#### **ARTICLE**



# Chalcophile element fertility and the formation of porphyry $\text{Cu} \pm \text{Au}$ deposits

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#### Abstract

Chalcophile element fertility, the chalcophile metal abundance in the source magma, is likely to be a critical factor for the formation of porphyry  $Cu \pm Au$  deposits. In this study, we provide evidence to support this hypothesis by comparing the platinum group element (PGE) geochemistry of barren and ore-bearing  $Cu \pm Au$  granitic suites. We report the PGE contents of three barren volcanic and subvolcanic suites from Argentina and Japan and two  $Cu \pm Au$  bearing suites from Indonesia and Chile. These results are compared with those from previous studies of a porphyry Cu-only subvolcanic suite from Chile and three porphyry Cu-Au-bearing suites from Australia and the USA. The barren suites are depleted in PGE abundances by the time of fluid exsolution (<0.1 ppb Pd and Pd/Pt <~3), which is attributed to early sulfide saturation in a mid to lower crustal magma chamber. In contrast, the  $Cu \pm Au$  ore-bearing suites contain at least an order of magnitude higher PGE contents than the barren ones at fluid saturation (up to ~10 ppb Pd and Pd/Pt of 0.1–40). They are characterized by late sulfide saturation, which allows both chalcophile elements and sulfur to concentrate by fractional crystallization before volatile saturation. We suggest that plots of Pd/MgO against Pd/Pt for igneous suites can be used to estimate chalcophile element fertility and distinguish between barren, porphyry Cu, and porphyry Cu-Au granitoid systems. The positive correlation of these chalcophile element fertility indicators and ore grades suggests that metal contents in magmas play an important role in controlling ore grade, particularly Au, in porphyry  $Cu \pm Au$  deposits.

Keywords Platinum group element · Porphyry deposit · Chalcophile element fertility · Sulfide saturation

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## Introduction

Porphyry deposits are the primary source of the world's Cu and Au. The metals are interpreted to have been deposited from magmatic fluids exsolved from the ore-associated porphyries (Burnham 1967). They predominantly occur above subduction zones, which has led to the suggestion that oxidized, water-rich subduction zone magmas may play an essential role in ore formation. However, only a limited number of porphyry suites host economic deposits. Most are barren. There are several critical factors required to form a large porphyry deposit: the metal endowment of the primary magma (McInnes et al. 1999; Mungall 2002), the metal content of the evolved magma at the time of volatile exsolution (Jenner et al. 2010; Wilkinson 2013; Park et al. 2015), the efficient transfer and precipitation of the ore metals by the magmatic fluid (Richards 2003;



Audétat and Simon 2012; Wilkinson 2013), and the volume of magma and the duration of magmatic activity (Chelle-Michou et al. 2017; Chiaradia and Caricchi 2017). Metal transfer by degassing from a deep to shallow magma chamber (Halter et al. 2005; Chiaradia et al. 2012) and destabilization of magmatic sulfide (Nadeau et al. 2010; Mungall et al. 2015) have also been suggested as contributing factors in ore formation.

The chalcophile metal contents in a magma, the chalcophile element fertility, at the point of fluid exsolution have been considered to be one of the key factors in the formation of porphyry  $\text{Cu} \pm \text{Au}$  deposits (e.g., Wilkinson 2013), but there is no empirical evidence supporting this hypothesis. This is mainly due to the difficulties in measuring the metal contents of the magma from which the ore-forming fluid exsolved. The Cu and Au contents of igneous rocks, particularly those associated with porphyry  $\text{Cu} \pm \text{Au}$  deposits, are often remobilized by hydrothermal alteration and do not reflect magmatic composition.

Recent studies on platinum group element (PGE) geochemistry of subvolcanic rocks associated with porphyry  $Cu \pm Au$  deposits (Cocker et al. 2015; Hao et al. 2017; Lowczak et al. 2018) have suggested that the relative timing of fluid and sulfide saturation plays an important role in the formation of porphyry  $Cu \pm Au$  deposits. These studies used PGEs to indicate chalcophile element fertility at fluid saturation. The PGEs are used in preference to Cu and Au for two reasons. First, their partition coefficients in immiscible sulfide melts are one to two orders of magnitude higher than those of Cu and Au respectively (Mungall and Brenan 2014). Second, because they are appreciably less affected by alteration than Cu and Au, they are more likely to preserve original igneous geochemistry (e.g., Crocket 2000; Park et al. 2016).

In this paper, we evaluate the role of chalcophile element fertility in determining the ore-forming potential of porphyry systems by comparing the PGE geochemistry of barren suites with that of economic Cu and Cu-Au suites. We measured PGE contents of three barren volcanic and intrusive suites, the Rachaite suite from Argentina, and the Ryoke and Sanyo ilmenite-series and Sanin magnetiteseries granitoids from Japan (Fig. 1). We also determined PGE contents of subvolcanic rocks associated with the Chuquicamata porphyry Cu (Chile) and the Grasberg porphyry Cu-Au deposits (Indonesia) for comparison (Fig. 1). These results will be discussed together with literature data from the El Abra porphyry Cu system in Chile, the Northparkes and Cadia Au-Cu porphyry systems in Australia (Cocker et al. 2015; Hao et al. 2017; Lowczak et al. 2018), and the Bingham porphyry Cu-Au system in the USA (Maughan et al. 2002).



#### **Barren suites**

The Rachaite complex is a small calc-alkaline suite of basaltic andesite to dacite flows in northwest Argentina that erupted between 7.2 and 8.4 Ma (Kay et al. 2010). It includes some of the least evolved lavas in the northern Puna Volcanic Complex. Whole-rock major and trace element geochemistry and Sr, Nd, Pb, and O isotopic data suggest that these northern Puna volcanic rocks crystallized from hybrid magmas that formed by mixing of mantle-derived magmas with pelitic crustal melts in a 50 to 75 km, contractionally thickened crust (Kay et al. 2010). During ascent, the magma ponded at a depth of 20-25 km and underwent further assimilation-fractional crystallization (AFC). The Rachaite suite is associated with Pb, Zn, and Ag mineralization, but not with Cu or Au mineralization (Coira and de Brodtkorb 1995). Ten samples from Kay et al. (2010), which range from olivine-bearing basalt to glassy dacite, were analyzed for PGE and selected trace elements.

