more mafic compositions in P2L, we suggest that the deeper source magma chamber was recharged at the time of emplacement of the P2 porphyries by injection of a more mafic S-Cl-rich magma. This recharge process may have triggered ore formation by causing voluminous exsolution of metalliferous hydrothermal fluids.

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Figure Captions

Fig. 1. Major terranes in the south of the Canadian Cordillera, showing Triassic to Jurassic magmatic belts and major associated porphyry deposits in the Stikine and Quesnel terranes (modified from Nelson and Colpron, 2007; Nelson et al., 2013; original graphics file provided by Joanne Nelson, British Columbia Geological Survey of Canada). The age of the Red Chris deposit is from this study (Table 1), and the ages for the other porphyry deposits are from Brown and Kahlert (1986; Red Mountain), Mortensen et al. (1995; Mt. Polley), Scott et al. (2008; Schaft Creek), Duuring et al. (2009; Kemess South), Taseko Mines Limited (2013; Gibraltar), Bath et al. (2014; Lorraine), Byrne and Tosdal (2014; Galore Creek), Devine et al. (2014; Lorraine), Logan and Mihalynuk (2014; Highland Valley, Copper Mountaine, Afton/Ajax, and Brenda), and Jago et al. (2014; Mt. Milligan). Fig. 2. Simplified geological map of the Red Stock and Red Chris Cu-Au deposit, showing the main mineralized zoned (named) and the locations of sampled drill holes (modified from Rees et al., 2015). Universal Transverse Mercator coordinates are based on the WGS84 datum. Fig. 3. Hand specimen photographs of samples of the P1, P2E, P2L, and P3 porphyry intrusions, and two late basaltic to andesitic dikes. (A) P1 porphyry with anhedral to subhedral altered amphibole and plagioclase phenocrysts, crosscut by pyrite-quartz veins (sample RC13-35). (B) P2E porphyry showing crowded texture with chloritized

amphibole and plagioclase phenocrysts; plagioclase grains are rimmed or replaced by

secondary K-feldspar. The brick-red color is due to fine-grained hematite (sample RC13-107). (C) P2L porphyry composed of abundant fresh amphibole and plagioclase phenocrysts, with quartz in the groundmass (sample RC13-32). (D) The P3 porphyry is similar to P2L, but is distinguished by the absence of quartz in the groundmass (sample RC13-78). (E) Andesitic dike with anhedral to subhedral amphibole phenocrysts, crosscut by a small calcite vein (sample RC13-62). (F) Basaltic dike with chloritized amphibole phenocrysts (sample RC13-106). See Digital Appendix Table A1 for sample locations.

Fig. 4. West-southwest–east-northeast cross section A–A', and north-northwest–south-southwest cross section B-B' (location of sections shown in Fig. 2), modified from Gillstrom et al. (2012) and Rees et al. (2015). Copper equivalent-grade zones, drill holes, and the boundary between potassic and post-potassic zones are shown: Cu equivalent (%) = Cu (%) + $0.486 \times Au$ (g/t).

Fig. 5. Hand specimen photographs and photomicrographs of alteration and vein minerals. (A, B) P2E porphyry (samples RC13-81 and RC13-75) with potassic alteration and A-type quartz veins comprising magnetite, secondary K-feldspar, and disseminated bornite and pyrite. Potassic alteration is characterized by secondary K-feldspar veins and selvedges around A-veins. Late unmineralized carbonate veins cut the A veins. (C) Amphibole phenocryst altered to secondary biotite, which has then been altered to chlorite, reflecting potassic alteration overprinted by chlorite—sericite alteration (plane-polarized transmitted light; P2E: sample RC13-30). (D) Plagioclase phenocryst partially overprinted by sericite (cross-polarized transmitted light; P2E porphyry: sample RC13-

11). (E) B-type quartz veins with centerline pyrite in P2L porphyry (sample RC13-44). (F) Pyritic D vein in P1 porphyry (sample RC13-40). (G) High-grade ore in sheeted A-type quartz-chalcopyrite-K-feldspar veins (sample RC13-31 in P2E porphyry). (H) Quartzcarbonate-pyrite-molybdenite-chalcopyrite vein in P2E porphyry (reflected light; sample RC13-88). Abbreviations: Amp = amphibole; Bi = biotite; Bn = bornite; Cbn = carbonate; Chl = chlorite; Cpy = chalcopyrite; Kfs = K-feldspar; Mo = molybdenite; Mt = magnetite; Pl = plagioclase; Py = pyrite; Qtz = quartz. See Digital Appendix Table A1 for sample locations. Fig. 6. Zircon U–Pb Tera-Wasserburg concordia diagrams for (A) P1, (B) P2E, (C) P2L, and (D) P3 porphyry samples dated by LA-MC-ICP-MS. Uncertainty ellipses and calculated ages are shown at 2σ . Fig. 7. Weighted mean Re-Os model age of three molybdenite vein samples from the Red Chris Cu-Au deposit. Fig. 8. Zr/Ti vs. Nb/Y discrimination diagram (Winchester and Floyd, 1977) for porphyry and basaltic-andesitic dike samples from Red Chris. Fig. 9. Selected whole-rock major element variation diagrams for porphyry and basaltic andesitic dike samples from Red Chris: (A) K₂O₂, (B) Na₂O₃, (C) TiO₂, (D) Al₂O₃, (E) total Fe₂O₃, and (F) MgO vs. SiO₂.

Fig. 10. (A) Primitive mantle-normalized trace element, and (B) chondrite-normalized rare earth element diagrams for porphyry and basaltic-andesitic dike samples from Red Chris. The normalization values for primitive mantle chondrite are from from Sun and McDonough (1989). Fig. 11. $\mathcal{E}_{Nd}(t)$ vs. initial ${}^{87}\text{Sr}/{}^{86}\text{Sr}$ ratios for porphyry and basaltic–andesitic dike samples from Red Chris, calculated at t = 200 Ma. All the samples fall in the field of Mesozoic igneous rocks in the Stikine island arc terrane, clearly different from the Late Cretaceous to Eocene plutons in the Northern Coast Plutonic Complex, which were derived from evolved crust. The depleted MORB mantle field is from Pilet et al. (2011); the Stikinia Mesozoic igneous rock field is from Samson et al. (1989); the Northern Coast Plutonic Complex field is from Samson et al. (1991) and Patchett et al. (1998); all data are recalculated at 200 Ma. Fig. 12. Histogram and relative probability curve for zircon $\mathcal{E}_{Hf}(t)$ values from (A) P1, (B) P2E, (C) P2L, and (D) P3 porphyry samples. Fig. 13. Classification diagram and plots of oxidation state, temperature, pressure, and magmatic water content estimated from amphibole compositions from P2E and P2L porphyry samples at Red Chris. A. ${}^{C}(Al^{VI} + Fe^{3+} + 2Ti^{4+})$ (apfu) vs. ${}^{A}(Na^{+} + K^{+})$ (apfu). B. ΔFMQ vs. temperature. C. ΔFMQ vs. pressure. D. ΔFMQ vs. magmatic water content.

The classification diagram for calcic amphibole is given by the Excel spreadsheet of

Locock (2014), and the superscript C and A represent C and A cations following the general amphibole formula (AB₂C₅T₈O₂₂W₂), respectively. Note that potassic-magnesiohastingsite is included in the field of magnesio-hastingsite. ΔFMQ values, temperatures, and magmatic water content were calculated from the spreadsheet of Ridolfi et al. (2010). Amphibole crystallization pressures were calculated using the equation of Mutch et al. (2016). The Δ FMQ values were calculated following the equation of Myers and Eugster (1983): $\log fO_2 = -24,441.9$ /T (K) + 8.290 (± 0.167). Abbreviation: apfu = atoms per formula unit. Fig. 14. Photomicrographs (cross-polarized transmitted light), Backscattered eelectron (BSE) images, and electron microprobe analysis profiles for FeO and anorthite proportion (X_{An}) for representative plagioclase crystals from the P2E, P2L, and P3 porphyries at Red Chris. Red circles on photomicrographs and white circles on BSE images denote the analyzed spots. The error bars for X_{An} and FeO analyses are smaller than the size of the symbols. Abbreviations: Amp = amphibole; Ap = apatite; Pl = plagioclase; Ser = sericite. Fig. 15. Chondrite-normalized REE patterns for zircons from (A) P1, (B), P2E, (C) P2L, and (D) P3 porphyry samples from Red Chris. Normalization values are from Sun and McDonough (1989). Fig. 16. Zircon Eu_N/Eu_N* vs. temperature diagram. Eu_N/Eu_N* is the europium anomaly,

calculated as $Eu_N/Eu_N^* = Eu_N/(Sm_N \times Gd_N)^{0.5}$, using the chondrite normalization values

of Sun and McDonough (1989). Oxidized suites have zircon Eu_N/Eu_N* values >0.4 (Dilles et al., 2015).

Fig. 17. Backscattered electron images of apatite crystals in samples from (A) P1 (RC13-39), (B) P2E (RC13-107), (C) P2L (RC13-33), and (D) P3 (RC13-78). Concentrations of SO₃ and Cl in apatite crystals are shown in wt. % (SO₃/Cl); red circles represent the analyzed spots. Higher concentrations are observed in apatites from P2E and P2L; some apatite microphenocrysts from P2 porphyries show zoning from SO₃-Cl-rich cores to SO₃-Cl-poorer rims (C).

Fig. 18. Plots of (A) S, and (B) F vs. Cl contents for apatite microphenocrysts from P1 to P3 porphyry samples at Red Chris. Abbreviation: apfu = atoms per formula unit. Data from Digital Appendix Table A8.

- 1 Elevated magmatic sulfur and chlorine contents in ore-forming magmas at the Red
- 2 Chris porphyry Cu-Au deposit, Northern British Columbia, Canada
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- 14 Abstract
- The Red Chris porphyry Cu-Au deposit is located in the Stikinia island-arc terrane in
- northwest British Columbia, it It is hosted by the Red Stock, which has four phases of
- porphyry intrusions: P1, P2E, P2L, and P3. New U-Pb dating of zircon shows that these
- intrusions were emplaced at 211.6 \pm 1.3 Ma (MSWD = 0.85), 206.0 \pm 1.2 Ma (MSWD =
- 19 1.5), 203.6 ± 1.8 Ma (MSWD = 1.5), and 201.7 ± 1.2 Ma (MSWD = 1.05), respectively.
- The ore-forming event at Red Chris was a short-lived event at 206.1 \pm 0.5 Ma (MSWD =
- 21 0.96; weighted average age of three Re-Os analyses), implying a duration of <1 m.y., as
- defined by the uncertainty range. This mineralization age coincides with the emplacement

age of the P2E porphyry, and is consistent with cross-cutting relationships that suggest P2E was the main syn-mineralization intrusion. Zircons from P1 to P3 porphyry rocks have consistently high Eu_N/Eu_N* ratios (mostly > 0.4), indicating that their associated magmas were moderately oxidized. The magmatic water contents estimated from plagioclase and amphibole compositions suggest H₂O contents of ~5 wt. %. Taken together, the P1 to P3 porphyries are interpreted to be moderately oxidized and hydrous. The four phases of porphyries are differentiated by sulfur and chlorine contents. The SO₃ contents of igneous apatite microphenocrysts from the mineralization-related P2 porphyries are higher (P2E: 0.30 ± 0.13 wt. %, n = 34; P2L: 0.29 ± 0.18 wt. %, n = 100) than those from the pre-mineralization P1 (0.11 \pm 0.03 wt. %, n = 34) and postmineralization P3 porphyries (0.03 \pm 0.01 wt. %, n = 13). The chlorine contents in apatite grains from the P2E and P2L porphyries are 1.18 ± 0.37 (n = 34) and 1.47 ± 0.28 wt. % (n = 100), also higher than those from P1 (0.51 \pm 0.3 wt. % Cl, n = 34) and P3 (0.02 \pm 0.02 wt. % Cl, n = 17). These results imply that the sulfur and chlorine contents of the P2E and P2L magmas were higher than in the P1 and P3 magmas, suggesting that elevated magmatic S-Cl contents in the P2 porphyries may have been important for oreformation. Although the process that caused the increase in sulfur and chlorine is not clear, reverse zoning seen in plagioclase phenocrysts from the P2 porphyry, and the occurrence of more mafic compositions in P2L suggest that recharge of the deeper magma chamber by a relatively S-Cl-rich mafic magma may have triggered the oreforming hydrothermal event.

46 Introduction

Compared to the relatively long-lived magmatic systems represented by a composite batholith (up to 10 m.y.; Matzel et al., 2006; Walker et al., 2007; Harris et al., 2008; Schaltegger et al., 2009; Paterson et al., 2011; Rezeau et al., 2016), the life spans of porphyry magmatic-hydrothermal ore-forming events are much shorter, probably on the order of several 105-hundred thousand years (up to 1 m.y.; Cathles et al., 1997; Hayba and Ingebritsen, 1997; Masterman et al., 2004; von Quadt et al., 2011; Weis et al., 2012; Chiaradia et al., 2013; Correa et al., 2016; Li et al., 2017). In many large magmatic systems, porphyry formation occurs at a relatively late stage in the system's evolution (Creasey, 1977; Titley and Beane, 1981; Candela, 1992; Richards, 2003; Rohrlach and Loucks, 2005; Yang et al., 2009; Correa et al., 2016). Although the multiple phases of porphyry stocks are commonly broadly cogenetic, they may be derived from packets of magma that evolved at different crustal levels over the history of the larger magmatic system (Annen et al., 2006). Understanding why ore-formation is only associated with a specific intrusive suite within these broader systems, and at discrete, commonly singular times, is a focus of this study. It is recognized that magmas with high sulfur (>1000 ppm), chlorine (>3000 ppm), and water (>4 wt. %) contents as well as relatively high oxidation states (higher than the fayalite-magnetite-quartz buffer, ΔFMQ , by 1–2 log fO_2 units) are fertile for the generation of magmatic-hydrothermal porphyry Cu deposits (Burnham, 1979; Candela, 1992; Richards, 2003, 2009, 2011, 2015; Wallace, 2005; Chambefort et al., 2008; Simon and Ripley, 2011; Chiaradia et al., 2012; Loucks, 2014; Hou et al., 2015; Lu et al., 2015, 2016; Chelle-Michou and Chiaradia, 2017). These ingredients are essential for the

transport of Cu (and Au) both in the magma (Zajacz et al., 2012) and in exsolved high temperature, SO₂-rich, saline magmatic fluids (Candela and Holland, 1984; Zajacz et al., 2011). However, whereas many magmas achieve such compositions, ore formation is a relatively rare and discrete event in such systems. Various mechanisms have been proposed that might trigger an ore forming event from a fertile magma source, including recharge of the magma chamber by hot, sulfur-rich melts (Hattori, 1993; Keith et al., 1997; Larocque et al., 2000; Hattori and Keith, 2001; Halter et al., 2005; Stern et al., 2007; Nadeau et al., 2010, 2016; Wilkinson, 2013; Tapster et al., 2016; Zhang and Audétat, 2017), fluxing by sulfur gases (Blundy et al., 2015), or increasing the water and oxidation state of magmas during long-term fractionation in deep magma chambers (Ballard et al., 2002; Wang et al., 2014a, 2014b; Dilles et al., 2015; Lu et al., 2015, 2016). The Red Chris porphyry Cu-Au deposit provides an opportunity to test these hypotheses, because ore-formation occurred at a discrete and relatively late stage in the ~10 m.y. history of the associated magmatic system. The Red Chris deposit is located in northwest British Columbia, and contains measured and indicated resources of 1,035 million metric tonnes of ore grading 0.35% Cu and 0.35 g/t Au (Gillstrom et al., 2012). Pre-, syn-, and post-mineralization porphyry intrusions have been recognized based on detailed core logging and petrographic work (Rees et al., 2015), with mineralization occurring in a singular episode during this magmatic history. In this paper, we present detailed geochronological and geochemical data for the porphyry phases at Red Chris, and show that they have similar bulk

compositions, including all being relatively hydrous and oxidized. However, the syn-

mineralization porphyry is characterized by plagioclase with reverse zoning and apatite with relatively high sulfur and chlorine contents. We interpret these data to indicate that high magmatic S and Cl contents, in addition to high H₂O contents and oxidation state, were critical for Cu-Au ore-formation at Red Chris. We speculate that there was an injection of relatively mafic (hotter, and more S- and Cl-rich) but cogenetic magma into the mid–upper crustal source magma chamber approximately coincident with emplacement of the syn-mineralization porphyry.

Geological Setting

Red Chris is situated within the island-arc terrane of Stikinia in the Intermontane Belt of the Canadian Cordillera of British Columbia (Fig. 1; Monger and Irving, 1980; Nelson and Colpron, 2007; Nelson et al., 2013). Stikinia consists primarily of Mesozoic arcrelated igneous and sedimentary rocks, formed in response to subduction processes prior to accretion to the ancestral North American margin in the Middle Jurassic (Gabrielse, 1991; Ricketts et al., 1992; Lang et al., 1995; Mihalynuk et al., 2004; Nelson and Colpron, 2007; Logan and Mihalynuk, 2014). In northwestern Stikinia, arc assemblages comprise the Middle to Late Triassic Stuhini Group, unconformably overlain by Late Triassic and Early to Middle Jurassic volcanic and sedimentary rocks of the Hazelton Group (Fig. 2; Brown et al., 1991; Gabrielse, 1991; Marsden and Thorkelson, 1992; Dostal et al., 1999; Gagnon et al., 2012; Nelson et al., 2013; Logan and Mihalynuk, 2014). The Hazelton Group is overlain by sedimentary rocks of the syn- to post-accretion Middle Jurassic to Early Cretaceous Bowser Lake Group.