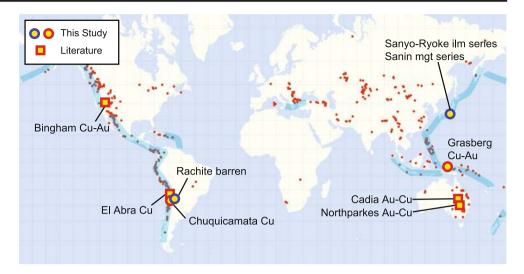
The Cretaceous-Paleogene granitoids in the Japan arc are classified into two groups, the reduced Ryoke and Sanyo ilmenite-series and the oxidized Sanin magnetite-series, based on their magnetic susceptibility (Kanaya and Ishihara 1973). The Ryoke and Sanyo suites are dominant in the southern parts of the Japan arc where the basement is mainly a Jurassic accretionary complex. In contrast, the Sanin magnetite-series granitic rocks are mainly distributed over the northern part of the arc (Takagi 2004). There are no porphyry deposits associated with the Cretaceous-Paleogene Japanese granitoids (Ishihara 2007), whereas the Sanin magnetite-series host Mo vein and Pb-Zn skarn deposits and the Sanyo-Ryoke granitoids are associated with W deposits and locally a few minor Cu skarn deposits (Ishihara 2007). Nine granitic samples from the Sanin suite and 11 samples from the Sanyo-Ryoke suite were analyzed in this study. Sample locations are reported in Electronic Supplementary Material (ESM) Fig. 1 and ESM Table 1.

# Porphyry Cu ± Au suites

The Chuquicamata porphyries, which host one of the largest porphyry Cu deposits in the Andes, are located at the southern end of the Chuquicamata Intrusive Complex. They intrude a pre-Cenozoic igneous, sedimentary, and metamorphic basement along the late Eocene to Oligocene West Fissure strike-slip fault zone (Ballard et al. 2001). The mineralized intrusive suite consists of the East, West, and Bench porphyries, which are composed of granodiorite, monzogranite, and monzodiorite. Six of the least altered samples from Ballard et al. (2001) were selected for this study.



Fig. 1 Locations of volcanic and subvolcanic rock suites included in this study. The red circles represent the locations of active porphyry copper mines (Singer et al. 2005). Thick light blue solid lines represent subduction zone



The Grasberg Cu-Au porphyry is the world's largest Au producer (Cooke et al. 2005). The deposit is situated within the Grasberg-Ertsberg porphyry-skarn mineralized district in the highlands of Papua, Indonesia. The Grasberg deposit is associated with a group of brecciated diorites to quartz monzonites intrusive units that form a pipe-shaped body, which intrudes the siliciclastic and carbonate units of the New Guinea Limestone Group (MacDonald and Arnold 1994; Leys et al. 2012). Five of the least altered samples from the Main Grasberg Intrusions, which are most closely associated with the Grasberg porphyry mineralization, and Kali intrusions were included in this study.

## **Analytical methods**

## Major and trace elements

Major and trace elements of some Rachaite samples were measured by EPMA and LA-ICP-MS at ANU using the method of Campbell (2003) and the results are reported in Electronic Supplementary Material (ESM Table 2). Synthetic glass beads were prepared using 0.375 g of sample powder and 0.75 g of lithium borate flux (12:22 Li<sub>2</sub>B<sub>4</sub>O<sub>7</sub>/LiBO<sub>2</sub> mass ratio). The EPMA was performed by a CAMECA SX100 electron microprobe in WDS mode with the beam conditions of 15 kV and 20 nA. The LA-ICP-MS analysis was carried out at a laser repetition rate of 5 Hz with a spot size of 105 µm. Each spot analysis consisted of 30 s of background measurement followed by 40 s of sample ablation. The BCR-2g (USGS) standard was analyzed before and after analyses of 10 unknown samples. The NIST 612 was used to monitor quality control. Ca concentrations measured by XRF (Kay et al. 2010) or EPMA (this study) were used as an internal standard to determine the absolute concentrations of the elements. Analytical uncertainties obtained by multiple analyses of

NIST 612 were  $\pm$  1 to 4% (1 $\sigma$ ) depending on the elements. Analytical accuracy is better than 10% for all elements, except for Cu (25%) and Cr (18%). The results are shown in ESM Table 3 where our measured values for NIST 612 are compared with the reference values from GeoReM database (Jochum et al. 2005).

#### **PGE and Re**

The whole-rock PGE and Re contents of the samples were determined by the Ni-sulfide fire assay-isotope dilution ICP-MS method described by Park et al. (2012b). The results are reported in Table 1. The sensitivities during analysis were  $\sim 3 \times 10^5$  cps/ ppb for mass 89, 140, and 205 using Agilent 7700x quadrupole ICP-MS at ANU. The Ir, Ru, Pt, Pd, and Re contents were measured by isotope dilution using the  $^{99}\mathrm{Ru}/^{101}\mathrm{Ru}.$   $^{191}\mathrm{Ir}/^{193}\mathrm{Ir}.$ <sup>195</sup>Pt/<sup>194</sup>Pt, <sup>105</sup>Pd/<sup>108</sup>Pd, and <sup>185</sup>Re/<sup>187</sup>Re ratios, respectively. Rhodium contents were determined by the method described in Meisel et al. (2003), which used the count rate of <sup>103</sup>Rh and <sup>106</sup>Pd assuming that the recovery rates of Rh and Pd were similar to each other during the analytical procedure. Molecular interferences on PGEs by Ni, Cu, Zn, Co, Hf, Mo, Zr, and Ta argides or oxides were monitored, and the effects of the interferences were corrected using measured oxide (0.3-0.4%) and argide production rate (0.003-0.006%). The interference on Ir, Pt, and Re by the metal oxide and argide is negligible. The interferences on <sup>105</sup>Pd and <sup>108</sup>Pd by <sup>65</sup>Cu<sup>40</sup>Ar, <sup>68</sup>Zn<sup>40</sup>Ar, <sup>102</sup>Zr<sup>16</sup>O, <sup>102</sup>Mo<sup>16</sup>O, and <sup>108</sup>Cd are corrected. The differences between data before and after the correction are < 2%. There is significant interference on <sup>101</sup>Ru by <sup>61</sup>Ni<sup>40</sup>Ar. The differences for Ru ranges from 9 to > 100% due to low contents of Ru in samples. Procedural blanks were measured from the sample-free analysis. The average procedural blanks and the method detection limits that are taken as three standard deviations of the procedural blanks are reported in ESM Table 4. The accuracy and precision of the analyses were assessed by multiple measurements (n = 7)



 Table 1
 Platinum group element and Re concentrations in volcanic and subvolcanic rocks

Sample	MgO (wt.%)	Cu (ppm)	Ir (ppb)	Ru (ppb)	Rh (ppb)	Pt (ppb)	Pd (ppb)	Re (ppb)	Pd/MgO $\times$ 10 <sup>9</sup>	Pd/Pt
Volcanic rocks from the	Rachaite s	suite, Argen	tina							
P17-2A	1.80	10	0.0015	< 0.0014	0.0016	0.039	0.080	0.019	0.044	2.0
P17-2B	1.89	15	0.0007	< 0.0014	< 0.0003	0.019	0.047	0.071	0.025	2.4
P17-3	0.77	11	< 0.0003	< 0.0014	< 0.0003	< 0.005	< 0.009	< 0.012	0.012	1.8
P17-5	1.54	4	< 0.0003	< 0.0014	< 0.0003	< 0.005	0.014	0.050	0.009	2.7
P17-7	1.50	7	< 0.0003	< 0.0014	< 0.0003	< 0.005	0.016	< 0.012	0.010	3.1
RA95-1	2.20	11	0.0055	0.0086	0.0029	0.030	< 0.009	0.079	0.004	0.3
RA95-2	3.54	17	0.0007	0.0066	< 0.0003	0.015	0.032	< 0.012	0.009	2.1
RA95-3	3.66	9	0.0014	0.0047	0.0012	0.033	0.017	0.012	0.005	0.52
P17-4	4.42	30	0.0098	0.0020	0.0011	0.071	0.046	0.066	0.010	0.65
P17-6	5.83	4	0.0281	0.0179	0.0183	0.344	0.214	0.038	0.037	0.62
Granitoids from the Sar		Sanin magi								
Sanyo-Ryoke magne										
SJ3	0.42	n.d.	0.0009	0.005	0.002	< 0.018	0.035	0.064	0.082	
SJ5	0.15	n.d.	< 0.0002	0.006	0.001	< 0.018	0.004	0.116	0.024	
SJ5 duplicate	0.15	n.d.	0.0003	0.004	< 0.0001	< 0.018	0.004	0.147	0.023	
SJ6	0.96	n.d.	0.0004	0.003	0.0003	0.035	0.012	0.907	0.012	0.33
SJ6 duplicate	0.96	n.d.	< 0.0002	< 0.003	0.0003	0.070	0.010	0.838	0.010	0.14
SJ8	0.50	n.d.	0.0006	0.006	0.002	0.019	0.009	0.120	0.019	0.49
SJ9	0.60	n.d.	< 0.0002	< 0.003	0.0003	0.037	0.009	0.253	0.016	0.25
SJ10	3.10	n.d.	0.0006	0.008	0.005	0.025	0.020	0.234	0.007	0.80
SJ10 duplicate	3.10	n.d.	0.0004	0.030	0.008	0.023	0.020	0.402	0.007	0.89
SJ27	0.79	n.d.	0.0006	< 0.003	0.001	0.023	0.008	0.107	0.010	0.33
SJ39	1.32	n.d.	0.0008	< 0.003	0.003	< 0.018	0.015	0.033	0.012	0.55
SJ42-1	0.34	n.d.	0.0003	< 0.003	0.002	0.019	0.008	0.347	0.024	0.41
Sanin ilmenite-series		ii.d.	0.0005	V 0.005	0.002	0.019	0.000	0.5 17	0.021	0.11
SJ11	5.98	n.d.	0.0002	0.004	< 0.0001	< 0.018	0.008	0.878	0.001	
SJ11 duplicate	5.98	n.d.	< 0.0002	0.007	0.0003	< 0.018	0.009	1.059	0.002	
SJ17 dapheate SJ17	1.00	n.d.	0.0002	0.007	0.0003	< 0.018	0.014	0.256	0.014	
SJ17 duplicate	1.00	n.d.	0.0003	< 0.003	0.003	< 0.018	0.011	0.263	0.011	
SJ18	0.12	n.d.	0.0003	< 0.003	0.0003	0.037	0.006	0.203	0.048	0.16
SJ18 duplicate	0.12	n.d.	0.0005	0.005	0.0003	0.025	0.006	0.118	0.052	0.10
SJ22	0.12	n.d.	< 0.0003	< 0.003	0.001	0.023	0.008	0.148	0.052	0.25
SJ45	0.13	n.d.	0.0002	0.003	< 0.0001	0.080	0.003	0.038	0.005	0.03
SJ50	1.45	n.d.	< 0.0007	0.007	0.0001	< 0.018	0.002	0.021	0.003	0.02
SJ50 duplicate	1.45	n.d.	< 0.0002	< 0.003	< 0.002	< 0.018	< 0.001	0.021	0.0007	
SJ51	5.91	n.d.	0.0002	< 0.003	0.001	0.042	0.019	0.319	0.003	0.45
SJ54	1.93	n.d.	0.0003	0.004	0.001	0.042	0.231	0.319	0.120	2.5
SJ60	0.18	n.d.	< 0.0028	0.004	0.003	< 0.018	0.231	0.067	0.120	2.5
SJ60 duplicate	0.18	n.d.	0.0002	< 0.003	0.033	< 0.018	0.013	0.059	0.069	
SJ61	1.05	n.d.	< 0.0003	0.003	0.0005	0.037	0.015	0.513	0.005	0.15
SJ61 duplicate	1.05	n.d.	0.0002	< 0.003	0.0003	0.037	0.003	0.313	0.003	0.13
SJ64-1	0.37	n.d.	0.0003	0.003	0.0003	< 0.020	0.011	0.438	0.011	0.50
SJ64-1 duplicate	0.37	n.d.	0.0004	< 0.003	0.0002	< 0.018	0.018	0.209	0.043	
Subvolcanic rocks from					0.001	< 0.010	0.016	0.237	U.U <del>1</del> 7	
98-508	0.56	ncamata po 397	< 0.0003	< 0.0014	< 0.0003	0.019	0.057	1.00	0.102	3.0
										2.0
										1.0
98-511 98-503	0.63 0.64	2615 11,213	0.0030 < 0.0003	0.0040 < 0.0014	< 0.0003 0.0015	0.036 0.031	0.071 0.031	1.62 6.28	0.113 0.048	



Table 1 (continued)

Sample	MgO (wt.%)	Cu (ppm)	Ir (ppb)	Ru (ppb)	Rh (ppb)	Pt (ppb)	Pd (ppb)	Re (ppb)	Pd/MgO × 10 <sup>9</sup>	Pd/Pt
98-601	0.42	1221	< 0.0003	< 0.0014	< 0.0003	0.094	1.88	4.79	4.485	20.1
98-505	0.59	600	< 0.0003	< 0.0014	< 0.0003	0.009	0.106	0.65	0.178	11.9
98-506	0.44	155	0.0025	< 0.0014	< 0.0003	0.015	0.080	0.79	0.182	5.2
Subvolcanic rocks from	m the Grasbe	erg porphyr	y Cu-Au depo	osit, Indonesia	a					
Gras11	1.36								0.310	1.8
Gras2	1.45								0.426	1.8
Gras12	1.69								0.548	2.7
Gras13	2.27								1.852	2.0
Gras14	1.77								3.147	8.9

Numbers in italics are calculated using the limit of detection values *n.d.* not determined

of reference material TDB-1. The results are shown in ESM Table 5. Reproducibility represented as the relative standard deviation (1 $\sigma$ ) is < 10% for Pt, Rh, Ru, and Re for 14% for Pd and 23% for Ir. The measured values are consistent with previously reported values within 2 $\sigma$ .