Several porphyry Cu±Au±Mo deposits occur in the region, hosted by Late Triassic to Early Jurassic arc-related plutons (Fig. 1), including Red Chris which formed in the Late Triassic Red Stock. Collectively, these regional intrusions have ages ranging from ~222 Ma to ~180 Ma, recording much of the pre-accretionary history of Stikinia (Lang et al., 1995; McMillan et al., 1995; Scott et al., 2008; Nelson et al., 2013; Logan and Mihalynuk, 2014). A significant percentage of the known porphyry Cu deposits, including Red Chris, formed during a 6-m.y. pulse of magmatism between 206 and 200 Ma, with compositions ranging from calc-alkaline to strongly alkaline (Lang et al., 1995; McMillan et al., 1995; Nelson et al., 2013; Bissig and Cooke, 2014; Logan and Mihalynuk, 2014; Micko et al., 2014; Pass et al., 2014).

Geology of the Red Chris Cu-Au deposit

The Red Chris Cu-Au deposit was discovered in the 1950s and explored intermittently in subsequent decades, with mining beginning in 2015 (Ash et al., 1995, 1996; Newell and Peatfield, 1995; Baker et al., 1997; Gillstrom et al., 2012; Rees et al., 2015). The deposit is hosted by the Red Stock, which is the largest of a suite of Late Triassic to Early Jurassic stocks and dikes that intrude the Stuhini Group in the district (Fig. 2; Friedman and Ash, 1997; Rees et al., 2015). The stock is tabular, elongate in an east to northeast direction, and approximately 6.5 km long by 300 to 1500 m wide (Fig. 2; Ash et al., 1995; Baker et al., 1997; Gillstrom et al., 2012). It has a steep northern contact against Stuhini Group country rocks, but its southeastern margin against Hazelton and Bowser Lake Group strata is poorly exposed, and has locally been truncated by the NE-

trending, steeply SE-dipping South Boundary fault (Fig. 2). This fault, and the East zone fault within the stock, probably reflect a long-lived and deep structure which guided the emplacement of the intrusions, mineralization, and subsequent deformation of the Red Stock.

The Red Stock is a composite intrusion consisting of several phases of porphyritic diorite to quartz monzonite. Mineralogically, the rocks consist of medium- to coarse-grained amphibole, plagioclase, and minor biotite phenocrysts, with K-feldspar, plagioclase, and quartz in the groundmass. Based on compositional and textural differences and crosscutting relationships, Rees et al. (2015) identified four distinct porphyry phases, P1, P2, P3, and P4. The P1 porphyry is a pre-mineralization leucodiorite which accounts for the main volume of the Red Stock, and is distinguished by sparse anhedral to subhedral amphibole (~10%) and abundant plagioclase (30–40 vol. %) phenocrysts with lengths up to 4 mm. The groundmass is composed of fine-grained plagioclase and minor quartz (Fig. 3A).

The P2 porphyry is a syn-mineralization quartz monzonite intrusion which is largely unexposed at surface but is observed in drill core to have intruded P1 at depth. It has been subdivided by Rees et al. (2015) into early (P2E), intermediate (P2I), and late (P2L) phases based on vein truncations and chilled margins. The P2 porphyries are generally characterized by tabular subhedral to euhedral amphibole (10–15 vol. %) and plagioclase (35–50 vol. %) phenocrysts. The amphibole crystals are mostly euhedral and larger than in P1, with lengths up to 10 mm. The groundmass comprises K-feldspar, plagioclase, and quartz. In this paper, the P2 subphases are simplified to an early stage (P2E; Fig. 3B) and late stage porphyry (P2L, probably corresponding to P2I and P2L of Rees et al., 2015;

Fig. 3C). The P2E porphyry has a crowded plagioclase texture with slightly smaller amphibole phenocrysts than P2L. In contrast, the P2L porphyry is relatively fresh and occurs only as small dikes (Fig. 4).

The post-mineralization P3 monzonite porphyry is much less abundant. It is texturally similar to the P2L porphyry with abundant amphibole phenocrysts (15–20 vol. %), but is distinguished by the absence of quartz in the groundmass, which is mainly composed of K-feldspar and plagioclase (Fig. 3D). Although no crosscutting relationships between the P3 and P2L porphyries have been found, the zircon U-Pb dating results (see below) confirm that P3 is younger. The P4 porphyry occurs as rare dikes and is typified by sparse fine-grained amphibole phenocrysts (Rees et al., 2015). No P4 samples were included in this study.

Several younger basaltic to andesitic dikes with sparse amphibole phenocrysts cut the Red Stock and the Stuhini Group host rocks (Figs. 3E and F). They postdate the porphyry-stage Cu-Au mineralization (Baker et al., 1997; Rees et al., 2015), but are mildly to strongly altered (Figs. 3E and F) and are crosscut by late quartz-calcite-pyrite veins.

Hydrothermal alteration

Alteration at Red Chris has been described previously by Baker et al. (1997),

Gillstrom et al. (2012), Norris (2012) and Rees et al. (2015). The alteration assemblages

are typical of calc-alkaline porphyry Cu systems (Lowell and Guilbert, 1970; Seedorff et

al., 2005; Sillitoe, 2010), and consists of early stage potassic alteration, overprinted by

chlorite–sericite, sericitic (phyllic), intermediate argillic, and minor late stage propylitic

Vein styles and mineralization

alteration (see paragenetic diagram in Rees et al., 2015) Alteration at Red Chris is typical of calc-alkaline porphyry Cu systems (Lowell and Guilbert, 1970; Seedorff et al., 2005; Sillitoe, 2010), and consists of early stage potassic alteration, overprinted by chloritesericite, sericitic (phyllic), intermediate argillic, and minor late stage propylitic alteration (see paragenetic diagram in figure 13 of Rees et al., 2015; Baker et al., 1997; Gillstrom et al., 2012; Norris, 2012). Potassic alteration is expressed by replacement of amphibole phenocrysts by secondary biotite, plagioclase replaced or rimmed by secondary Kfeldspar, and by K-feldspar veins (Figs. 3B and 5A-C). It is best preserved in the deeper levels of the deposit where it is spatially associated with the syn-mineralization P2E porphyry; however it locally extends into pre-mineralization P1 porphyry wall rocks (Fig. 4). The P2L and P3 porphyries are were only weakly affected by potassic alteration (Rees et al., 2015). Chlorite-sericite alteration is characterized by chlorite replacing secondary biotite (Fig. 5C) and sericite replacing feldspar (Fig. 5D). At shallower levels in the system, potassic alteration is completely overprinted by phyllic and intermediate argillic alteration (Gillstrom et al., 2012), characterized by sericite after plagioclase (phyllic; Fig. 5D), and illite and kaolinite (intermediate argillic; Norris, 2012). This lower temperature alteration overprint affects all the porphyry phases, but is less pervasive at depth. Propylitic alteration at Red Chris is mainly observed as minor chlorite and epidote in the outer part of the Red Stock, and extends for 100 to 200 m into the Stuhini volcanic country rocks (Gillstrom et al., 2012; Norris, 2012; Rees et al., 2015).

Detailed descriptions of vein styles at Red Chris have been given by Norris (2012) and Rees et al. (2015). A-type quartz veins (Gustafson and Hunt, 1975) and stockworks are associated with potassic alteration and host the bulk of the copper-gold mineralization. These veins are most intensely developed around the apex of the principal P2 porphyry body, but extend for hundreds of metres into the P1 wall rocks (see figure 13 in Rees et al., 2015). Typical A veins contain K-feldspar, biotite, chalcopyrite, bornite, and magnetite, with K-feldspar alteration halos (Figs. 5A–B and G). Copper sulfides also occur as disseminations in the host porphyry. Bornite is more abundant in the apex of P2E, and progressively decreases outwards where chalcopyrite is the dominant Cusulfide (Norris, 2012; Rees et al., 2015). At shallower depths in the preserved system, early bornite was sulfidized to chalcopyrite, and pyrite becomes increasingly dominant. Rees et al. (2015) delineated a high-sulfur contour (>4% S) in section above which total sulfide (dominantly pyrite) ranges from 4 to 10%. Microscopic native gold and electrum occur as inclusions in bornite (Rees et al., 2015). The grades of Cu and Au are positively correlated with quartz vein density (Gillstrom et al., 2012). In high-grade zones at depth (e.g., 4.12 % Cu and 8.83 g/t Au in hole 09-350 from 540 to 692.5 m down-hole depth), vein abundance exceeds 80 vol. % in sheeted arrays (Fig. 5G; Rees et al., 2015). B- and D-type veins (Gustafson and Hunt, 1975) are relatively minor at Red Chris, and host only minor amounts of Cu sulfides. B quartz veins are characterized by relatively straight margins with sulfide centerlines (pyrite and minor chalcopyrite, and locally molybdenite; Fig. 5E). Pyritic D veins have variable widths (1 to 10 mm; Fig. 5F). Carbonate and minor chlorite veins cut all the earlier veins and are generally barren (Norris, 2012).

Sampling and Analytical Methods

Samples of the P1, P2E, P2L, and P3 porphyries were collected from drill core, and descriptions and locations are listed in Digital Appendix Table A1; sampled drill hole locations are also shown on Figure 2. Fourteen least-altered samples of the porphyry intrusions and three samples of late basaltic to andesitic dikes were selected for whole-rock geochemical analysis. Eight of these samples were selected for determination of Nd-Sr isotopes. Three samples of quartz-carbonate-pyrite-molybdenite-chalcopyrite veins (Fig. 5H) were collected for Re-Os dating. Four samples of the P1 (RC13-40), P2E (RC13-107), P2L (RC13-33), and P3 (RC13-78) intrusions were selected for zircon U-Pb dating, Hf isotopic, and trace element analyses. Details of analytical methods are provided in Appendix 1.

Electron microprobe analyses

Primary igneous minerals such as plagioclase and amphibole in the porphyry rocks are widely altered to K-feldspar, sericite, and chlorite (Figs. 3A–B and 5C–D). However, a few least-altered samples of the P2E, P2L, and P3 porphyries contained unaltered plagioclase and amphibole grains. In addition, igneous apatite grains were typically preserved as inclusions within plagioclase and amphibole phenocrysts. Compared with hydrothermal apatite (acicular crystals intergrown with other hydrothermal minerals such as quartz, sericite, chlorite, and sulfides), igneous apatite grains typically showed stubby prismatic habits, as described by Richards et al. (2017). Detailed analytical methods are described in Appendix 1.

Re-Os molybdenite dating

Three samples of molybdenite from quartz-carbonate-pyrite-molybdenite-chalcopyrite veins were collected from the Gully zone (RC13-88 and RC13-103), and the East zone (RC13-82; Digital Appendix Table A1; Fig. 2). A molybdenite mineral separate was produced for each sample by metal-free crushing followed by gravity and magnetic concentration methods. Dating was conducted at the Canadian Centre for Isotopic Microanalysis at the University of Alberta, Canada, using methods described in Appendix 1.

Geochronological Results

Zircon U-Pb ages of the Red Stock

Zircon U-Pb results are presented in Digital Appendix Table A2 and illustrated in Figure 6; all ages are illustrated and reported with 2σ errors. All the zircons show oscillatory zoning under BSE imaging. Sample RC13-40 was collected from P1 leucodiorite porphyry (Digital Appendix Table A1). Analyzed zircon grains form a tightly clustered age population, mostly with low common lead contents. Except for one inherited or xenocryst zircon (apparent $^{206}\text{Pb}/^{238}\text{U}$ age = 261 ± 12 Ma), the twenty-seven grains yielded an intercept age of 211.6 ± 1.3 Ma (MSWD = 0.85; Fig. 6A), similar to the weighted mean $^{206}\text{Pb}/^{238}\text{U}$ age of 211.8 ± 1.3 Ma (MSWD = 0.8). Sample RC13-107 was collected from P2E quartz monzonite porphyry (Digital Appendix Table A1). One xenocrystic zircon grain yielded an older age (apparent $^{206}\text{Pb}/^{238}\text{U}$ age = 316 ± 10 Ma), but the other twenty-nine zircons contained low amounts

of common lead and intersected the concordia line at 206.0 \pm 1.2 Ma (MSWD = 1.5; Fig.

6B), in good agreement with the weighted mean $^{206}\text{Pb}/^{238}\text{U}$ age of 205.9 ± 1.5 Ma (MSWD = 1.5).Sample RC13-33 was collected from P2L quartz monzonite porphyry (Digital Appendix Table A1). Two xenocryst grains were found among thirty-one analyzed zircons, with $^{206}\text{Pb}/^{238}\text{U}$ ages of 298 ± 16 Ma and 441 ± 13 Ma. The remaining twenty-nine zircon grains had low common lead contents and yielded an intercept age of 203.6 ± 1.8 Ma (MSWD = 1.5; Fig. 6C), similar to the weighted mean 206 Pb/ 238 U age of 203.3 \pm 1.5 Ma (MSWD = 1.04).Sample RC13-78 was collected from P3 monzonite porphyry (Digital Appendix Table A1). Thirty-two zircons with low common lead contents yielded an intercept age of 201.7 ± 1.2 Ma (MSWD = 1.05; Fig. 6D), and a weighted mean 206 Pb/ 238 U age of 201.6 ± 1.2 Ma (MSWD = 1.04). All of the zircons have magmatic textures, and the intercept ages above are interpreted to be the crystallization ages of the individual intrusions. The ages are consistent with crosscutting relationships described by Rees et al. (2015), and define a 10 m.y. span of magmatism from 211.6 \pm 1.3 Ma (P1) to 201.7 \pm 1.2 Ma (P3). The relative ages of the two samples of syn-mineralization P2 porphyry are consistent with crosscutting relationships, but the dates (P2E: 206.0 ± 1.2 Ma; P2L: 203.6 ± 1.8 Ma) overlap within the 2 σ uncertainty. Hence the apparent 2.4 m.y. age difference is not statistically robust, and their true ages may in fact be closer as suggested by their close relationship with ore

Re-Os molybdenite ages

mineral paragenesis.

The Re-Os model ages for three samples are shown in Table 1. They have relatively high rhenium contents ranging from 497.8 to 1771 ppm, with common 187 Os of 1078 to 3821 ppb. Although the three samples are from two different mineralization zones (i.e., East and Gully zones; Fig. 2), they yielded indistinguishable model ages within uncertainty: 206.5 ± 0.8 Ma, 205.7 ± 0.9 Ma, and 205.9 ± 1.1 Ma (± 2 s.d.). The small grain size of the molybdenite (< 1 mm) minimizes the risk of decoupling between Re and 187 Os in these samples (caused by diffusion: Selby and Creaser, 2004). The results yielded a weighted average age for all three samples of 206.1 ± 0.5 Ma (95% confidence level with MSWD = 0.96; Fig. 7). This age is consistent with the age of the synmineralization P2E porphyry (206.0 ± 1.2 Ma).

Geochemical and Isotopic Results

Whole-rock major and trace elements

Whole-rock major and trace element compositions for fourteen samples of the Red Stock and three basaltic—andesitic dike samples are listed in Table 2. All the P1 to P3 porphyry intrusions have relatively homogeneous major element compositions, but have moderate to high loss-on-ignition values (LOI: 2.3 to 10.9%) reflecting varying degrees of potassic and sericitic alteration. On an LOI-free basis, they are mostly intermediate composition (SiO₂ = 56.71–63.16 wt. %; Digital Appendix Table A3), and straddle the boundary between granodiorite (diorite) and syenite on a Zr/Ti versus Nb/Y diagram (Fig. 8). The late basaltic—andesitic dikes have distinct compositions, and plot in the field of diorite and gabbro on Figure 8. On Harker-type diagrams of SiO₂ versus selected major elements, the alkali elements (K₂O and Na₂O; Figs. 9A–B) show significant scatter,

likely due to hydrothermal alteration, as confirmed by a rough correlation between alkali contents and LOI. However, other elements such as Fe₂O₃, MgO, TiO₂, and Al₂O₃ show roughly correlated trends with SiO₂ (Figs. 9E–F), suggesting a broadly cogenetic suite, with the exception of the late basaltic to andesitic dikes, which do not plot on the same trends for K_2O and TiO_2 . On primitive mantle-normalized extended trace element and chondrite-normalized rare earth element (REE) diagrams (Figs. 10A–B), the porphyritic rocks show largely indistinguishable patterns consistent with typical subduction-related igneous rocks (Hawkesworth et al., 1993; Pearce, 1996), characterized by large-ion lithophile element (LILEs: Rb, Ba, Th, U, K) and light rare earth element enrichments (LREE), low concentrations of high field strength elements (HFSE: Nb, Ta, Zr, Hf, and Ti), relative depletions in compatible elements and middle to heavy rare earth elements (MREE, HREE; La/Yb = 8.9 ± 1.7 , n = 14), and flat to listric-shaped patterns from MREE to HREE. Such listric patterns likely reflect amphibole fractionation from hydrous magmas, and are an indication of magma fertility for porphyry ore formation (Richards and Kerrich, 2007). Most samples have flat or slightly positive Eu anomalies with Eu_N/Eu_N* of 1.1 ± 0.15 (n = 14; Fig. 10B), also likely reflecting amphibole fractionation and lack of plagioclase fractionation. Three samples of late basaltic to andesitic dikes show distinct trace element patterns, especially for HFSE and REE, which are enriched relative to the porphyries; they are clearly not cogenetic with the earlier porphyries. Excluding two samples with high Sr values that may reflect calcite veining, the

porphyries have relatively high Sr/Y (53 \pm 23, n = 12) and V/Sc ratios (9.0 \pm 0.9, n = 14),

which overlap the range of fertile rocks for porphyry formation (Richards and Kerrich, 2007; Loucks, 2013, 2014). Whole-rock Nd-Sr isotopes Nd and Sr isotopic compositions of the porphyritic rocks are very homogeneous (Table 4 and Fig. 11). They show a narrow range of initial Sr isotopic ratios from 0.7042 to $0.7047 ((^{87}\text{Sr}/^{86}\text{Sr})_t = 0.7044 \pm 0.0002, n = 8)$, with relatively high $\mathcal{E}_{Nd}(t)$ values from 2.4 to 3.6 (average = $+3.1 \pm 0.4$, n = 8), consistent with a mantle source with minor crustal contamination. The single-stage Nd model ages (T_{DM1}) calculated following the methods of Goldstein et al. (1984) range from 0.74 to 0.88 Ga (average = 0.80 ± 0.05 Ga, n = 8), and are indistinguishable for the different porphyry phases. Two basaltic-andesitic dike samples show slightly lower Sr isotopic ratios ($(^{87}\text{Sr})^{86}\text{Sr}$)_t = 0.7042 and 0.7043) and higher $\mathcal{E}_{Nd}(t)$ ratios (+3.5 and +3.9) than the porphyry rocks (Fig. 11). Zircon Hf isotopes Zircon Hf isotopic results are listed in Digital Appendix Table A4, and illustrated in Figure 12. The four zircon samples from porphyritic rocks show indistinguishable Hf isotopic compositions and single-stage Hf model ages of 375 ± 52 Ma (range = 264-527Ma, n = 56), with relatively high $\mathcal{E}_{Hf}(t)$ values of 12.0 \pm 0.4 (weighted mean value, 95% confidence level; range = 8.1-14.8, n = 56). These data suggest that the porphyries shared

Amphibole, Plagioclase, Zircon, and Apatite Compositions

a common primitive mantle source, consistent with their island arc origin.