## **Results and discussion**

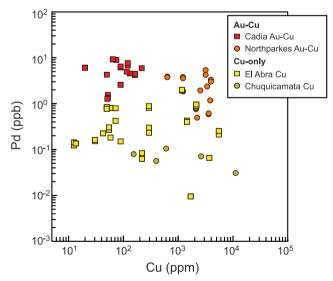
The major and trace element and PGE contents of the Chuquicamata Cu porphyry and Grasberg Cu-Au porphyry, obtained for this study, are presented in Table 1 and ESM Table 2. These results will be discussed and compared with those of previous studies on the El Abra Cu porphyry in Chile (Cocker et al. 2015), the Northparkes and Cadia Au-Cu porphyries in Australia (Hao et al. 2017; Lowczak et al. 2018), and the Bingham porphyry Cu-Au in the USA (Maughan et al. 2002).

## Effect of alternation and mineralization

Because the emphasis of this study is on primary magmatic processes, the samples analyzed were the freshest possible. Nevertheless, some of the porphyry-related rocks used in this and previous studies contain some hydrothermal sulfides, including hydrothermal Cu. Furthermore, some mineralized porphyry systems are known to contain hydrothermal Pd and Pt (Economou-Eliopoulos 2005 and references therein), showing that hydrothermal fluids can mobilize Pd and Pt and, in rare cases, porphyry Cu deposits can contain up to 2400 ppb Pd (e.g., Skouries Cu porphyries, Greece; Economou-Eliopoulos and Eliopoulos 2000). However, empirical geochemical investigations (Yudovskaya et al. 2008; Tagirov et al. 2013; Zelenski et al. 2013, 2015; Park et al. 2016), supported by the results of experimental

and thermodynamic studies (Wood 1987; Sassani and Shock 1990; Fleet and Wu 1993, 1995; Taran et al. 1995; Hanley et al. 2005; Simon and Pettke 2009), show that the PGEs are normally appreciably less mobile than Cu and Re in magmatic-hydrothermal fluids.

The relative mobilities of Cu and Pd in the felsic suites considered in this study can be assessed from Fig. 2, which shows a plot of Cu against Pd for the porphyries associated with the Cu and Au-Cu deposits discussed in this paper. Copper varies by  $\times$  100 in samples from the Chuquicamata Cu porphyry and by almost  $\times$  10,000 in El Abra samples with no evidence of a correlated increase in Pd as might be expected if Pd was an important component in the hydrothermal



**Fig. 2** Plot of Pd against Cu for the ore-bearing and barren suites included in this study. The Cadia, Northparkes, and El Abra data are from Lowczak et al. (2018), Hao et al. (2017), and Cocker et al. (2015) respectively



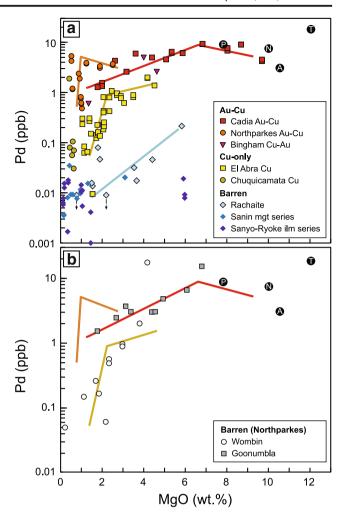
fluid. For example, the least mineralized sample from Chuquicamata (98–506), with 155 ppm Cu, and the most mineralized sample (98–503) with 11,213 ppm Cu, have similar PGE contents (Table 1). The Cu content of the samples from the Au-Cu deposits vary by only × 10, so the influence of increasing Cu on Pd is less clear. What is clear is that there is no correlation between these elements and that samples from the Cu-rich Northparkes deposit contain less Pd than those from Cu-poor Cadia.

## PGE geochemistry of ore-bearing suites

Figure 3 shows the Pd concentrations of the barren, Cu porphyry and Cu-Au porphyry suites plotted against MgO. The Pd contents of the volcanic and subvolcanic rocks associated with the Cadia and Northparkes Au-Cu suites are generally > 3 ppb and remain enriched until late in magma differentiation (< 1 wt.% MgO). The PGE contents of the highest MgO samples are comparable to those of primitive arc picrites and basalts (Fig. 3a), suggesting that they experienced little (Cadia) or no (Northparkes) sulfide segregation while fractionating in crustal magma chambers until a late stage of magma differentiation (Hao et al. 2017; Lowczak et al. 2018). Once the magma became sulfide-saturated PGE concentrations dropped rapidly due to their strong affinity for immiscible sulfide melts  $(D^{\text{sul-sil}}_{PGEs} > \sim 1 \times 10^6;$ Mungall and Brenan 2014). The Pd of the melt follows Cu and Au because all of these elements are incompatible with silicate and oxide minerals but strongly compatible with sulfide melts (e.g., Park et al. 2013, 2015). Therefore, the Cadia and Northparkes magmas, which reached sulfide saturation late in their history, had sufficient Cu and Au at the time of fluid exsolution to form Au-Cu ore deposits (Hao et al. 2017; Lowczak et al. 2018).

The Pd contents of intrusive rocks associated with the Chilean El Abra porphyry Cu deposit are seen to be consistently lower than those associated with the Australian porphyry Au-Cu deposits at the same MgO content (Fig. 3a). The decrease in the abundance of Pd seen in the El Abra samples with < 2.5 wt.% MgO suggests that sulfide segregation was taking place before the onset of volatile saturation which was placed at 1.1-1.5 wt.% MgO by Cocker et al. (2015). Cocker et al. (2015) further suggested that the El Abra magma formed a "Cu-only" porphyry deposit because small amounts of sulfide segregation during fractional crystallization before fluid saturation depleted the magma in the more chalcophile Au and not the less chalcophile Cu  $(D^{\text{sul-sil}}_{\text{Au}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{\text{sul-sil}}_{\text{Cu}} = \sim 1 \times 10^4 \text{ and } D^{$ 10<sup>3</sup>; Ripley et al. 2002; Mungall and Brenan 2014 and references therein).

The intrusive rocks associated with the Chuquicamata porphyry Cu deposit have similar Pd contents to the El Abra porphyries with the contents in both being at least an order



**Fig. 3** a Plot of Pd against MgO for the ore-bearing and barren suites included in this study. The colored lines are references, which show variations in Pd during magma differentiation for each suite. The kinks in the reference lines show the point of sulfide saturation. Black dotted arrows represent data below the detection limit. Black circles with letters are average Pd contents for arc basalts and picrites with MgO contents between 8 and 14 from different localities. P = Pual Ridge basalt (sample MD3; Park et al. 2013), N = Niuatahi-Motutahi basalts (average of samples DG26-2 and DG26-3; Park et al. 2015), A = Ambae picrite (sample 68638; Park et al. 2012a), T = Tonga picrites (average of samples ND03-01 and 04-01; Dale et al. 2012). **b** Plot of Pd against MgO for the barren Wombin and Goonumbla suites from Northparkes. Reference lines for three of the ore-bearing suites in **a** are shown for comparison

of magnitude higher than those in barren suites at a similar wt.% MgO. This implies that the Chuquicamata magma had a chalcophile element budget similar to that of the El Abra magma at fluid saturation, which can explain why Chuquicamata is also a Cu-only deposit.