The compositions of amphibole, plagioclase, zircon, and apatite from samples of least-altered porphyritic rocks are listed in Digital Appendix Tables A5, A6, A7, and A8, respectively. Based on these analyses, water and sulfur contents as well as oxygen fugacity of the magma have been estimated qualitatively and quantitatively.

Amphibole compositions

Amphibole compositions can be used to estimate magmatic oxidation state, crystallization temperatures and pressures, as well as water contents (e.g., Ridolfi et al., 2010; Zhang et al., 2012; Wang et al., 2014b). However, fresh amphibole grains were only observed in the P2E and P2L porphyries. All of the analyzed grains are calcic amphiboles, and two generations of amphibole were recognized in both P2E and P2L. Early stage grains occur as phenocrysts and are characterized by higher Al contents (range = 9.35-12.86 and 8.79-13.46 wt. % Al₂O₃ for P2E and P2L, respectively; Digital Appendix Table A5); they are mostly classified as magnesio-ferri-hornblende and magnesio-hastingsite (including potassic-magnesio-hastingsite; Fig. 13A). The late stage grains were developed in the groundmass or as recrystallized phenocrysts, typified by lower Al concentrations (range = 1.62-4.76 and 4.33-7.53 wt. % Al₂O₃ for P2E and P2L, respectively; Digital Appendix Table A5). They are classified as magnesio-ferrihornblende or rare actinolite (Fig. 13A). The early stage amphibole grains are intergrown with plagioclase phenocrysts (Fig. 14), and igneous apatite crystals occur as inclusions within amphibole (Fig. 14B). This indicates that these three minerals are broadly coeval. The crystallization temperature, magmatic water contents, and oxygen fugacities were estimated for the two stages of amphiboles using the spreadsheet of Ridolfi et al. (2010),

and the crystallization pressures were calculated following the equation of Mutch et al. (2016). The Al-in-hornblende geobarometer used is applicable to granitoids that contain amphibole, plagioclase, quartz, and alkali feldspar, broadly consistent with the mineral assemblage within P2 porphyry rocks. The calculated results are listed in Digital Appendix Table A5 and illustrated in Figure 13 (B–C). Amphibole phenocrysts from the P2E and P2L porphyry samples have similar compositions, with calculated crystallization pressures from 3.5 to 6.7 kbar (average = 4.6 ± 0.8 kbar, n = 51), temperatures from 855 to 983 °C (average = 900 ± 30 °C, n = 51), Δ FMQ values from 0.5 to 1.8 (average = 1.1 ± 0.3, n = 51), and H₂O contents in melts from 4.0 to 6.1 wt. % (average = 5.1 ± 0.4 wt. %, n = 51). Late stage (low-Al) amphiboles in P2E porphyry samples yielded crystallization pressures from 0.6 to 1.4 kbar (average = 1.1 ± 0.3 kbar, n = 12), temperatures from 637° to 774 °C (average = 719 \pm 34 °C, n =12), Δ FMQ values from 2.6 to 3.3 (average = 3 \pm 0.2, n = 12), and magmatic water contents from 3.6 to 5.1 wt. % (average = 4 ± 0.4 Ma, n = 12). Low-Al amphibole grains from P2L samples yielded slightly higher crystallization pressures (2.0 \pm 0.4 MPa; range = 1.3–2.6 kbar, n = 8) and temperatures (800 \pm 24 °C; range = 751-800 °C, n = 8), slightly lower Δ FMQ values (2.4 ± 0.3 ; range = 1.1-2.9, n = 8), but similar H₂O contents $(4.0 \pm 0.2; \text{ range} = 3.7-4.4, \text{ n} = 8)$ to those calculated for P2E. There are clear trends of increasing oxidation state and decreasing crystallization pressures, temperatures, and magmatic water contents from early to late stage amphiboles in both P2E and P2L porphyries (Figs. 13B–D). These trends are consistent with the

high-Al amphibole phenocrysts having crystallized at depth before final crystallization of

the magma (and low-Al amphibole) after emplacement at shallow levels (e.g., Rutherford and Devine, 2003). The decreasing magmatic water contents might be attributed to degassing during ascent and crystallization, which can also result in a small increase in oxygen fugacity (Mathez, 1984; Candela, 1986; Burgisser and Scaillet, 2007; Zimmer et al., 2010; Bell and Simon, 2011; Dilles et al., 2015).

Plagioclase compositions

Plagioclase phenocrysts from all P1 porphyry samples studied have been altered, but partially unaltered phenocrysts were found in some P2 and P3 porphyry samples. Analyses with $K_2O > 1$ wt. % are interpreted to reflect potassic alteration and have been excluded. Plagioclase grains from the P2E porphyry (sample RC13-13) are classified as oligoclase (\geq An₁₇) to labradorite (\leq An₇₀), with an average anorthite (An) content of 44 \pm 12 mol % (n = 28). Nearly all these crystals exhibited reverse zoning in An content, with compositional ranges up to ~16 mol \%, and FeO contents that correlate positively with X_{An} values (Digital Appendix Table A6; Fig. 14A). Plagioclase grains from the P2L porphyry (samples RC13-26 and RC13-33) have compositions similar to P2E (An mol % = 48 ± 5 , n = 38), and also show reverse zoning with amplitudes up to ~19 mol % An that correlate with FeO contents (Fig. 14B). Such reverse zoning with positive An-FeO correlations were observed both in partially altered (i.e., sericite alteration; Fig. 14A) and relatively fresh plagioclase crystals (Fig. 14B), indicating that it was not likely to have been generated by hydrothermal alterations. In contrast, plagioclase grains from the P3 porphyry (RC13-78) display relatively uniform An compositions (An mol $\% = 51 \pm 5$, n = 37), with no clear zoning in X_{An} and FeO (Fig. 14C).

Plagioclase compositions (An proportions) are very sensitive to dissolved water content and temperature of the melt, and thus can be used as a hygrometer if corrected for temperature (Mathez, 1973; Lange, 2009; Waters and Lange, 2015). We used apatite saturation temperatures (AST) following the formula of Piccoli and Candela (1994) to calculate magma temperatures for P2 and P3 samples (Table 4; Digital Appendix Table A3). The whole-rock SiO₂ and P₂O₅ values were assumed to approximate the melt contents at the time of crystallization, although this assumption can be challenged (Piccoli and Candela, 1994, 2002). Nevertheless, the calculated temperatures are reasonable for magmas of intermediate, hydrous composition, and are also in agreement with the estimated crystallization temperatures of early stage amphibole (Digital Appendix Table A5): 932 °C for P2E (RC13-13), 892 °C (RC13-26) and 905 °C (RC13-33) for P2L, and 928 °C for P3 (RC13-78; Table 4). The updated spreadsheet of Waters and Lange (2015) has been used to calculate magmatic water contents (Digital Appendix Table A9). The calculated values are relatively uniform and indistinguishable between the different porphyry phases: 5.2 ± 0.2 (n = 28) for P2E, 5.5 ± 0.2 (n = 38) for P2L, and 5.1 ± 0.1 wt. % (n = 37) for P3. For the P2E and P2L porphyries, the magmatic water contents calculated from plagioclase and early stage amphibole compositions agree well with each other (Digital Appendix Tables A5 and A8). Zircon trace element compositions

Twenty trace element spot analyses were obtained for zircons from samples of P1 to

P3 porphyries (Digital Appendix Table A7; Fig. 15). It is common to encounter small

0.9, average = 0.6 ± 0.1).

mineral inclusions, especially of apatite, titanite, in zircon during LA-ICP-MS analyses (e.g., Lu et al., 2016). We have taken Ca > 200 ppm or La > 0.3 ppm as an indication of apatite contamination, and Ti > 20 ppm to reflect titanite contamination, and such data were excluded (Digital Appendix Table A7). The remaining analyses have low LREE and elevated HREE contents, with small negative Eu and strongly positive Ce anomalies (Fig. 15). These REE patterns and total contents ($\Sigma REE = 351-1125$ ppm), as well as Th/U ratios above 0.1 (0.26–0.63) (Digital Appendix Table A7) are typical of igneous zircons from relatively oxidized magmas (Hoskin and Schaltegger, 2003). Zircon crystals from the P1 to P3 porphyries show slightly different calculated Ti-inzircon temperatures and Eu anomalies $(Eu_N/Eu_N^* = Eu_N/(Sm_N \times Gd_N)^{0.5})$; Fig. 16; Digital Appendix Table A7). Titanium-in-zircon temperatures were calculated using the equation of Ferry and Watson (2007), where it is assumed that $\log a_{SiO2} = 1$ because of the existence of quartz in the host porphyries, and $\log a_{TiO2} = 0.7$ due to the presence of titanite (a common accessory mineral in these rocks). Zircon crystals from the P2E porphyry (sample RC13-107) have the highest calculated temperatures (average of 760 \pm 29 °C; range = 723° – 819° C, n = 14) with relatively uniform Eu_N/Eu_N* values from 0.5 to 0.7 (average = 0.6 ± 0.1 ; n = 14). Zircon crystals from the P3 porphyry (sample RC13-78) show the lowest calculated temperatures (average = 726 ± 24 °C; range = 693°-758°C, n = 18) and lowest Eu_N/Eu_N^* ratios (0.5 ± 0.1; range = 0.3–0.6, n = 18). Zircon crystals from the P1 and P2L porphyries have intermediate calculated temperatures (P1: 736 \pm 26 °C, range = 699° -810 °C, n = 16; P2L: 730 ± 24 °C, range = 705° -795 °C, n = 17), and a similar wide range of Eu_N/Eu_N* values (P1: 0.3–1.0, average = 0.6 ± 0.2 ; P2L: 0.3–

Trace element compositions in zircon are sensitive to magmatic water content and oxygen fugacity, and have been used to distinguish between fertile and infertile suites in porphyry Cu ± Au ± Mo systems (Ballard et al., 2002; Liang et al., 2006; Qiu et al., 2014; Wang et al., 2014b; Dilles et al., 2015; Shen et al., 2015; Lu et al., 2016; Xu et al., 2016; Loader et al., 2017). In hydrous magmas, early plagioclase crystallization is suppressed (Moore and Carmichael, 1998), leading to no pronounced depletion of Eu in melts (Eu²⁺ substitutes for Ca²⁺ in plagioclase). On the other hand, in oxidized magmas, Eu is predominantly present as Eu³⁺ and is partitioned into zircon along with other REE (Ballard et al., 2002; Hoskin and Schaltegger, 2003; Trail et al., 2011; Dilles et al., 2015). Therefore, the small negative Eu anomalies in zircon samples (as indicated by Eu_N/Eu_N* ratios >0.3 and mostly > 0.4; Fig. 16A), and slightly positive anomalies in whole rocks (Fig. 10) indicate relatively high magmatic water contents and/or oxidation states (Ballard et al., 2002; Dilles et al., 2015; Lu et al., 2016). Ballard et al. (2002) defined the zircon Ce⁴⁺/Ce³⁺ ratio as an indicator of magmatic oxidation state. However, Dilles et al. (2015) and Lu et al. (2016) argue that these values are difficult to estimate accurately because the abundances of the adjacent elements, La and Pr, which are used as a baseline to calculate the magnitude of the Ce anomaly, are low and close to the analytical detection limit. Confirming this reservation, Ce⁴⁺/Ce³⁺ ratios and fO_2 values calculated for our samples using the equation of Trail et al. (2011) yielded unrealistic and widely variable Δ FMQ values (-9 to +5).

Apatite compositions

The SO₃, Cl, and F analyses of igneous apatites are listed in Digital Appendix Table A8 and illustrated in Figures 17 and 18. The results show that igneous apatites from the P2 porphyries have higher sulfur and chlorine concentrations than those from P1 and P3 rocks. The high SO₃ contents in apatites from the P2 porphyries (P2E: 0.11 to 0.8 wt. % SO_3 , average = 0.30 ± 0.13 wt. %, n = 34; P2L: 0.07 to 1.2 wt. % SO_3 , average = 0.29 ± 0.18 wt. %, n = 100) are similar to values from global porphyry Cu deposits (Streck and Dilles, 1998; Imai, 2002; Li et al., 2012; Pan et al., 2016; Richards et al., 2017). In comparison, apatite crystals from the P1 porphyry (RC13-39) have lower sulfur contents $(0.05 \text{ to } 0.19 \text{ wt. } \% \text{ SO}_3; \text{ average} = 0.11 \pm 0.03 \text{ wt. } \%, \text{ n} = 34), \text{ whereas those from the P3}$ porphyry (RC13-78) have the lowest sulfur concentrations (from 0.05 wt. % to below the analytical detection limit of SO₃; average of analyses above limit of detection = $0.03 \pm$ 0.01, n = 13). Although the ranges for P1 and P2L porphyries overlap at their outer limits, a t-test shows that the two populations are significantly different (t = 5.87, p = 0). Similarly, the Cl contents in apatite crystals from P1 and P3 porphyries are significantly lower (0.51 \pm 0.3 wt. %, n = 34, and 0.1 \pm 0.06 wt. %, n = 17, respectively) compared to values from the P2 porphyries (P2E: 1.18 ± 0.37 wt. %, n = 34; P2L: $1.47 \pm$ 0.28 wt. %, n = 100). There is a rough positive correlation between sulfur and chlorine contents of apatites from the four porphyries (Fig. 18A), with relatively constant molar S/Cl ratios (P1: 0.13 ± 0.08 , n = 34; P2E: 0.13 ± 0.08 , n = 34; P2L: 0.10 ± 0.14 , n =100; P3: 0.16 ± 0.09 , n =13; Table 4). Detailed analyses also show that single apatite crystals are commonly zoned in sulfur, with decreasing core-to-rim SO₃ contents (and small decreases in Cl; Fig. 17C), possibly reflecting progressive degassing of SO₂ (and Cl)

from the melt during crystallization (Richards et al., 2017).

Fluorine contents in apatites from the P1 and P3 porphyries are similar (P1: 3.64 \pm 0.48 wt. %, n = 34; P3: 3.6 ± 0.4 , n = 17) and distinctly higher than in the P2E (2.33 \pm 0.29 wt. %, n = 34) and P2L (2.32 \pm 0.23 wt. %, n = 100) porphyries. Some hydrothermal apatite grains were also analyzed for comparison with igneous grains, and they show wide variations in sulfur and chlorine content (Digital Appendix Table A8), as observed by other researchers (e.g., Li et al., 2012; Richards et al., 2017). The sulfur partition coefficient between apatite and melt is jointly controlled by temperature, oxygen fugacity, and the S content in the silicate melt (Peng et al., 1997; Parat and Holtz, 2005; Parat et al., 2011; Webster and Piccoli, 2015; Konecke et al., 2017a, b), and no formula currently exists to accurately calculate magmatic sulfur concentrations from apatite SO₃ contents. However, two semi-quantitative formulae can be used to estimate relative magmatic S content (Peng et al., 1997; Parat et al., 2011). Using the equation of Peng et al. (1997), the magmas related to the P2E and P2L porphyry intrusions are calculated to have sulfur concentrations of 0.03 ± 0.01 wt. % (n = 34) and 0.02 ± 0.01 wt. % (n = 100), significantly higher than the corresponding S contents for P1 (0.010 \pm 0.003 wt. % S, n = 34) and P3 magmas (0.003 \pm 0.001 wt. % S, n = 13; Table 4). The results calculated by the method of Parat et al. (2011) are more variable, and yielded lower absolute values than those from Peng et al. (1997). However, they also support the conclusion that the P2E and P2L magmas had significantly higher sulfur contents than the P1 and P3 magmas (Table 4). The chlorine partition coefficient between apatite and melt is a complex function of magma composition and Cl concentration in the melt (Zhu and Sverjensky, 1991; Piccolli

and Candela, 1994; Mathez and Webster, 2005; Webster et al., 2009; Chelle-Michou and

Chiaradia, 2017). Mathez and Webster (2005) proposed a value of 0.8 as the partition coefficient (mass ratio) between apatite and basaltic melt (51.1 wt. % SiO₂). Based on this semi-quantitative equation (and recognizing that these magmas were not basaltic), we estimate Cl concentrations in the P2E and P2L magmas of 1.48 ± 0.46 wt.% (n = 34) and 1.83 ± 0.34 wt.% (n = 100),respectively, significantly higher than for P1 (0.63 \pm 0.38 wt.%, n = 34) and P3 (0.12 \pm 0.07 wt.%; Table 4).