## PGE geochemistry of barren suites

The PGE contents of barren rocks from the Rachaite suite, Argentina, and the Sanyo-Ryoke-Sanin suites, Japan, are



systematically depleted compared with the Cu ± Au orebearing igneous suites at the same MgO content (Fig. 3a and ESM Fig. 2). This suggests that either they have lost most of their chalcophile element budget by sulfide segregation at depth or that they originated from a chalcophile elementpoor source. The Pd and Pt contents of the Rachaite volcanic rocks gradually decrease with decreasing MgO from 5.8 to 0.8 wt.% (Fig. 3a and ESM Fig. 2). These trends are different from those typically observed in sulfide-saturated magmatic systems in which Pd and Pt drop rapidly after sulfide saturation (Fig. 3 and ESM Fig. 2). As already noted, the Rachaite suite magmas were produced by mixing between mantlederived basaltic and pelitic crustal melts (Kay et al. 2010). We performed mixing and AFC model following methods reported in Kay et al. (2010). The detailed conditions and parameters for the AFC model are in ESM Appendix I. The results show that the major and trace element geochemistry of the Rachaite suite is best explained by mixing between the mafic and the felsic end-members (Fig. 4 and ESM Fig. 3). The Pd variations for most samples can also be ascribed to mixing between the two end-members, although some show depletion in Pd/Nd and fall below the mixing curve, suggesting sulfide segregation may have occurred locally during mixing (Fig. 4). If magmas become locally equilibrated with sulfide melt, Pd will decrease markedly with no corresponding change in Nd or Rb due to the high  $D^{\text{sul-sil}}\text{Pd} > \sim 1 \times 10^5$  (Mungall and Brenan 2014), resulting in low Pd/Nd and Pd/Rb ratios.

The Pd content of the mafic end-member, which is an order of magnitude lower than that of primitive arc magmas (Fig. 3a), suggests that the mafic magma has lost some chalcophile elements before mixing with the felsic crustal magma. We suggest that the Rachaite primary magma lost its fertility through a combination of early sulfide segregation in the lower crust and mixing with sulfide-saturated, a chalcophile element-poor felsic magma in a mid-crustal chamber. This process may have depleted the Rachaite magma in Au and Cu, but not in Pb and Zn as it hosts Cu-poor, Pb-Zn polymetallic deposits (Coira and de Brodtkorb 1995). The

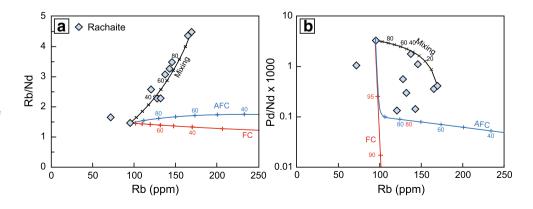
Rachaite case provides an example of how a chalcophile element-rich mantle-derived magma became barren during crustal evolution.

The Sanin-Sanyo-Ryoke granitoids, Japan, are characterized by marked depletion in all PGEs, including the mafic samples with > 3 MgO wt.% (Fig. 3a), which suggests that the Sanin-Sanyo-Ryoke magma may have either undergone extensive sulfide segregation in a mid-lower crustal magma chamber or that the primary magma was chalcophile element-poor. There have been several suggestions as to why there are no porphyry Cu deposits associated with the Cretaceous-Paleogene intrusive rocks of southeastern Japan, including shallow erosion depth and lack of thick continental crust (Qin and Ishihara 1998). However, our results show that the low chalcophile element fertility can be an important factor.

The Sanyo-Ryoke ilmenite-series granitoids are represented by high initial  $^{87}$ Sr/ $^{86}$ Sr ratios between 0.7050 and 0.7106 and high  $\delta^{18}$ O (> 8 % $_{o}$ ), suggesting the extensive involvement of crustal materials during their magma generation (Ishihara 2007 and references therein). The primary magma of the Sanyo-Ryoke ilmenite-series granitoids is interpreted to have formed by mixing between mantle-derived basaltic magma and partial melts of carbonaceous meta-sedimentary rocks (Ishihara 2007). Assimilation of sedimentary carbon into the magma will lower its oxygen fugacity and therefore its sulfur solubility. As a consequence, the Sanyo-Ryoke magmas are expected to leave sulfide in their source region following partial melting and to reach sulfide saturation early in their evolution. They are not prospects for Cu and Au mineralization.

The Sanin magnetite-series granitoids are interpreted to have originated from mantle-derived basaltic melt that assimilated carbon-poor meta-igneous basement rocks (Ishihara 2007). Their relatively low  $\delta^{18}$ O ( $\sim 5.5-8\% c$ ; Ishihara 2007), compared to those of the ilmenite-series, imply lesser amounts of crustal input during their genesis. The Sanin magnetite-series granitoids are oxidized relative to the ilmenite-series because they had little interaction with carbon-rich materials

Fig. 4 Mixing, AFC and FC modeling for plots of a Rb/Nd and b Pd/Nd against Rb for the Rachaite suite. Labeled markings on AFC and FC curves represent the percentage of melt (F) remaining in a system. Numbers on the mixing curves indicate the percentage of the mafic endmember in a system





during melting. This has led to the suggestion that they may have the potential for porphyry Cu deposits (Qin and Ishihara 1998). However, our results show that their chalcophile element depletion is similar to that of the ilmenite-series granitoids, which suggests that either the primary Sanin magma was deficient in chalcophile elements or that it was not sufficiently oxidized to exhaust all the sulfide phases in the source during melting.

# Northparkes barren suites

The igneous rocks exposed around the Northparkes Au-Cu suites in Australia include two barren suites, the Goonumbla and Wombin Volcanics (Lickfold et al. 2003). These Middle to Upper Ordovician lavas occur associated with monzodioritic intrusions with a zircon U-Pb age of  $\sim 450$  and 439 Ma, respectively (Butera et al. 2001). These ages coincide with the main Au-Cu mineralization period of the Northparkes deposit, which is between 445 and 437 Ma (Lickfold et al. 2003).

Figure 3b compares Pd variations in the Goonumbla and Wombin Volcanics with those of the Northparkes Au-Cu suites. The progressive depletion in Pd with decreasing MgO indicates that they have undergone early sulfide saturation and lost some Cu and Au to cumulus sulfide phases in a deep parental magma chamber (Hao et al. 2017). However, their Pd contents are still comparable to those of Cu- and Au-Cu suites at <3 wt.% MgO, implying some potential for forming a porphyry Cu  $\pm$  Au deposit.

The evolution of the Goonumbla and Wombin suites are characterized by plagioclase-pyroxene fractionation whereas the Northparkes Au-Cu suites are dominated by plagioclase-amphibole fractionation (Hao et al. 2017). They are also not associated with porphyries or magmatic-hydrothermal alteration (Crawford et al. 2007). These observations suggest that Goonumbla and Wombin magmas were dry and may not

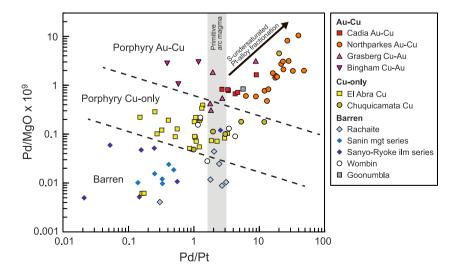
have experienced the extensive fluid exsolution required to form a porphyry Cu deposit despite their chalcophile element budgets being comparable to those of some other porphyry Cu  $\pm$  Au suites. The Northparkes barren suites show that although the chalcophile element fertility is a critical factor, it is not the only factor required to form a porphyry Cu  $\pm$  Au deposits.

# Chalcophile element fertility indicator

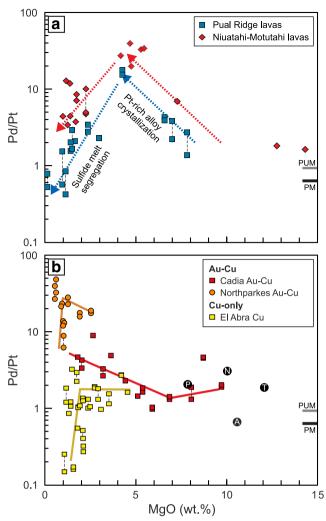
This study shows that the PGE can be used as a tracer for Cu and Au and their geochemistry reflects changes in chalcophile element fertility during magma evolution, confirming the results of recent studies on the PGE geochemistry of felsic-intermediate igneous rock suites (Cocker et al. 2015; Park et al. 2015; Hao et al. 2017; Lowczak et al. 2018).

We suggest that plots of Pd/MgO against Pd/Pt are indicators of chalcophile element fertility and can be used to test its role in the formation of porphyry  $Cu \pm Au$  deposits (Fig. 5). Palladium contents are normalized by MgO to correct for variations in Pd abundance caused by differences in fractional crystallization. The Pd/Pt ratio is used as a measure of the amount of fractional crystallization before sulfide saturation. This is because prior to sulfide saturation, Pd behaves incompatibly whereas Pt typically is sequestered by Pt-Fe alloys during fractionation of arc magma systems so that the Pd/Pt ratio increases with fractionation (Fig. 6a; e.g., Park et al. 2013, 2016). Once the magma becomes sulfide saturated, its Pd/Pt ratio begins to fall because Pd is more chalcophile than Pt (Fig. 6a; Park et al. 2016; Hao et al. 2017). The Northparkes and El Abra suites exhibit an abrupt decrease in Pd/Pt ratio after the onset of sulfide segregation (Fig. 6b). The seemingly opposite trend for the Cadia suite may be ascribed to their open system evolution (e.g., magma replenishment, Lowczak et al. 2018).

Fig. 5 Chalcophile element fertility diagram showing the difference between barren, Cuonly, and Cu-Au suites. Samples with < 2.5 wt.% MgO were plotted except for the Cadia and Bingham suites, for which samples with < 3.2 wt.% MgO and 1.4-4.6 wt.% MgO respectively were used. The barren, porphyry Cu, and porphyry Cu-Au bearing suites can be divided by dotted lines. The gray field represents a range of Pd/Pt values of primitive arc magmas (Dale et al. 2012; Park et al. 2012a, 2013, 2015)







**Fig. 6** Plot of Pd/Pt against MgO for **a** the Niuatahi-Motutahi lavas, Tonga rear arc (Park et al. 2015), and the Pual Ridge lavas, Eastern Manus Basin (Park et al. 2013) and **b** the Northparkes, El Abra, and Cadia suites. The Pd/Pt ratios of some Niuatahi-Motutahi lavas (diamonds with dotted line) are minimum estimation values obtained using Pd data of each sample and Pt data of DG26-3 in Park et al. (2015). Upward trending arrows result from Pt-rich alloy crystallization, downward trend lines sulfide melt segregation, and flat trend the absence of both alloy and sulfide precipitation. The primitive upper mantle (PUM) and primitive mantle (PM) values are from Becker et al. (2006) and McDonough and Sun (1995) respectively. P = Pual Ridge basalt (sample MD3; Park et al. 2013), N = Niuatahi-Motutahi basalts (average of samples DG26-2 and DG26-3; Park et al. 2015), A = Ambae picrite (sample 68638: Park et al. 2012a), T = Tonga picrites (average of samples ND03-01 and 04-01; Dale et al. 2012)

Bearing in mind that most porphyry  $\text{Cu} \pm \text{Au}$  deposits are hosted by highly evolved andesitic to rhyolitic rocks, only samples with < 2.5 wt.% MgO are plotted on Fig. 5, which assumes that these low MgO samples are the most representative of the magma composition at the time of fluid saturation. Relatively mafic samples were included for the Cadia (< 3.2 wt.% MgO) and Bingham suites (1.4–4.6 wt.% MgO), as their  $\text{Cu} \pm \text{Au}$  mineralization is

associated with andesitic intrusions (Lowczak et al. 2018; Maughan et al. 2002).

Low Pd/MgO and Pd/Pt ratios distinguish most barren suites from the ore-bearing suites (Fig. 5) and porphyry Cu from porphyry Cu-Au suites. Most of the Cu-Au suites have Pd/Pt ratios higher than those of primitive arc magmas, suggesting they have experienced prolonged sulfide-undersaturated fractionation or limited sulfide melt segregation (e.g., Cadia), which increased their chalcophile element fertility. The Goonumbla and Wombin barren suites have chalcophile element fertility similar to those of porphyry Cu suites but do not have Cu mineralization due to the low water content of their magmas.

## Implications for the porphyry ore-forming process

The seminal study of Cline and Bodnar (1991) and recent numerical modeling studies (Chelle-Michou et al. 2017; Chiaradia and Caricchi 2017) argued that intrusion volume and duration of magma activity are the first-order factors controlling metal endowments, and particularly metal-enriched magmas are not essential for the genesis of economic porphyry Cu deposits. Chelle-Michou et al. (2017) suggested that any "normal" hydrous arc magmas have potential to produce fluids capable of forming porphyry Cu deposits and can form large porphyry deposits (> 20 Mt Cu) if they have long-lasting activity (> 1 Ma) and large volume (> 1000 km<sup>3</sup>) regardless of metal contents in a magma.