Discussion

Magmatic duration and timing of ore formation at Red Chris

Four phases of the Red Stock, P1, P2E, P2L, and P3, have been dated at 211.6 ± 1.3 Ma, 206.0 ± 1.2 Ma, 203.6 ± 1.8 Ma, and 201.7 ± 1.2 Ma, respectively, spanning a period of ~10 m.y. (Fig. 6; Digital Appendix Table A2). These ages are consistent with crosscutting relationships described by Rees et al. (2015). Re-Os dating of molybdenite intergrown with chalcopyrite in quartz veins yielded a weighted mean age of 206.1 ± 0.5 Ma (Fig. 7), reflecting the timing of mineralization and in good agreement with the emplacement age of the mineralized P2E quartz monzonite porphyry. Ore formation occurred at a relatively late stage in the ~10 m.y. evolution of the Red Stock, and over a relatively brief period time (<1 m.y., as defined by the uncertainty on the average of three Re-Os analyses). This relatively late and short duration of mineralization is similar to the timing observed in many other porphyry deposits (e.g., Cathles et al., 1997; Masterman et al., 2004; von Quadt et al., 2011; Chiaradia et al., 2013; Correa et al., 2016; Li et al., 2017).

The age of mineralization at ~206 Ma is contemporaneous with the major pulse of mineralization in the Stikine and Quesnel terranes, representing a particularly fertile preaccretionary magmatic event (Nelson and Colpron, 2007; Nelson et al., 2013; Logan and Mihalynuk, 2014).

Igneous geochemistry and petrogenesis

Whole-rock major and trace element compositions of P1 to P3 porphyry rocks are almost identical to each other, and show characteristics of subduction-related magmatic rocks with high LILE and low HFSE concentrations (Figs. 9 and 10; Hawkesworth et al., 1993; Pearce, 1996). The high Sr/Y ratios (53 \pm 23, n = 12) and slightly positive Eu anomalies (Eu_n/Eu_n* = 1.1 ± 0.15 ; n = 14) indicate the suppression of plagioclase crystallization and early crystallization of amphibole from hydrous and oxidized magmas (Richards and Kerrich, 2007; Loucks, 2014). Rees et al. (2015) concluded from their analysis of petrochemistry that the Red Stock evolved from subalkaline (P1, P2) to marginally alkaline (P3), and classified the overall system as high-K calc-alkalic. All the porphyritic rocks from P1 to P3 show mantle-like, relatively low (87Sr/86Sr)_t (0.7044 ± 0.0002) and high $\mathcal{E}_{Nd}(t)$ (+3.1 ± 0.4) values, and similar single-stage Nd model ages (0.80 \pm 0.05 Ga). Zircon crystals from these rocks yield positive zircon $\mathcal{E}_{Hf}(t)$ values (+8.1 to +14.8) and single-stage Hf model ages of 375 ± 52 Ma (Figs. 11 and 12). These data indicate minimal involvement of ancient crustal components in the petrogenesis of these magmas, as observed for other Mesozoic igneous rocks in the Stikine terrane, and are consistent with an oceanic island arc setting (Samson et al., 1989; Logan and Mihalynuk, 2014).

Late basaltic to andesitic dikes have similar Nd–Sr isotopic compositions to the porphyritic rocks, but show distinct trace element and REE patterns, and lower Sr/Y and La/Yb ratios, suggesting that they are not cogenetic. Oxygen fugacity and sulfur-chlorine-water contents of the porphyry magmas Magmatic oxidation state, and sulfur, chlorine, and water contents are fundamental factors for the formation of porphyry Cu deposits (e.g., Burnham, 1979; Candela, 1992; Richards, 2003; Loucks, 2014). It is difficult to measure the original magmatic oxidation state in altered plutonic rocks, but normalized Ce and Eu anomalies in zircons can be used to provide qualitative estimates (Ballard et al., 2002; Dilles et al., 2015; Lu et al., 2016). Zircon grains from the P1 to P3 porphyries display large positive Ce and small negative Eu anomalies (Eu_N/Eu_N* mostly > 0.4; Figs. 15 and 16), similar to zircons from other mineralized porphyry systems worldwide (Wang et al., 2014b; Dilles et al., 2015; Lu et al., 2016). These data are also consistent with estimates of magmatic oxidation state from early stage amphibole phenocrysts from the P2E and P2L porphyries, which yielded Δ FMQ values of 1.5 \pm 0.2 (n = 7) and 1.1 \pm 0.3 (n = 44). Consequently, we conclude that the four phases of magma in the Red Stock were all at least moderately oxidized. It is similarly difficult to estimate original magmatic water contents from altered plutonic rocks. However, the presence of amphibole phenocrysts in the P1 to P3 porphyries suggest that the magmas contained at least 4 wt. % H₂O (Burnham, 1979, Naney, 1983; Merzbacher and Eggler, 1984; Rutherford and Devine, 1988; Ridolfi et al.,

2010). This observation is supported by: (1) high whole-rock Sr/Y ratios (53 \pm 23, n =

12), which reflect the suppression of plagioclase relative to amphibole crystallization

under hydrous conditions (Moore and Carmichael, 1998; Richards and Kerrich, 2007); (2) high $10000 \times (\text{Eu}_{\text{N}}/\text{Eu}_{\text{N}}^*)/\text{Y}$ (>1) and low Dy/Yb (<0.3) ratios in zircons (Digital Appendix Table A7; Lu et al., 2016); (3) high whole-rock V/Sc ratios (9.0 \pm 0.9, n = 14) (Loucks, 2014); and (4) estimates of magmatic water content from plagioclase phenocryst compositions, which indicate values of ~5 wt.% H₂O. The compositions of igneous apatite inclusions in plagioclase and amphibole phenocrysts have been used to estimate magmatic sulfur and chlorine contents. The results suggest that apatite microphenocrysts from the P2E and P2L porphyries had higher S and Cl contents than for P1 and P3 (Fig. 18). The SO₃ contents in apatite are controlled by both magmatic sulfur concentration and oxidation state (Peng et al., 1997; Parat and Holtz, 2005; Webster and Piccoli, 2015). Sulfur occurs in the apatite structure mainly as S⁶⁺ and its content will increase in relatively oxidized environments (Boyce et al., 2010; Parat et al., 2011; Konecke et al., 2017b). The magmas associated with the four porphyry phases at Red Chris were all moderately oxidized, and therefore the differences in apatite sulfur content likely mainly reflect variations in the sulfur fugacity in the associated melts. The data indicate that the P2E and P2L porphyry magmas contained significantly higher S contents than the P1 and P3 magmas (Fig. 18). Models used to estimate magmatic sulfur content in equilibrium with apatite (Peng et al., 1997) are not strictly accurate because of uncertainties in the details of exchange reactions (Streck and Dilles, 1998; Mao et al., 2016) but are expected to be correct in relative terms. Our data yield higher values for the P2E and P2L porphyries (0.02–0.03 wt. % S) compared to

estimates for P1 (0.01 wt. % S) and P3 (0.003 wt. % S). Consequently, we conclude that

the syn-mineral porphyry (P2E and P2L) magmas were more S-rich than the premineralization P1 and especially the post-mineralization P3 porphyry magmas. Similarly, the magmatic Cl contents calculated from apatite compositions in the P2E and P2L porphyries are also higher than in the P1 and P3 porphyries, whereas F contents are lower (Table 4). Volatile species such as S, Cl, and F are differentially affected by degassing during ascent and crystallization of magmas: S and Cl will be preferentially lost to the vapor phase relative to F (Webster et al., 2014; Stock et al., 2016), leading to high F/Cl and F/S ratios in degassed magmas. The data shown in Figure 18 suggest that the P2E and P2L were S-Cl-rich, whereas the P1 and P3 were S-Cl-poor but enriched in F. These differences could reflect either loss of volatiles from the source magma chamber (presumed to be an underlying batholith) prior to emplacement of the P1 and P3 magmas but not in the case of P2, or conversely addition of a volatile phase or a volatilerich magma to the magma chamber immediately prior to emplacement of P2. Volatile loss is expected to be a general condition of upper crustal magma emplacement, and this may well be a factor contributing to the low Cl/F ratios in the late P3 magmas, as well as the negatively correlated trend for F and Cl in P1. The P2 magma was emplaced after P1, and so might be expected to be similarly degassed if this represents a comagnatic sequence. However, the 6–8 m.y. age difference between P1 (211.6 \pm 1.3 Ma) and P2 (P2E: 206.0 ± 1.2 Ma; P2L: 203.6 ± 1.8 Ma) indicates that P1 and P2 are not directly comagmatic. One explanation is that the P2 event represents a pulse of S-Cl-rich magma injected at a relatively late stage into the mid-crustal magma reservoir, leading to a pulse of fluid exsolution and injection of P2 magma into the shallower level Red Stock, where

syn-P2, quartz-vein hosted mineralization formed the Red Chris Cu-Au deposit. The

post-mineralization P3 (201.7 \pm 1.2 Ma) and P4 porphyries may represent the final stages of intrusive activity emanating from the now degassed underlying magma chamber. Although both P2E and P2L porphyries are S-Cl-rich, apatites from P2L are marginally the most enriched. Samples of P2L are also the most mafic (SiO₂-poor) in the suite, and it is therefore tempting to speculate that the source magma chamber was recharged with more mafic, S-Cl-H₂O-rich magma during the P2 stage. The solubility of S and Cl is higher in oxidized mafic magmas compared to felsic melts (> 1000 ppm S and > 3000 ppm Cl; Carmichael and Ghiorso, 1986; Webster, 1997; Webster et al., 1999; Hattori and Keith, 2001; Jugo et al., 2005; Jugo, 2009; Chelle-Michou and Chiaradia, 2017). Therefore, a key step in porphyry ore formation at Red Chris may have been recharge of the batholithic system by a less evolved, although broadly cogenetic, magma (cf. Hattori and Keith, 2001; Steinberger et al., 2013; Large et al., 2018). Support for this recharge model is provided by high magmatic temperatures calculated from zircon grains from some P2 samples (Fig. 16), and reverse zoning

calculated from zircon grains from some P2 samples (Fig. 16), and reverse zoning observed in plagioclase phenocrysts from the P2E and P2L porphyries, but which is not observed in P3 (Fig. 14C). In detail, zircons from the P2E porphyry show the highest Ti-in-zircon temperatures and Eu_N/Eu_N* values, and P3 the lowest values. This might reflect an influx of higher temperature melt during the evolution of the P2E magma, and greater degrees of plagioclase fractionation from the late P3 magma. The compositional ranges of up to ~19 mol % An for plagioclase are higher than the upper limit caused by chemical diffusion (~10 mol % An; Pearce and Kolisnik, 1990). Furthermore, these reverse zoning patterns are matched by FeO contents in the phenocrysts (Figs. 14A and 14B), which suggests that the zonation reflects changes in magma composition (as opposed to simply

changes in pressure, temperature, or water content; Ginibre and Wörner, 2007; Lange, 2009; Ustunisik et al., 2014; Waters and Lange, 2015). We therefore interpret these changes to reflect late-stage (rim) growth from a more mafic magma that mixed into a more evolved resident magma.

687 Conclusions

Zircon U-Pb ages indicate that the Red Stock was emplaced over a period of \sim 10 m.y. (211.6 \pm 1.3 Ma, 206.0 \pm 1.2 Ma, 203.6 \pm 1.8 Ma, and 201.7 \pm 1.2 Ma for the P1, P2E, P2L, and P3 porphyries, respectively). The ore-forming event at Red Chris occurred relatively late in this magmatic history, synchronous with P2E at 206.1 \pm 0.5 Ma, and over a short period of <1 m.y. (as constrained by the error on the average of three Re-Os analyses). The four phases of porphyry and their related magmas were chemically fairly homogeneous and isotopically primitive, consistent with an island arc origin. The magmas were all moderately oxidized and hydrous (\sim 5 wt. H₂O), but the P2 magmas had distinctly higher S and Cl contents. Combined with evidence for reverse zonation in plagioclase phenocrysts from P2 porphyries, and more mafic compositions in P2L, we suggest that the deeper source magma chamber was recharged at the time of emplacement of the P2 porphyries by injection of a more mafic S-Cl-rich magma. This recharge process may have triggered ore formation by causing voluminous exsolution of metalliferous hydrothermal fluids.

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Figure Captions

Fig. 1. Major terranes in the south of the Canadian Cordillera, showing Triassic to Jurassic magmatic belts and major associated porphyry deposits in the Stikine and Quesnel terranes (modified from Nelson and Colpron, 2007; Nelson et al., 2013; original graphics file provided by Joanne Nelson, British Columbia Geological Survey of Canada). The age of the Red Chris deposit is from this study (Table 1), and the ages for the other porphyry deposits are from Brown and Kahlert (1986; Red Mountain), Mortensen et al. (1995; Mt. Polley), Scott et al. (2008; Schaft Creek), Duuring et al. (2009; Kemess South), Taseko Mines Limited (2013; Gibraltar), Bath et al. (2014; Lorraine), Byrne and Tosdal (2014; Galore Creek), Devine et al. (2014; Lorraine), Logan and Mihalynuk (2014; Highland Valley, Copper Mountaine, Afton/Ajax, and Brenda), and Jago et al. (2014; Mt. Milligan). Fig. 2. Simplified geological map of the Red Stock and Red Chris Cu-Au deposit, showing the main mineralized zoned (named) and the locations of sampled drill holes (modified from Rees et al., 2015). Universal Transverse Mercator coordinates are based on the WGS84 datum. Fig. 3. Hand specimen photographs of samples of the P1, P2E, P2L, and P3 porphyry intrusions, and two late basaltic to andesitic dikes. (A) P1 porphyry with anhedral to subhedral altered amphibole and plagioclase phenocrysts, crosscut by pyrite-quartz veins (sample RC13-35). (B) P2E porphyry showing crowded texture with chloritized

amphibole and plagioclase phenocrysts; plagioclase grains are rimmed or replaced by

secondary K-feldspar. The brick-red color is due to fine-grained hematite (sample RC13-107). (C) P2L porphyry composed of abundant fresh amphibole and plagioclase phenocrysts, with quartz in the groundmass (sample RC13-32). (D) The P3 porphyry is similar to P2L, but is distinguished by the absence of quartz in the groundmass (sample RC13-78). (E) Andesitic dike with anhedral to subhedral amphibole phenocrysts, crosscut by a small calcite vein (sample RC13-62). (F) Basaltic dike with chloritized amphibole phenocrysts (sample RC13-106). See Digital Appendix Table A1 for sample locations. Fig. 4. West-southwest-east-northeast cross section A-A', and north-northwest-southsouthwest cross section B-B' (location of sections shown in Fig. 2), modified from Gillstrom et al. (2012) and Rees et al. (2015). Copper equivalent-grade zones, drill holes, and the boundary between potassic and post-potassic zones are shown: Cu equivalent (%) $= Cu (\%) + 0.486 \times Au (g/t).$ Fig. 5. Hand specimen photographs and photomicrographs of alteration and vein minerals. (A, B) P2E porphyry (samples RC13-81 and RC13-75) with potassic alteration and Atype quartz veins comprising magnetite, secondary K-feldspar, and disseminated bornite and pyrite. Potassic alteration is characterized by secondary K-feldspar veins and selvedges around A-veins. Late unmineralized carbonate veins cut the A veins. (C)

Amphibole phenocryst altered to secondary biotite, which has then been altered to

chlorite, reflecting potassic alteration overprinted by chlorite-sericite alteration (plane-

polarized transmitted light; P2E: sample RC13-30). (D) Plagioclase phenocryst partially

overprinted by sericite (cross-polarized transmitted light; P2E porphyry: sample RC13-

 total Fe₂O₃, and (F) MgO vs. SiO₂.