Two of the critical factors that control the economics of a porphyry system are its size and grade. The former is logically controlled by the size of the magma event associated with the hydrothermal mineralization and the latter by a combination of chalcophile element fertility and efficient transfer of metals from the melt to the fluid. Therefore, the higher the Cu and Au contents of the ore-associated magma at the time of volatile saturation, the higher tenor of the ore for a given fluid/melt partition coefficient for Cu and Au. The ore grade and tonnage data for the porphyry deposits considered in this study are presented in Table 2. The Northparkes and Grasberg magmas, which have the highest chalcophile element fertility at fluid saturation (Fig. 5; Table 2), are associated with the highest Cu ore grades of > 1 wt.% among studied suites. Given the tonnage of the Northparkes deposit is  $\sim 0.15$  Gt, which is at least an order of magnitude smaller than those of other porphyry deposits in this study (Table 2), the magma volume involved in the formation of the Northparkes deposit may have been at least an order of magnitude smaller than other deposits. However, the Northparkes magma formed an economically significant porphyry Au-Cu deposit (1.6 Mt Cu and 70 t Au; Cooke et al. 2007) due to high chalcophile element fertility. Magma composition appears to be particularly controlling the Au grade of a porphyry deposit. Figure 7 shows that the Au grades of porphyry ores are positively correlated with the Pd



2

3

2, 4

Grasberg

Bingham<sup>b</sup>

Cadia

2.5

1.3

3.2

Deposit Tonnage (Gt) Cu (Mt) Au (t) Cu (wt.%) Au (g/t) Au/Cu grade Pd/Pt Pd/MgO References for t he ore deposit data Cu-only 0.49  $0.01^{a}$ 0.02 El Abra 1.8 8.8 18 1.2 0.2 1 Chuquicamata 7.5 66.4 301 0.55 0.04 0.07 7.2 0.9 2 Au-Cu Northparkes 0.15 1.6 70 1.03 0.46 0.45 22 2.5 3

1.05

0.74

0.38

0.93

2.39

0.43

 $\textbf{Table 2} \quad \text{Tonnage, metal endowment, and grade of the porphyry } \ \text{Cu} \pm \text{Au deposits included in this study and chalcophile element fertility indicators of the associated suites}$ 

References: 1 = Singer et al. (2008), 2 = Cooke et al. (2005), 3 = Cooke et al. (2007), 4 = Maughan et al. (2002)

2604

1050

1227

1.13

0.31

0.88

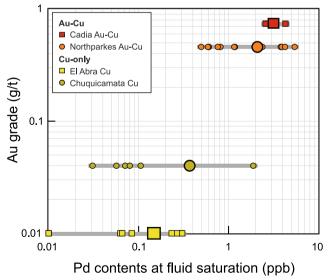
content of the magma at fluid saturation, as determined from the Pd content of the ore-associated porphyries. This observation suggests that the Au grade of the studied deposits is controlled by the Au contents of the magma at fluid saturation.

30.5

4.4

28.5

Mantle-derived arc magmas can be expected to undergo assimilation and fractional crystallization as they ascend through the crust and this will modify their compositions. The principal processes that control the Cu and Au contents in magmas at volatile exsolution are the relative timing of fluid and sulfide saturation. Furthermore, because Au is more



**Fig. 7** Plot of Au grade in the ore against Pd concentration in the melt at fluid saturation. The large symbols represent the average of each suite. References for the Au grades are given in Table 2. MgO contents of the samples used to constrain the Pd concentrations in the melt at fluid saturation are 0.5–1.5 wt.% for the Northparkes (Hao et al. 2017), 1.0–1.5 wt.% for the El Abra (Cocker et al. 2015), 2.5–3.5 wt.% for the Cadia (Lowczak et al. 2018), and 0.4–0.7 wt.% for the Chuquicamata suite (this study)

chalcophile than Cu, sulfide saturation leads to marked changes in the Au/Cu ratio of the melt. Figure 8 shows a theoretical model of the effect of sulfide saturation relative to fluid saturation using the Rayleigh fractionation equation after Cocker et al. (2015). Models 1 and 2 assume sulfide saturation at 30 and 70% fractionation, respectively, with an assumed sulfide melt fraction of 0.2 wt.% (e.g., Hao et al. 2017; Park et al. 2013, 2015) and gradual linear increase in the rate of sulfide segregation over a 10% fractionation interval (e.g., Keays and Tegner 2015). The sulfide melt and silicate melt partition coefficients ( $D^{\text{sul-sil}}$ Metal) used for these models are  $2 \times 10^5$  for Pd,  $1 \times 10^4$  for Au, and  $1.3 \times 10^3$  for Cu (Mungall and Brenan 2014 and references therein). Model 1 represents the Rachaite and Sanin-Sanyo-Ryoke cases (Fig. 8a). The magmas experienced early sulfide saturation due to the assimilation of reduced C-rich crustal materials and lost most of their chalcophile element and sulfur budget in cumulates within a deep crustal magma chamber. Their chalcophile element fertility was therefore low at the time of fluid saturation, which is assumed to occur long after sulfide saturation. Magmas from these suites are expected to be barren.

3.4

4.5

3.2

1.3

0.9

0.7

Model 2 represents suites associated with porphyry  $Cu \pm Au$  deposits, such as those included in this study. If fluid saturation occurs at, or slightly after, 70% fractionation, the melts are enriched in both Cu and Au and therefore have the potential to form an Au-rich porphyry Cu deposit (Fig. 8b). It should be noted that 70% fractionation of a sulfide-undersaturated magma differentiation increases not only Cu and Au but also volatile elements such as S and Cl in the melt by more than a factor of three times the initial values, which will significantly improve the ore grade and reduce the duration and volume of magmatism required for economic porphyry  $Cu \pm Au$  deposits. The PGE geochemistry of the Northparkes suite is consistent with this model (Fig. 3; Hao et al. 2017) and the high Pd/MgO and Pd/Pt ratios of the Grasberg samples ( $\sim 1.5$  wt.% MgO; Fig.



<sup>&</sup>lt;sup>a</sup> Assumed value, no information due to Au is not present in detectable concentrations (Singer et al. 2008)

<sup>&</sup>lt;sup>b</sup> Pd/Pt and Pd/MgO values were calculated by average of Tick-52A, 59A, and 69A from Maughan et al. (2002)

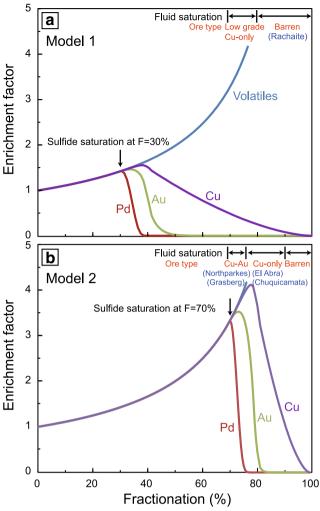


Fig. 8 Models simulating the Pd, Au, and Cu contents in the silicate melt as a function of fractionation with sulfide saturation at a 30% fractionation and b 70% fractionation. The single-headed arrow represents the assumed timing of sulfide saturation. The double-headed arrows represent the assumed timing of fluid exsolution and the expected ore types formed from the fluid

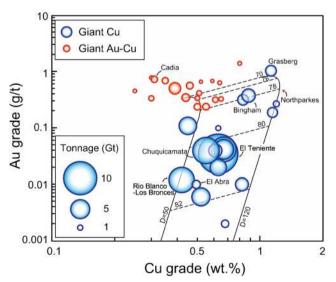
5) suggest that they also remained sulfide undersaturated until the final stages of magma differentiation.

If fluid saturation occurs after sulfide saturation at  $\sim\!80\%$  fractionation, the 10% fractionation interval between sulfide and fluid saturation can deplete the melt in Au without removing a significant amount of Cu because the partition coefficient of Au into an immiscible sulfide melt is about ten times higher than that of Cu (Mungall and Brenan 2014 and references therein; Fig. 8b). The result is a "Cu-only" deposit as seen in the El Abra (Cocker et al. 2015) and Chuquicamata suites (this study).

An apparent exception is Cadia, which reached sulfide saturation at 6.8 wt.% MgO (Fig. 3a), well before fluid saturation at  $\sim 3$  wt.% MgO (Lowczak et al. 2018). Nevertheless, the Cadia magma retained enough Cu and Au to form a porphyry Au-Cu deposit due to a combination of the fraction of sulfide

melt to precipitate from the silicate melt (0.023 wt.%) being about a factor of ten less than other arc-related magmas ( $\sim$ 0.2 wt.%; Hao et al. 2017; Park et al. 2013, 2015) and the *R* factor being less than  $D_{\rm Au}$  (*R* = silicate melt/sulfide melt ratio = 14,000; Lowczak et al. 2018). The small fraction of sulfide melt segregation may be due to destabilization of the sulfide melt by interaction with volatile phases (e.g. Mungall et al. 2015).

The Cu and Au grades of the 37 largest known porphyry Cu and Au-Cu deposits (Cooke et al. 2005), including those investigated in this study, are presented in Fig. 9. For the giant porphyry Cu deposits (> 10 Mt Cu and Au/ Cu grade  $< 0.1 \times 10^{-3}$ ), the Cu and Au grades are positively correlated regardless of the tonnage of each deposit, suggesting that the duration and volume of magma activity are not the primary controls on the ore grade. Note that as the Cu grade decreases from 1.2 to 0.3 wt.%, the Au grade shows almost two orders of magnitude depletion from  $\sim 0.4$  to  $\sim 0.002$  g/t. Fluid composition and emplacement depth have been considered to be significant factors controlling the ore grade and Au/Cu ratio (Murakami et al. 2010). However, we suggest that this trend can also be explained by variations in chalcophile element fertility due to differences in the interval between sulfide and fluid saturation in each deposit. To test this hypothesis, we calculated the Cu and Au grades of the ores precipitated from the fluid in equilibrium with the melt compositions predicted by model 2 assuming initial Cu and Au contents of 50 ppm and 4 ppb respectively (Lee et al. 2012; Park et al.



**Fig. 9** Plot of Cu and Au grades in ores for giant porphyry deposits, including those investigated in this study. Names of ore deposits with tonnages of > 5 Gt are also shown. Data are from Cooke et al. (2005, 2007). The solid lines represent the Cu and Au grades of the ore formed from the fluid in equilibrium with the melt for model 2 in Fig. 7, assuming  $D^{\text{fluid-melt}}$ Cu and  $D^{\text{fluid-melt}}$ Au of 50–120 and precipitation efficiency of 50%. The dashed lines are fractionation (%) from model 2 in Fig. 7



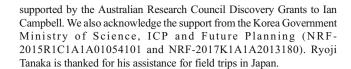
2015). The trends of the ore grades of the giant porphyry Cu deposits can be best explained if fluid saturation occurs between 70 and 82% fractionation in model 2, with D<sup>fluid-melt</sup>Cu and D<sup>fluid-melt</sup>Au of 50–120 and precipitation efficiency of 50% (e.g., Chiaradia and Caricchi 2017). These D values compare well with experimentally determined values (Simon et al. 2006; Simon et al. 2007). The results show that as the interval between sulfide and fluid saturation increases. the Au grade abruptly decreases with the slight decrease of the Cu grade, supporting for the role of chalcophile element fertility in controlling the ore grade of the giant porphyry Cu deposits. Although this calculation has considerable uncertainty introduced by the assumptions, which include the initial metal contents, the timing and degree of sulfide melt segregation, and precipitation efficiency, the positive correlation between Cu and Au grades is not dependent on these assumptions. Note that the porphyry Au-Cu deposits do not show positive correlations between Cu and Au grades, which suggests that their genesis involved a particularly Au-rich magma source (e.g., Richards 2009) or fluid (e.g., Murakami et al. 2010).

#### **Conclusion**

This study investigated the systematic differences in PGE geochemistry of barren and porphyry  $\text{Cu} \pm \text{Au}$  suites. The results show that the barren suites are marked by substantial depletion in PGEs with low Pd/MgO ( $<\sim 0.1\times 10^9$ ) and Pd/Pt ( $<\sim 3$ ) ratios whereas the  $\text{Cu} \pm \text{Au}$  suites are systematically enriched in PGEs with high Pd/MgO ( $>\sim 0.1\times 10^9$ ) and Pd/Pt ( $\sim 0.1$ –40) ratios at late stages of magma differentiation (< 2.5 wt.% MgO). The barren magmas commonly undergo early sulfide saturation, which substantially lowers their chalcophile element fertility. In contrast, the magmas associated with porphyry  $\text{Cu} \pm \text{Au}$  deposits are characterized by late sulfide saturation, which allows Cu and Au, as well as S, to be concentrated by fractional crystallization, increasing chalcophile element fertility.

We suggest that Pd/MgO and Pd/Pt ratio can be used as chalcophile element fertility indicators, which can distinguish barren suites from the  $Cu \pm Au$  suites. The positive correlations between these indicators and the ore grades, particularly for Au, suggest that the ore grades are controlled by magmatic Cu and Au contents, in combination with a fluid composition, as suggested by previous studies. Given that Au is a critical byproduct that improves the profitability of many porphyry Cu deposits, the PGE geochemistry-based chalcophile element fertility indicators, as outlined in this paper, will help identify Au-rich exploration targets.

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