11). (E) B-type quartz veins with centerline pyrite in P2L porphyry (sample RC13-44). (F) Pyritic D vein in P1 porphyry (sample RC13-40). (G) High-grade ore in sheeted A-type quartz-chalcopyrite-K-feldspar veins (sample RC13-31 in P2E porphyry). (H) Quartzcarbonate-pyrite-molybdenite-chalcopyrite vein in P2E porphyry (reflected light; sample RC13-88). Abbreviations: Amp = amphibole; Bi = biotite; Bn = bornite; Cbn = carbonate; Chl = chlorite; Cpy = chalcopyrite; Kfs = K-feldspar; Mo = molybdenite; Mt = magnetite; Pl = plagioclase; Py = pyrite; Qtz = quartz. See Digital Appendix Table A1 for sample locations. Fig. 6. Zircon U–Pb Tera-Wasserburg concordia diagrams for (A) P1, (B) P2E, (C) P2L, and (D) P3 porphyry samples dated by LA-MC-ICP-MS. Uncertainty ellipses and calculated ages are shown at 2σ . Fig. 7. Weighted mean Re-Os model age of three molybdenite vein samples from the Red Chris Cu-Au deposit. Fig. 8. Zr/Ti vs. Nb/Y discrimination diagram (Winchester and Floyd, 1977) for porphyry and basaltic-andesitic dike samples from Red Chris. Fig. 9. Selected whole-rock major element variation diagrams for porphyry and basaltic andesitic dike samples from Red Chris: (A) K₂O₂, (B) Na₂O₃, (C) TiO₂, (D) Al₂O₃, (E)

Fig. 10. (A) Primitive mantle-normalized trace element, and (B) chondrite-normalized rare earth element diagrams for porphyry and basaltic-andesitic dike samples from Red Chris. The normalization values for primitive mantle chondrite are from from Sun and McDonough (1989). Fig. 11. $\mathcal{E}_{Nd}(t)$ vs. initial ${}^{87}Sr/{}^{86}Sr$ ratios for porphyry and basaltic–andesitic dike samples from Red Chris, calculated at t = 200 Ma. All the samples fall in the field of Mesozoic igneous rocks in the Stikine island arc terrane, clearly different from the Late Cretaceous to Eocene plutons in the Northern Coast Plutonic Complex, which were derived from evolved crust. The depleted MORB mantle field is from Pilet et al. (2011); the Stikinia Mesozoic igneous rock field is from Samson et al. (1989); the Northern Coast Plutonic Complex field is from Samson et al. (1991) and Patchett et al. (1998); all data are recalculated at 200 Ma. Fig. 12. Histogram and relative probability curve for zircon $\mathcal{E}_{Hf}(t)$ values from (A) P1, (B) P2E, (C) P2L, and (D) P3 porphyry samples. Fig. 13. Classification diagram and plots of oxidation state, temperature, pressure, and magmatic water content estimated from amphibole compositions from P2E and P2L porphyry samples at Red Chris. A. ${}^{C}(Al^{VI} + Fe^{3+} + 2Ti^{4+})$ (apfu) vs. ${}^{A}(Na^{+} + K^{+})$ (apfu). B. ΔFMQ vs. temperature. C. ΔFMQ vs. pressure. D. ΔFMQ vs. magmatic water content.

The classification diagram for calcic amphibole is given by the Excel spreadsheet of

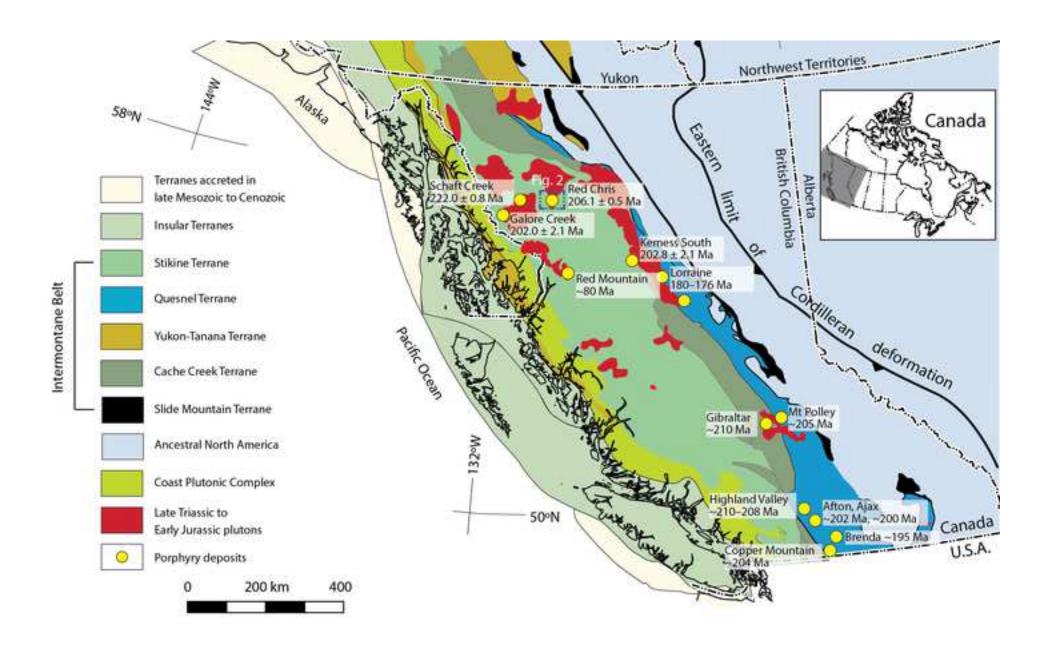
 Locock (2014), and the superscript C and A represent C and A cations following the general amphibole formula (AB₂C₅T₈O₂₂W₂), respectively. Note that potassic-magnesiohastingsite is included in the field of magnesio-hastingsite. ΔFMQ values, temperatures, and magmatic water content were calculated from the spreadsheet of Ridolfi et al. (2010). Amphibole crystallization pressures were calculated using the equation of Mutch et al. (2016). The Δ FMQ values were calculated following the equation of Myers and Eugster (1983): $\log fO_2 = -24,441.9$ /T (K) + 8.290 (± 0.167). Abbreviation: apfu = atoms per formula unit. Fig. 14. Photomicrographs (cross-polarized transmitted light), Backscattered eelectron (BSE) images, and electron microprobe analysis profiles for FeO and anorthite proportion (X_{An}) for representative plagioclase crystals from the P2E, P2L, and P3 porphyries at Red Chris. Red circles on photomicrographs and white circles on BSE images denote the analyzed spots. The error bars for X_{An} and FeO analyses are smaller than the size of the symbols. Abbreviations: Amp = amphibole; Ap = apatite; Pl = plagioclase; Ser = sericite. Fig. 15. Chondrite-normalized REE patterns for zircons from (A) P1, (B), P2E, (C) P2L, and (D) P3 porphyry samples from Red Chris. Normalization values are from Sun and McDonough (1989).

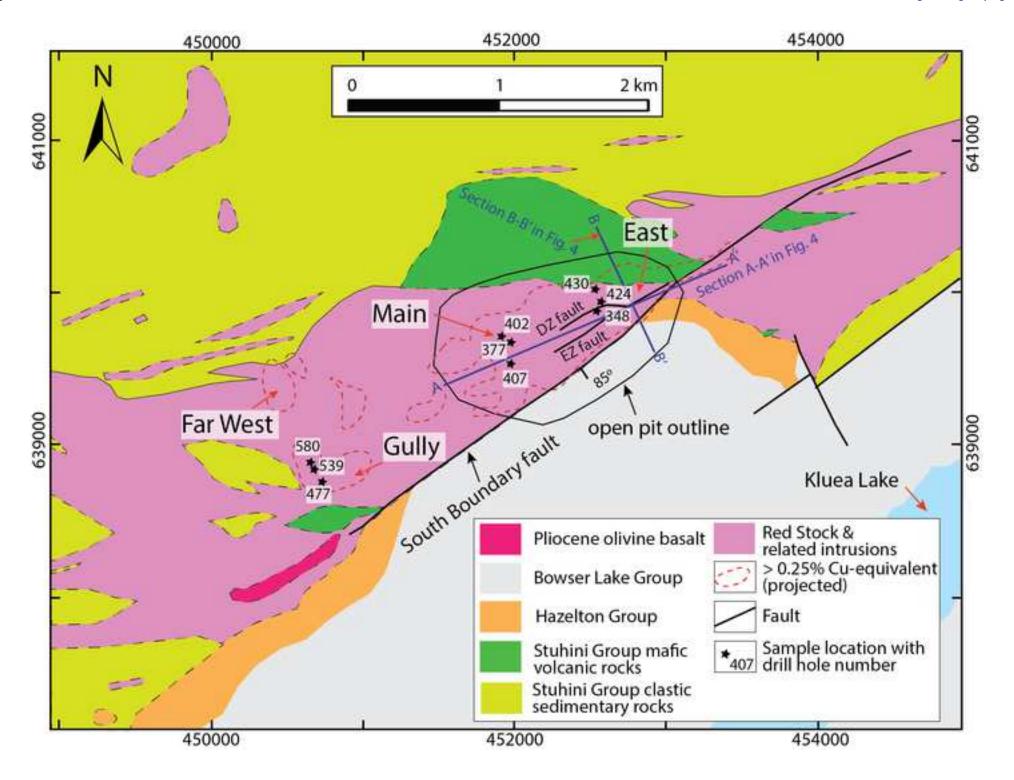
Fig. 16. Zircon Eu_N/Eu_N^* vs. temperature diagram. Eu_N/Eu_N^* is the europium anomaly, calculated as $Eu_N/Eu_N^* = Eu_N/(Sm_N \times Gd_N)^{0.5}$, using the chondrite normalization values

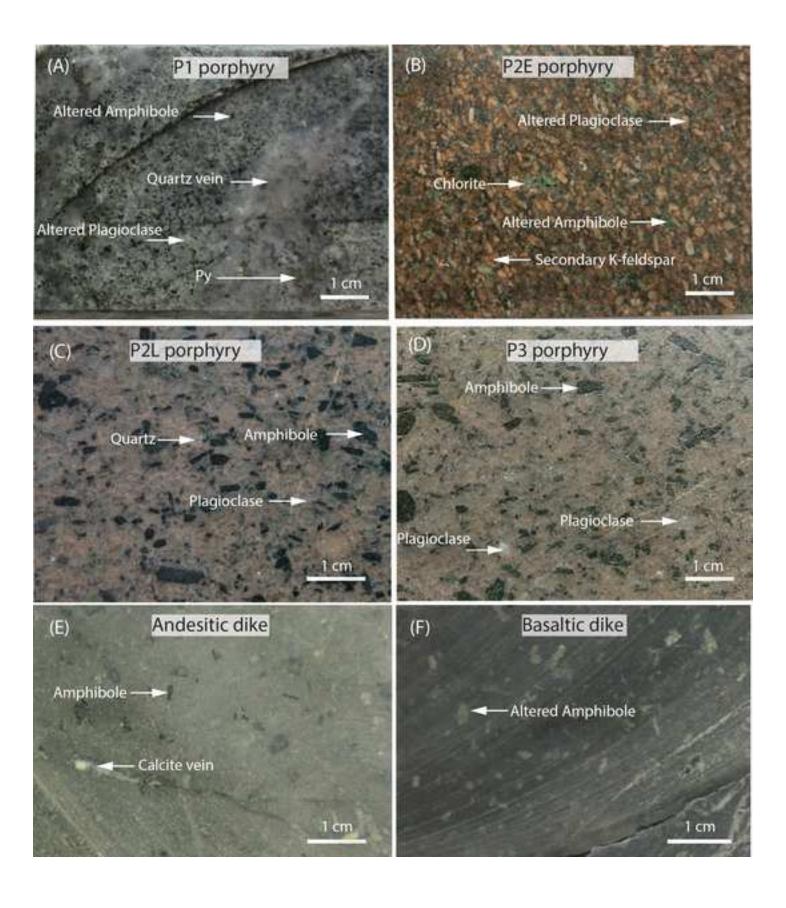
of Sun and McDonough (1989). Oxidized suites have zircon Eu_N/Eu_N* values >0.4 (Dilles et al., 2015).

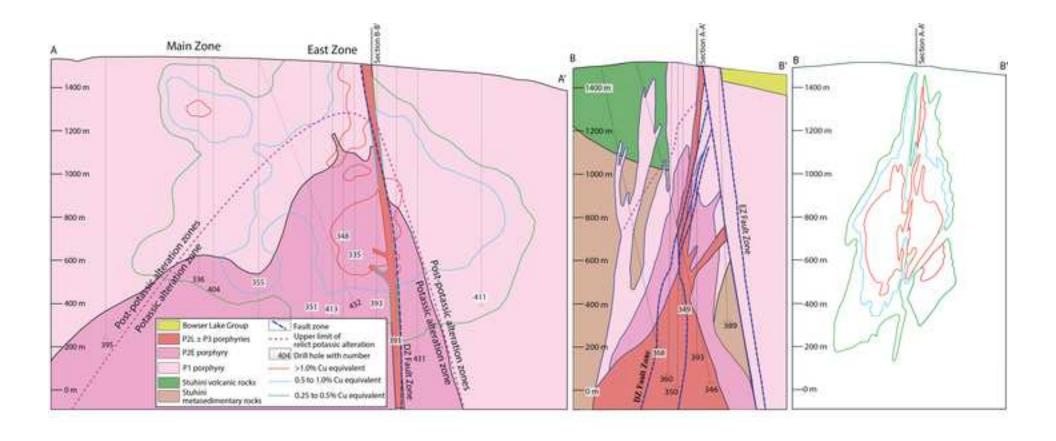
Fig. 17. Backscattered electron images of apatite crystals in samples from (A) P1 (RC13-39), (B) P2E (RC13-107), (C) P2L (RC13-33), and (D) P3 (RC13-78). Concentrations of SO₃ and Cl in apatite crystals are shown in wt. % (SO₃/Cl); red circles represent the analyzed spots. Higher concentrations are observed in apatites from P2E and P2L; some apatite microphenocrysts from P2 porphyries show zoning from SO₃-Cl-rich cores to SO₃-Cl-poorer rims (C).

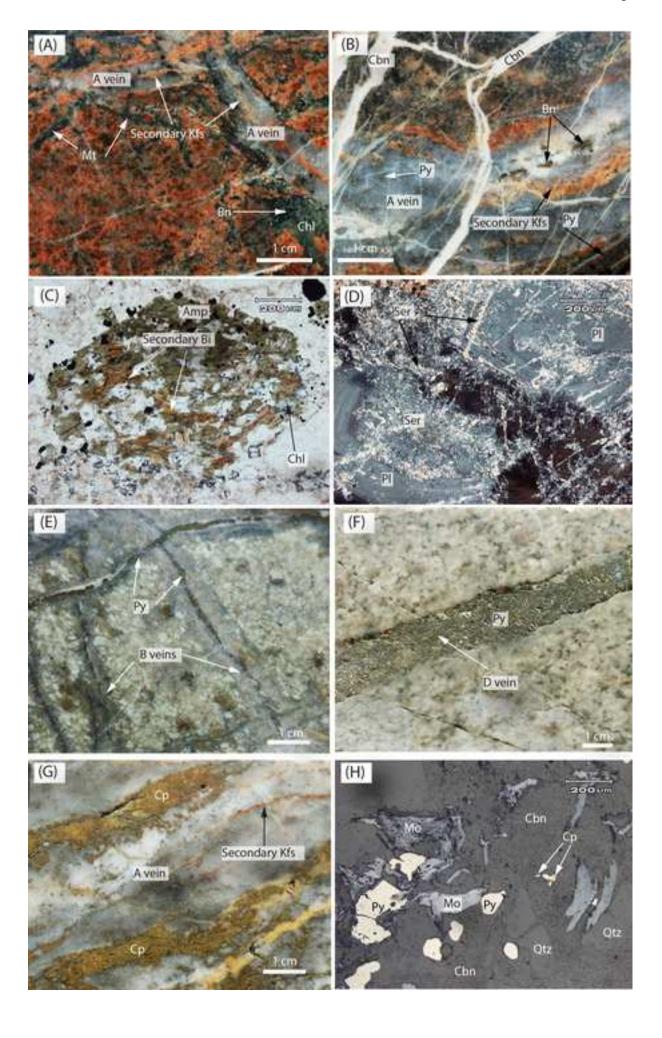
Fig. 18. Plots of (A) S, and (B) F vs. Cl contents for apatite microphenocrysts from P1 to P3 porphyry samples at Red Chris. Abbreviation: apfu = atoms per formula unit. Data from Digital Appendix Table A8.

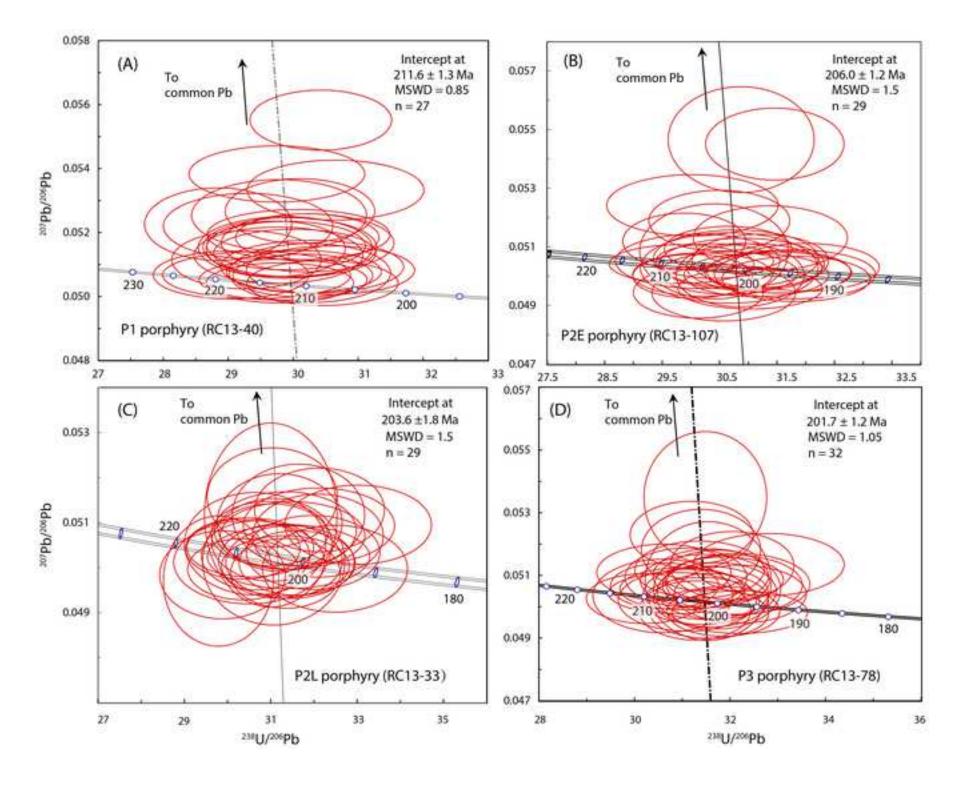


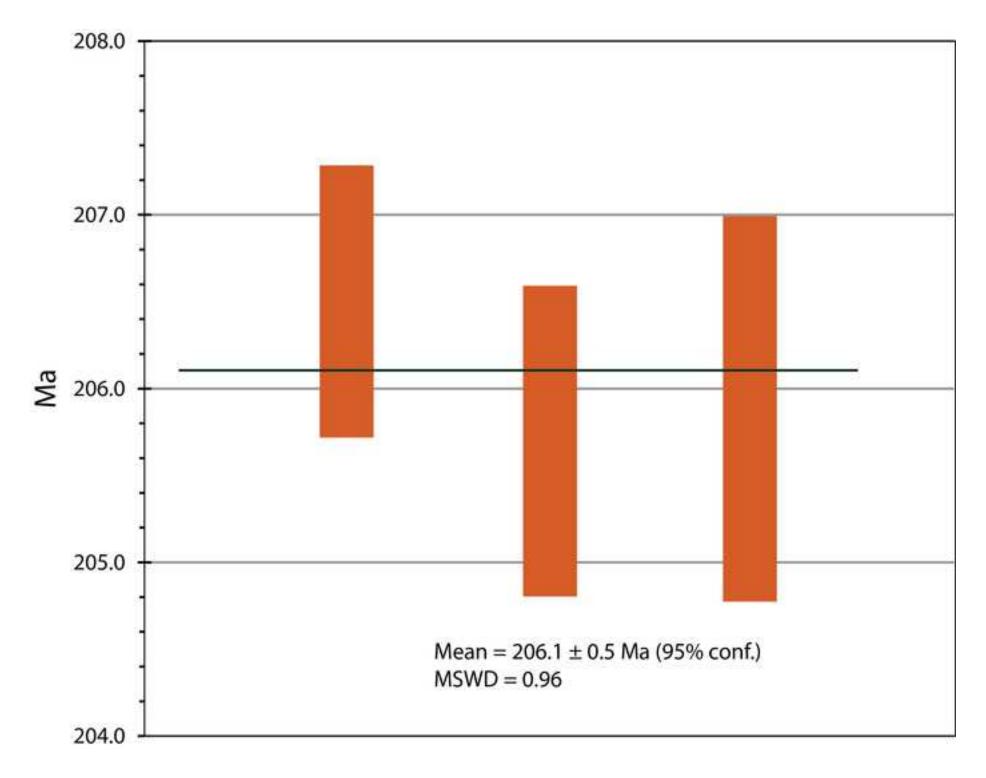


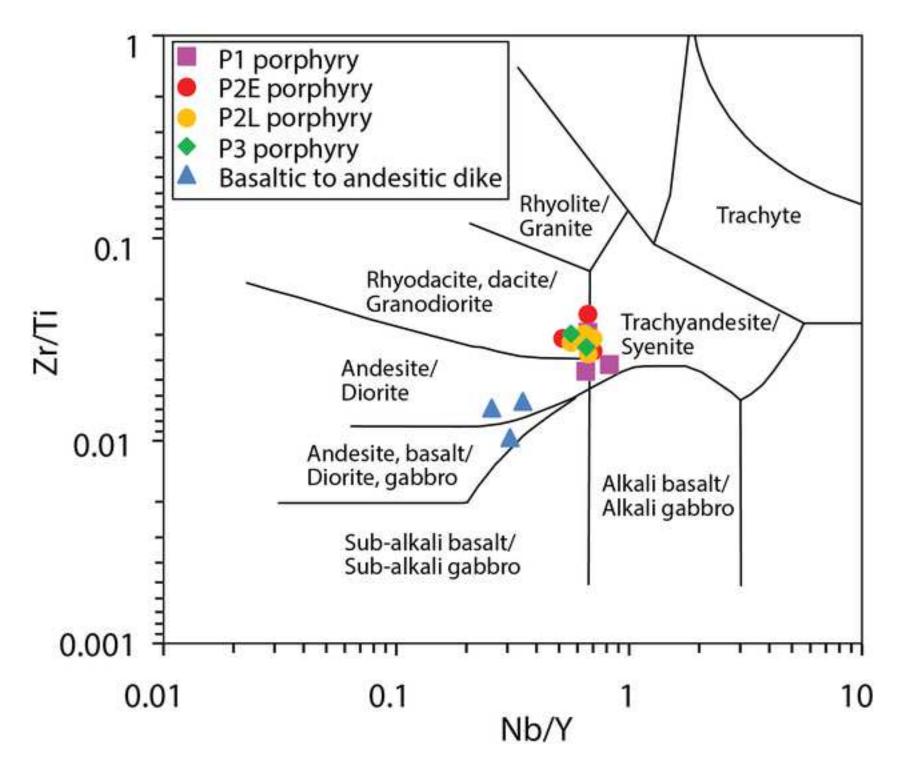


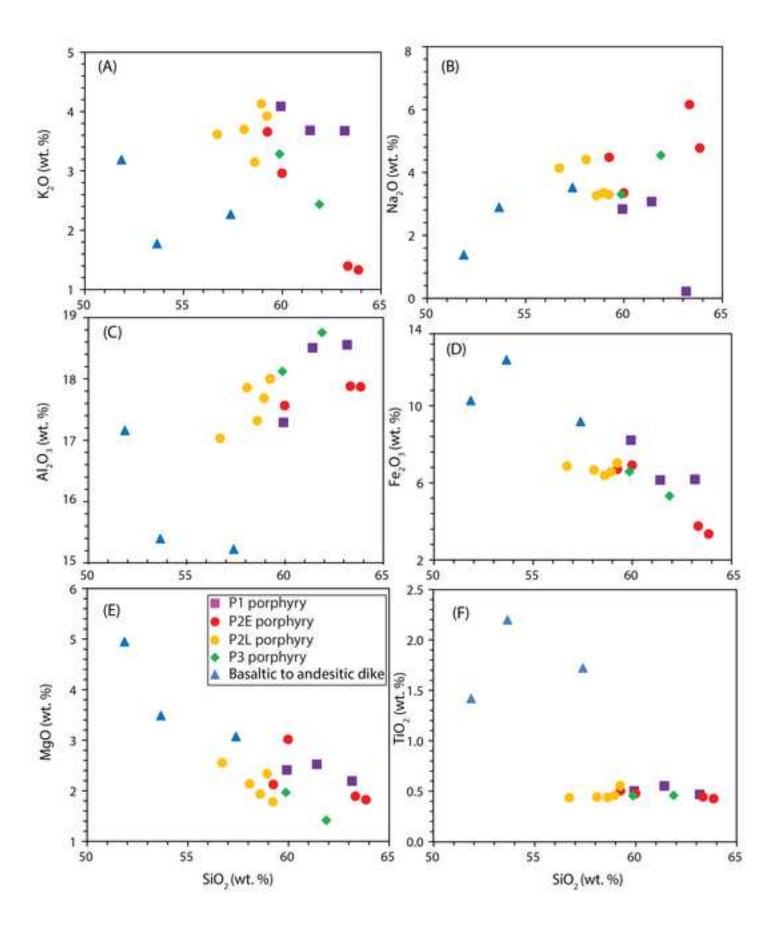


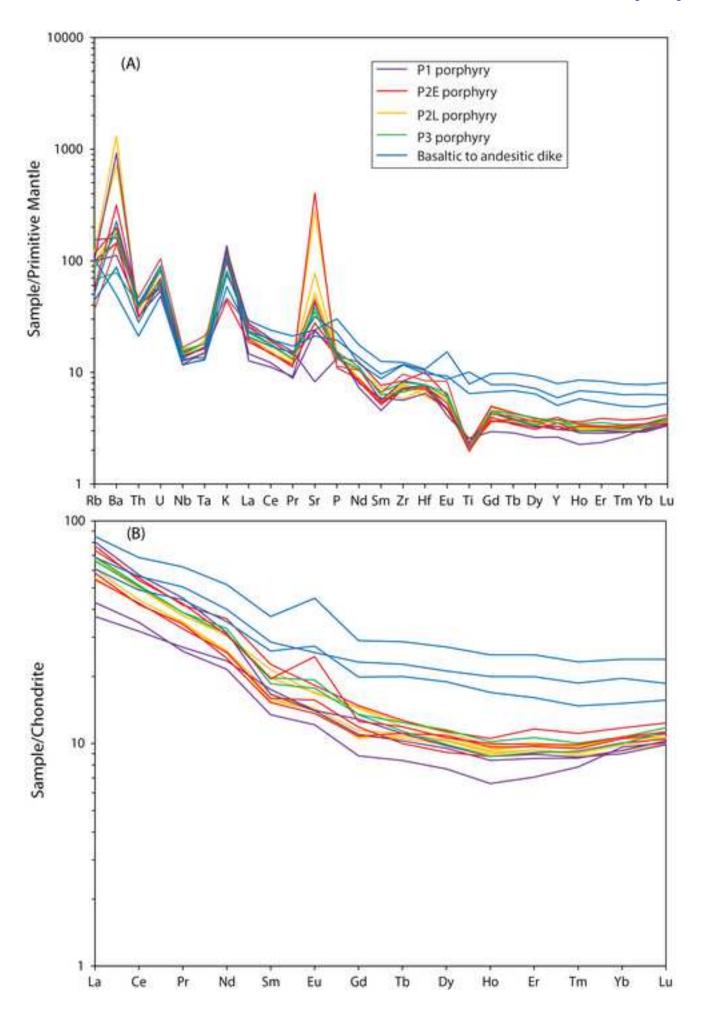


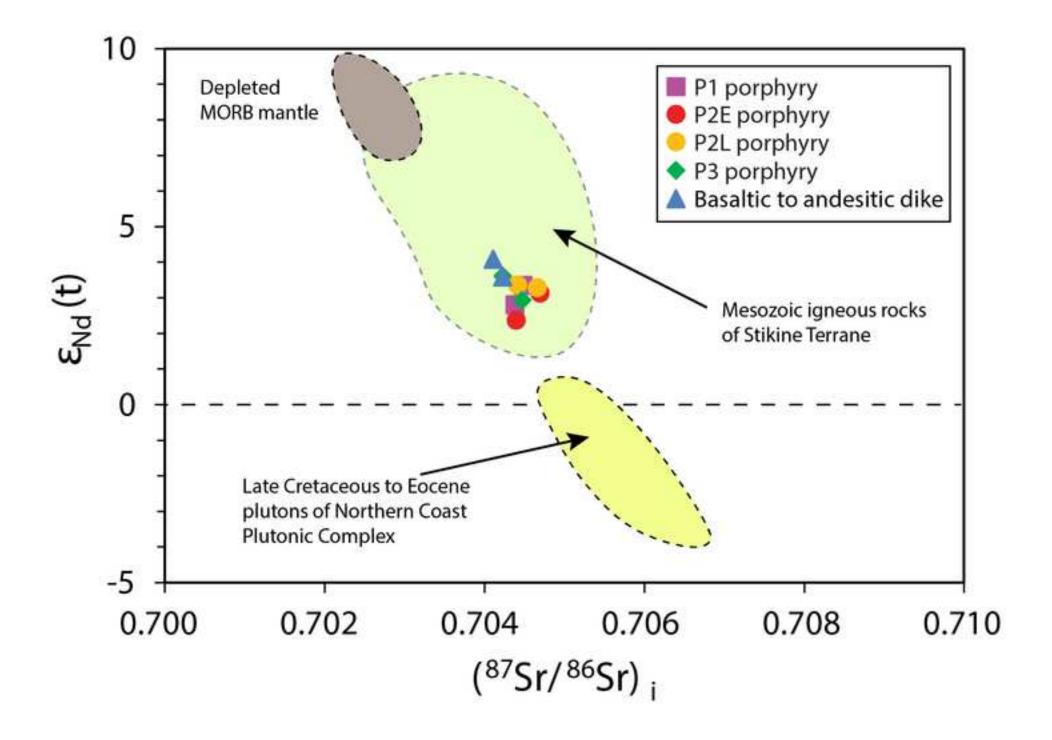


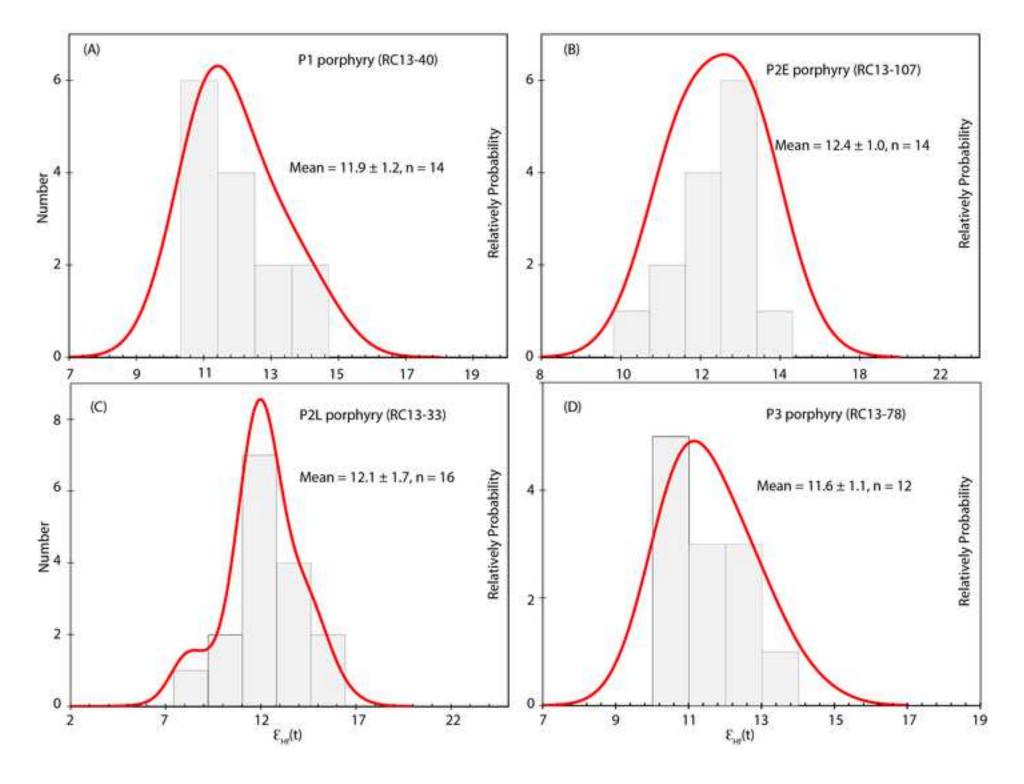


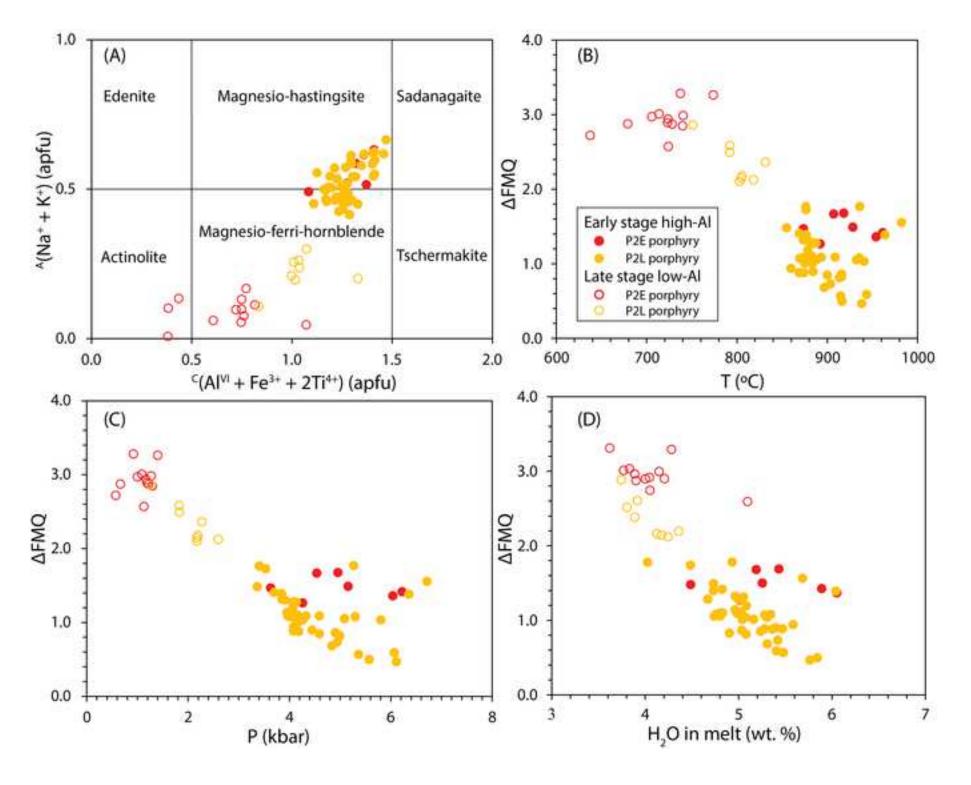


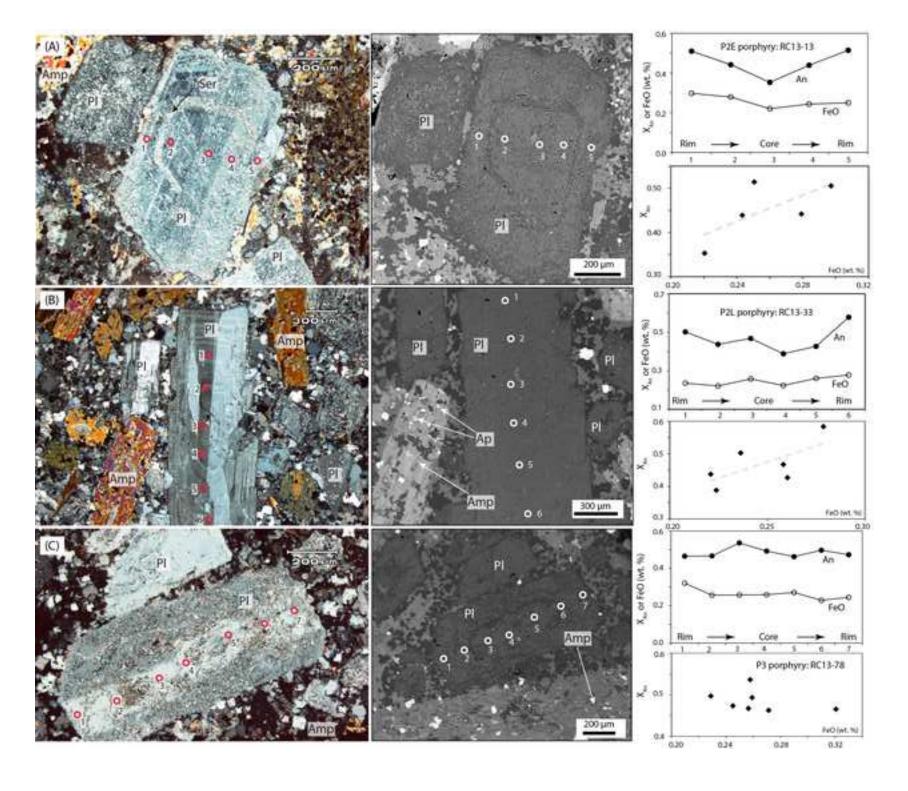


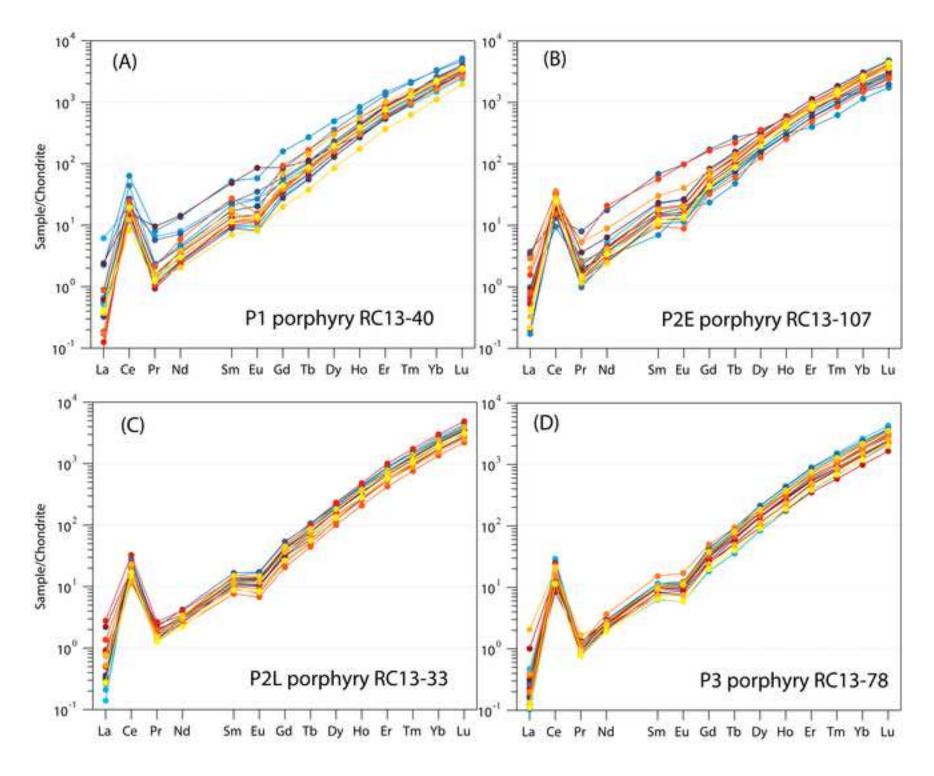


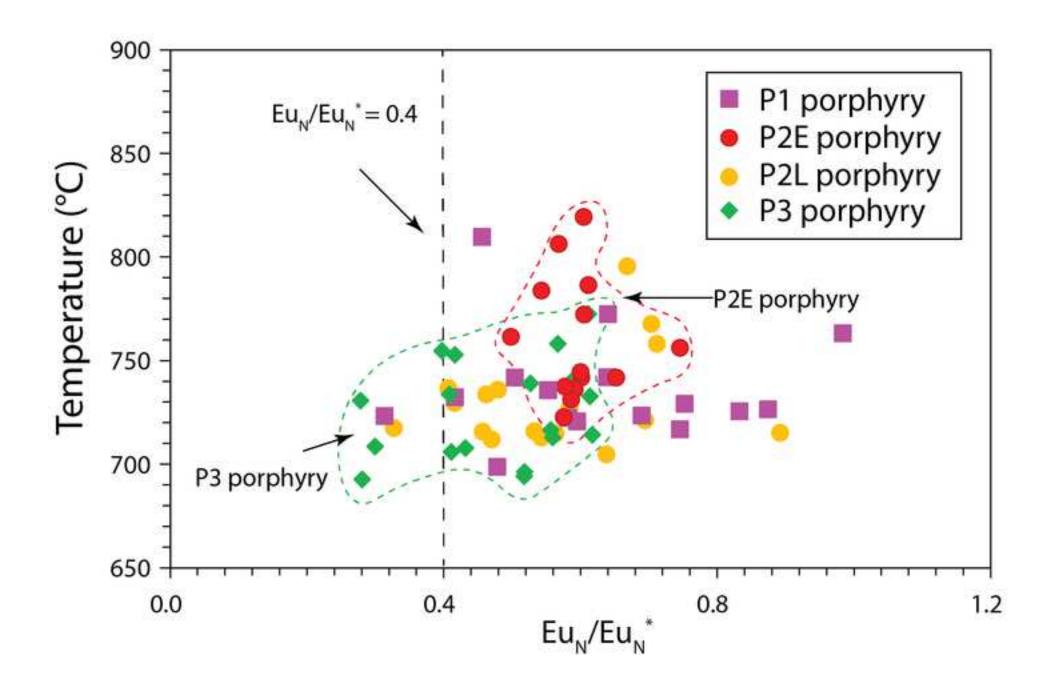


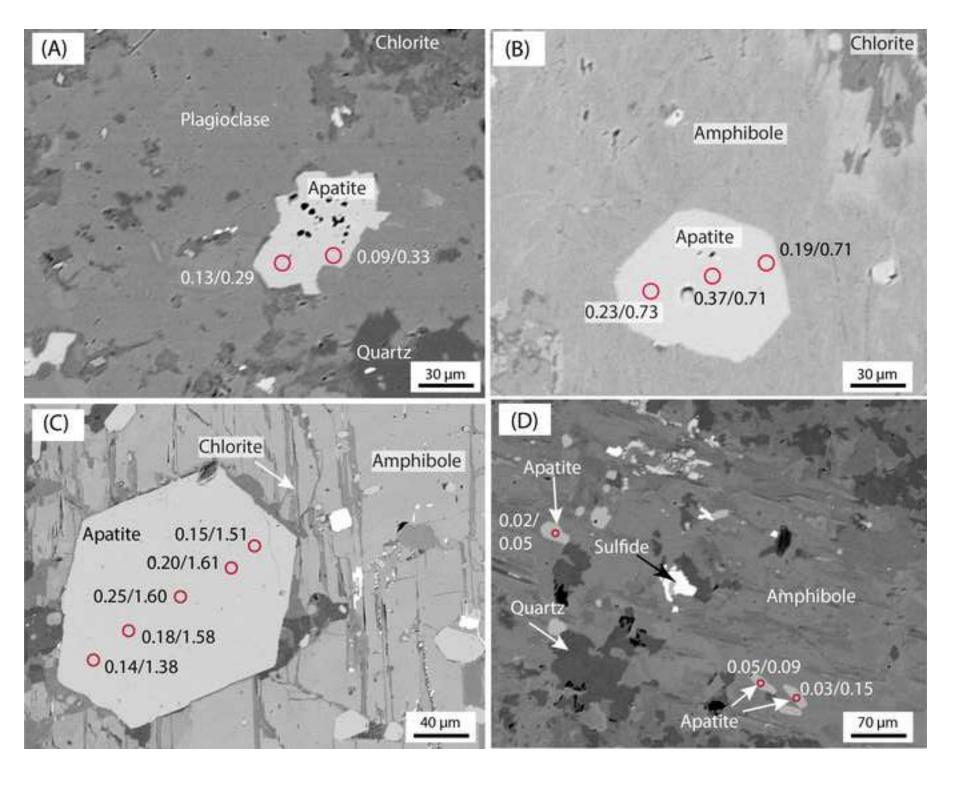












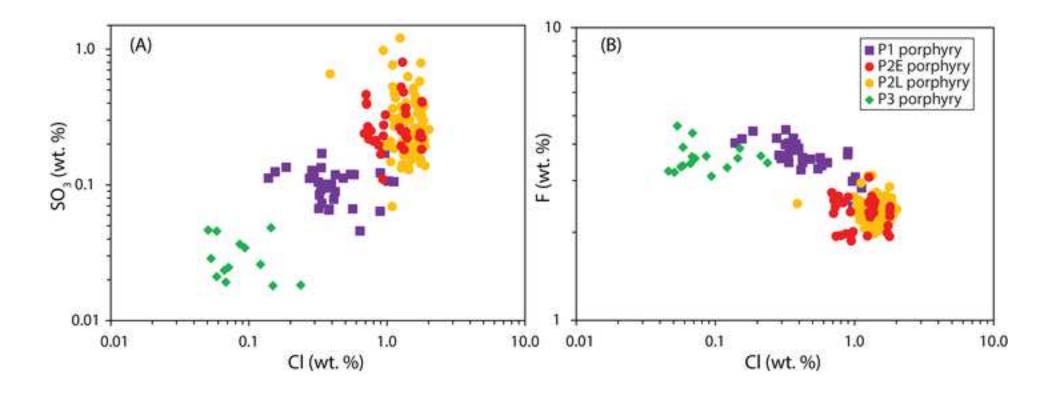


Table 1. Re-Os Isotope Analyses for molybdenite from the Red Chris Cu-Au deposit

Sample	Description	Ρο (μα/α)	± 2σ	¹⁸⁷ Re	± 2σ	¹⁸⁷ Os	± 2σ	Model Age
Sample	Description	Re (µg/g)	± 20	(μg/g)		(ng/g)	± 20	(Ma)
RC13-82	Quartz-carbonate-pyrite-	497.8	1.3	312.9	0.8	1078	0.0	206.5
RC13-88	molybdenite-chalcopyrite	1771	5.0	1113	3.0	3821	2.0	205.7
RC13-103	vein in P2E porphyry	1124	3.0	706	1.8	2427	4.0	205.9

Note: See the Digital Appendix Table A1 for sample locations.

± 2σ (Ma) 0.8 0.9 1.1 Table 2. Major and trace element compositions of porphyry and basaltic to andesitic dike re

Table 2							•			
	Detection									
Lithology	<u> </u>	P1	P1	P1	P2E	P2E	P2E	P2E	P2L	P2L
Weight %										
SiO_2	0.01	55.43	56.57	58.25	55.54	58.61	61.29	56.4	53.86	56.05
Al_2O_3	0.01	15.94	16.57	17.5	16.21	16.35	17.25	17.08	16.12	17.18
$Fe_2O_3(T)$	0.01	7.59	5.53	5.83	6.4	3.06	3.63	6.38	6.52	6.42
MnO	0.001	0.067	0.079	0.107	0.135	0.028	0.057	0.152	0.134	0.125
MgO	0.01	2.23	1.96	2.39	2.79	1.67	1.83	2.02	2.42	2.06
CaO	0.01	2.54	3.12	2.11	5.53	3.86	3.63	3.47	4.78	4.57
Na_2O	0.01	2.62	0.19	2.91	3.1	4.38	5.96	4.26	3.93	4.25
K_2O	0.01	3.78	3.29	3.49	2.74	1.22	1.35	3.48	3.43	3.57
TiO_2	0.001	0.463	0.419	0.523	0.441	0.39	0.429	0.48	0.414	0.426
P_2O_5	0.01	0.29	0.26	0.32	0.3	0.22	0.24	0.29	0.3	0.29
LOI		7.51	10.44	5.15	7.42	8.22	3.22	4.81	5.03	3.49
S	0.001	4.63	3.65	0.386	0.026	1.63	1.19	1.22		0.157
Total		98.45		98.57	100.6	98.02	98.9	98.81	96.94	98.43
Parts per	million									
Ba	1	6440	783	1123	1012	2201	980	1382	9182	5082
Co	0.1	10.5	9.4	16.2	12.7	10.6	8.3	9.3	11.4	12
Cr	0.5	24.9	30.2	26.3	115	50.1	19.4	26	31.2	32.8
Cs	0.1	3.7	2.1	4.2	6.7	4.5	0.8	1.1	4.7	4.9
Ga	1	17	14	17	15	15	15	17	15	17
Ge	0.5	3.2	2.4	1.4	2	3.1	2	1.8	1.7	1.9
Hf	0.1	2	2.4	2.4	2.2	2.3	2.6	3.1	2.2	2.2
Nb	0.2	9	9.6	10	9.9	8.4	11.9	10.8	10.1	11.5
Ni	1	5	7	32	23	3	7	3	5	6
Pb	_			_						
PU	5	14	8	< 5	< 5	< 5	< 5	< 5	< 5	< 5
Rb	1	64	63	98	63	35	23	73	74	76
Rb Sc	1 0.01	64 19.9	63 11.8	98 15	63 17.6	35 10.5	23 9.78	73 13	74 15.9	76 15.9
Rb Sc Sr	1 0.01 2	64 19.9 884	63 11.8 174	98 15 493	63 17.6 586	35 10.5 8516	23 9.78 944	73 13 508	74 15.9 5880	76 15.9 1639
Rb Sc Sr Ta	0.01 2 0.01	64 19.9 884 0.61	63 11.8 174 0.69	98 15 493 0.67	63 17.6 586 0.68	35 10.5 8516 0.62	23 9.78 944 0.87	73 13 508 0.74	74 15.9 5880 0.67	76 15.9 1639 0.73
Rb Sc Sr Ta Zr	1 0.01 2 0.01 1	64 19.9 884 0.61 63	63 11.8 174 0.69 94	98 15 493 0.67 80	63 17.6 586 0.68 79	35 10.5 8516 0.62 81	23 9.78 944 0.87 108	73 13 508 0.74 94	74 15.9 5880 0.67 85	76 15.9 1639 0.73 65
Rb Sc Sr Ta Zr Hf	1 0.01 2 0.01 1 0.1	64 19.9 884 0.61 63 2	63 11.8 174 0.69 94 2.4	98 15 493 0.67 80 2.4	63 17.6 586 0.68 79 2.2	35 10.5 8516 0.62 81 2.3	23 9.78 944 0.87 108 2.6	73 13 508 0.74 94 3.1	74 15.9 5880 0.67 85 2.2	76 15.9 1639 0.73 65 2.2
Rb Sc Sr Ta Zr Hf U	1 0.01 2 0.01 1 0.1 0.01	64 19.9 884 0.61 63 2 1.13	63 11.8 174 0.69 94 2.4 1.72	98 15 493 0.67 80 2.4 1.19	63 17.6 586 0.68 79 2.2 1.46	35 10.5 8516 0.62 81 2.3 1.26	23 9.78 944 0.87 108 2.6 2.2	73 13 508 0.74 94 3.1 1.75	74 15.9 5880 0.67 85 2.2 1.31	76 15.9 1639 0.73 65 2.2 1.69
Rb Sc Sr Ta Zr Hf U Th	1 0.01 2 0.01 1 0.1 0.01 0.05	64 19.9 884 0.61 63 2 1.13 2.71	63 11.8 174 0.69 94 2.4 1.72 3.23	98 15 493 0.67 80 2.4 1.19 3.38	63 17.6 586 0.68 79 2.2 1.46 2.64	35 10.5 8516 0.62 81 2.3 1.26 3.41	23 9.78 944 0.87 108 2.6 2.2 3.97	73 13 508 0.74 94 3.1 1.75 3.52	74 15.9 5880 0.67 85 2.2 1.31 2.63	76 15.9 1639 0.73 65 2.2 1.69 3.2
Rb Sc Sr Ta Zr Hf U Th	1 0.01 2 0.01 1 0.1 0.01 0.05 5	64 19.9 884 0.61 63 2 1.13 2.71	63 11.8 174 0.69 94 2.4 1.72 3.23 105	98 15 493 0.67 80 2.4 1.19 3.38 134	63 17.6 586 0.68 79 2.2 1.46 2.64 147	35 10.5 8516 0.62 81 2.3 1.26 3.41 109	23 9.78 944 0.87 108 2.6 2.2 3.97	73 13 508 0.74 94 3.1 1.75 3.52	74 15.9 5880 0.67 85 2.2 1.31 2.63 135	76 15.9 1639 0.73 65 2.2 1.69 3.2
Rb Sc Sr Ta Zr Hf U Th V	1 0.01 2 0.01 1 0.1 0.01 0.05 5	64 19.9 884 0.61 63 2 1.13 2.71 147	63 11.8 174 0.69 94 2.4 1.72 3.23 105	98 15 493 0.67 80 2.4 1.19 3.38 134	63 17.6 586 0.68 79 2.2 1.46 2.64 147	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16	23 9.78 944 0.87 108 2.6 2.2 3.97 110	73 13 508 0.74 94 3.1 1.75 3.52 114	74 15.9 5880 0.67 85 2.2 1.31 2.63 135	76 15.9 1639 0.73 65 2.2 1.69 3.2 135
Rb Sc Sr Ta Zr Hf U Th V Y	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724 63	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1 1 0.05	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1 0.05 0.05	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724 63 0.59 8.74 19.8	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr	1 0.01 2 0.01 1 0.01 0.05 5 1 1 0.05 0.05 0.05	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724 63 0.59 8.74 19.8 2.54	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1 1 0.05 0.05	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm	1 0.01 2 0.01 1 0.1 0.01 0.05 5 1 1 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd Tb	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2 0.38	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76 0.31	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16 0.41	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37 0.37	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51 0.44	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96 0.47	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88 0.44
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd Tb Dy	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 14 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57 0.41 2.44	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2 0.38 2.38	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76 0.31 1.92	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16 0.41 2.72	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37 0.37 2.28	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51 0.44 2.65	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96 0.47 2.83	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11 0.4 2.56	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88 0.44 2.67
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd Tb Dy Ho	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57 0.41 2.44 0.49	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2 0.38 2.38 0.47	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76 0.31 1.92 0.37	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16 0.41 2.72 0.54	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37 0.37 2.28 0.49	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51 0.44 2.65 0.56	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96 0.47 2.83 0.59	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11 0.4 2.56 0.51	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88 0.44 2.67 0.53
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd Tb Dy Ho Er	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57 0.41 2.44 0.49 1.43	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2 0.38 2.38 0.47 1.37	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76 0.31 1.92 0.37 1.13	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16 0.41 2.72 0.54 1.55	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37 0.37 2.28 0.49 1.45	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51 0.44 2.65 0.56 1.58	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96 0.47 2.83 0.59 1.86	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11 0.4 2.56 0.51 1.5	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88 0.44 2.67 0.53 1.55
Rb Sc Sr Ta Zr Hf U Th V Y Zn Zr Tl La Ce Pr Nd Sm Eu Gd Tb Dy Ho	1 0.01 2 0.01 1 0.1 0.05 5 1 1 0.05 0.05 0.05 0.	64 19.9 884 0.61 63 2 1.13 2.71 147 724 63 0.59 8.74 19.8 2.54 10.8 2.62 0.801 2.57 0.41 2.44 0.49	63 11.8 174 0.69 94 2.4 1.72 3.23 105 15 123 94 1.02 18.9 35.5 4.23 14.5 2.49 0.799 2.2 0.38 2.38 0.47 1.37 0.215	98 15 493 0.67 80 2.4 1.19 3.38 134 12 58 80 0.6 10.1 21.7 2.44 9.9 2.02 0.691 1.76 0.31 1.92 0.37	63 17.6 586 0.68 79 2.2 1.46 2.64 147 14 101 79 0.51 13.7 26.1 3.23 11.1 2.29 0.778 2.16 0.41 2.72 0.54	35 10.5 8516 0.62 81 2.3 1.26 3.41 109 16 29 81 0.17 12.8 26.5 3.08 11.8 2.39 0.893 2.37 0.37 2.28 0.49	23 9.78 944 0.87 108 2.6 2.2 3.97 110 18 32 108 0.12 18.1 33.5 3.99 14.1 2.93 1.4 2.51 0.44 2.65 0.56	73 13 508 0.74 94 3.1 1.75 3.52 114 17 56 94 0.12 17.3 34.4 3.93 16.7 3.4 1.04 2.96 0.47 2.83 0.59	74 15.9 5880 0.67 85 2.2 1.31 2.63 135 14 72 85 0.28 13.7 26 3.24 11.7 2.33 0.805 2.11 0.4 2.56 0.51	76 15.9 1639 0.73 65 2.2 1.69 3.2 135 16 79 65 0.06 15.6 30.9 3.52 14.4 2.94 0.968 2.88 0.44 2.67 0.53

Lu	0.002	0.247	0.257	0.252	0.276	0.261	0.282	0.309	0.25	0.266
La/Yb		6.069	12.685	6.558	8.155	7.950	10.585	9.202	9.257	9.070
Sr/Y		63.143	11.600	41.083	41.857	532.250	52.444	29.882	420.000	102.438
Nb/Y		0.643	0.640	0.833	0.707	0.525	0.661	0.635	0.721	0.719
Zr/Ti		0.021	0.034	0.024	0.028	0.032	0.041	0.031	0.033	0.025
V/Sc		7.387	8.898	8.933	8.352	10.381	11.247	8.769	8.491	8.491
Eu_N/Eu_N^*		0.938	1.037	1.114	1.063	1.140	1.569	0.996	1.103	1.011

Note: See the Digital Appendix Table A1 for sample locations.

ocks at Red Chris

RC13-32			RC13 77	RC13.78	RC13 62	RC13 70	RC13-97
P2L	P2L	P2L	P3	P3			
r ZL	r 4L	r 4L	F 3	13	DIKE	DIKE	DIKE
57.57	56.05	57.38	58.47	57.45	46.23	49.38	54.17
17.22	16.51	17.38	17.67	17.34	15.25	14.12	14.32
6.41	6.09	6.8	5.02	6.31	9.15	11.39	8.66
0.13	0.107	0.116	0.066	0.11	0.108	0.168	0.19
2.28	1.85	1.72	1.34		4.41	3.21	2.9
5.66	6.6	6.04	4.84	6.1	7.78	7.3	5.72
3.27	3.12	3.19	4.29	3.17	1.23	2.66	3.32
4.03	3.01	3.8	2.3	3.15	2.84	1.63	2.14
0.444	0.418	0.536	0.432	0.436	1.25	2.008	1.608
0.28	0.3	0.3	0.28	0.32	0.44	0.39	0.62
2.33	4.36		5.51	4.03	10.85	7.97	5.61
0.296	0.053	0.155	0.633	0.112	0.231	0.255	0.092
99.63	98.4	100.4	100.2		99.53	100.2	99.28
77.05	70.1	100.1	100.2	100.5	77.55	100.2	<i>>></i> .20
1322	1323	1002	545	1232	343	614	1572
11.2	8.6		8		27.1	22.6	10.1
22.6	46.8	36	35.1	19.3	15.8	21.3	16.4
1.7	2.1	1.6	2.9	1.9	2.4	0.7	0.7
17	15	15	16	18	16	16	16
1.6	1.8	1.4	1.7	1.7	2.4	1.7	2
2.3	1.9	2.1	2.4	2.4	3	3.2	3.3
10.3	9.9	9.2	10.4	11.1	8.3	8.4	9.4
5	5	6	6		4	3	5
	< 5			< 5			< 5
62	61	57	43		66	28	32
14.2	14.5	15.8	12.8	15.5	22.1	32	26.2
978	1083	985	737		668	447	507
0.69	0.79		0.75	0.74	0.54		0.56
74	86	88	91	75	131	132	138
2.3	1.9	2.1	2.4		3		3.3
1.46		1.27	1.74			1.34	
3.48	3.01	2.7	3.45		1.8	2.39	3.46
129	130 14	148	122 18	132	221 23	480 27	129
17 66	52	15 60	36	16 50	23 84	99	36 111
74	86	88	91	75	131	132	138
0.06	0.24	0.2	0.14	0.08	0.61	0.14	0.17
16	14.3	13	16.2	15.5	14.3	16.1	20
31.4	27.4	26.1	31.9		30.3	34.9	42.4
3.54	3.3	3.26	3.64		4.16	4.75	5.84
14	11.7	12	14.3	15.1	16.2	18.4	23.8
3.22	2.4	2.51	2.96		3.9	4.28	5.57
0.958	0.798	0.818	1.1	1.01	1.56	1.46	2.56
2.91	2.14	2.28	2.69	2.7	3.98	4.64	5.8
0.46	0.39	0.41	0.42	0.46	0.74	0.84	1.06
2.8	2.47	2.59	2.48	2.88	4.73	5.29	6.78
0.55	0.5	0.52	0.49	0.57	0.95	1.12	1.4
1.61	1.5	1.56	1.46		2.57	3.19	4
0.243	0.226	0.241	0.23		0.369	0.467	0.581
1.71	1.57	1.68	1.6	1.71	2.42	3.14	3.82

0.282	0.265	0.264	0.281	0.294	0.391	0.465	0.597
9.357	9.108	7.738	10.125	9.064	5.909	5.127	5.236
57.529	77.357	65.667	40.944	50.000	29.043	16.556	14.083
0.606	0.707	0.613	0.578	0.694	0.361	0.311	0.261
0.027	0.033	0.027	0.033	0.028	0.016	0.010	0.014
9.085	8.966	9.367	9.531	8.516	10.000	15.000	4.924
0.951	1.070	1.039	1.185	1.120	1.203	0.996	1.369

Table 3. Whole-rock Nd-Sr isotopic results for the porphyry and basaltic to andesitic dike rocks

Sample	Rb (μg/g)	Sr (µg/g)	87 Rb/ 86 Sr	⁸⁷ Sr/ ⁸⁶ Sr	± 2σ	T (Ma)	$(^{87}\text{Sr}/^{86}\text{Sr})_{\text{T}}$	Sm (µg/g)	Nd (µg/g)
P1 porphyry									
RC13-39	44.10	640.50	0.199213	0.705022	0.000020	211.6	0.7044	2.29	9.96
RC13-40	36.38	96.01	1.096378	0.707480	0.000032	211.6	0.7042	2.94	15.77
P2E porphyry									
RC13-92	18.75	703.50	0.077095	0.704633	0.000019	206.0	0.7044	3.06	15.31
RC13-107	43.23	320.70	0.390074	0.705837	0.000037	206.0	0.7047	2.97	14.70
P2L porphyry									
RC13-32	27.07	509.50	0.153691	0.705103	0.000035	203.6	0.7047	2.65	12.77
RC13-33	47.40	846.00	0.162098	0.704873	0.000017	203.6	0.7044	2.63	12.87
P3 porphyry									
RC13-77	37.47	531.70	0.203855	0.704991	0.000016	201.7	0.7044	2.80	14.06
RC13-78	25.78	427.70	0.174403	0.704748	0.000024	201.7	0.7042	2.67	13.03
Mafic to andesitic dike									
RC13-97	23.90	385.10	0.179519	0.704792	0.000020	200.0	0.7043	6.22	25.72
RC13-79	15.73	252.40	0.180303	0.704741	0.000030	200.0	0.7042	4.81	20.18

Notes: ${}^{1}N/A = no$ data; the used ages for calculations are from zircon U-Pb dating results for porphyry rocks, and assuming $T_{DM1} = (1/\lambda) \times Ln[((^{143}Nd/^{144}Nd)sample - (^{143}Nd/^{144}Nd)DM)/((^{147}Sm/^{144}Nd)sample - (^{147}Sm/^{144}Nd)DM + 1)]$. See Appendiant depleted mantle. See the Digital Appendix Table A1 for sample locations.

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¹⁴⁷ Sm/ ¹⁴⁴ Nd	$(^{143}\text{Nd}/^{144}\text{Nd})_0$	± 2σ	$\varepsilon_{Nd}(t=0)$	$(^{143}\text{Nd}/^{144}\text{Nd})_{\text{T}}$	T _{DM1} (Ga)	$\varepsilon_{Nd}(T)$
0.139132	0.512732	0.000009	1.8	0.512540	0.88	3.4
0.112591	0.512672	0.000010	0.7	0.512516	0.74	2.9
0.120965	0.512659	0.000007	0.4	0.512496	0.83	2.4
0.122321	0.512700	0.000010	1.2	0.512535	0.77	3.2
0.125462	0.512713	0.000008	1.5	0.512546	0.78	3.3
0.123603	0.512713	0.000010	1.5	0.512548	0.76	3.4
0.120209	0.512686	0.000012	0.9	0.512527	0.78	2.9
0.123767	0.512727	0.000010	1.7	0.512564	0.74	3.6
0.146221	0.512751	0.000006	2.2	0.512559	N/A	3.5
0.144047	0.512766	0.000011	2.5	0.512578	N/A	3.9

200~Ma for basaltic to andesitic dikes based on geological relationship, see text for details. ix 1 for the $^{143}\rm{Nd}/^{144}\rm{Nd}$ and $^{147}\rm{Sm}/^{144}\rm{Nd}$ values of present day chondrite

Table 4. Estimates of magmatic temperature and sulfur-chlorine content from igneous apatite and whole-rock compositions

Porphyry Phase	Sample	Whole rock SiO ₂ (wt. %) ¹	Whole rock P ₂ O ₅ (wt. %) ¹	AST ² (°C)	Apatite SO ₃ (wt. %) (n) ³ [maximum, minimum]	Apatite Cl (wt. %) (n) ³	Apatite F (wt. %) (n) ³	Apatite Molar S/Cl (n) ³	Apatite Molar Cl/F (n) ³	Average magmatic S content (wt. %) ⁴ [maximum]	Average magmatic S content (wt. %) ⁵ [maximum]	Average magmatic Cl content (wt. %) ⁶
P1	RC13-39	59.93	0.31	921	$0.11 \pm 0.03 (34)$ [0.05, 0.19]	$0.51 \pm 0.3 (34)$	$3.64 \pm 0.48 (34)$	$0.13 \pm 0.08 (34)$	$0.08 \pm 0.06 (34)$	0.010 ± 0.003 [0.017]	0.002 ± 0.0004 [0.003]	0.63 ± 0.38
DOE	RC13-13 ⁷	63.33	0.25	932	$0.28 \pm 0.10 (19)$ [0.11, 0.46]	1.47 ± 0.22 (19)	2.32 ± 0.29 (19)	0.09 ± 0.03 (19)	$0.35 \pm 0.10 (19)$	0.030 ± 0.010 [0.050]	0.006 ± 0.004 [0.015]	1.84 ± 0.27
P2E	RC13-107	59.25	0.30	909	$0.32 \pm 0.17 (15)$ [0.18, 0.80]	$0.82 \pm 0.10 (15)$	2.36 ± 0.30 (15)	$0.17 \pm 0.09 (15)$	$0.19 \pm 0.05 (15)$	0.031 ± 0.017 [0.078]	0.014 ± 0.031 [0.012]	1.02 ± 1.13
Day	RC13-26	58.08	0.30	892	0.30 ± 0.22 (48) [0.07, 1.2]	1.57 ± 0.33 (48)	2.23 ± 0.22 (48)	0.11 ± 0.20 (48)	0.39 ± 0.10 (48)	0.017 ± 0.013 [0.070]	0.0490 ± 0.2380 [1.6168]	1.96 ± 0.41
P2L	RC13-33	58.61	0.31	905	$0.28 \pm 0.13 (52)$ [0.14, 0.79]	1.37 ± 0.17 (52)	2.40 ± 0.21 (52)	0.09 ± 0.04 (52)	$0.31 \pm 0.05 (52)$	0.020 ± 0.009 [0.056]	0.008 ± 0.017 [0.117]	1.71 ± 0.21
P3	RC13-78	59.86	0.33	928	$0.03 \pm 0.01(13)$ [0.02, 0.05]	0.02 ± 0.02 (17)	3.60 ± 0.40 (17)	0.16 ± 0.09 (13)	$0.01 \pm 0.01 (17)$	0.003 ± 0.001 [0.005]	0.001 ± 0.0001 [0.001]	0.12 ± 0.07

Notes:

- 1. Normalized to 100 wt. % (Digital Appendix Table A3).
- 2. Apatite saturation temperature (AST) calculated from whole-rock SiO₂ and P₂O₅ concentrations using the equation of Piccoli and Candela (1994).
- 3. Average of all igneous apatite analyses (Digital Appendix Table A8).
- 4. Estimated from apatite SO₃ contents (Digital Appendix Table A8) using the temperature-dependent apatite—melt partition coefficient formula of Peng et al. (1997): $lnK_D = 21130/T 16.2$ (where T is in Kelvin).
- 5. Estimated from apatite SO₃ contents (Digital Appendix Table A8) using the temperature-dependent apatite—melt partition coefficient formula of Parat et al. (2011): SO₃ apatite (wt. %) = $0.157 \times \ln SO_3$ glass (melt, wt.%) + 0.9834 (r² = 0.62).
- 6. Estimated from apatite Cl contents (Digital Appendix Table A8) using the apatite—melt partition coefficient value (mass ratios) of Mathez and Webster (2005), which is 0.8 for basaltic melt (51.1 wt. % SiO₂) and tends to be similar for rhyodacitic melt at 200 MPa (Webster et al., 2009).
- 7. SiO₂ and P₂O₅ compositions for this sample taken from sample RC13-92 with same lithology as RC13-13 but less altered (Digital Appendix Table A3).

Digital Appendix Tables A1-A9

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