Sulfidation State of Fluids in Active and Extinct Hydrothermal Systems: Transitions from Porphyry to Epithermal Environments

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Abstract

Use of the concept of "sulfidation state", in parallel with oxidation state, in the study of ore deposits finds its beginnings with the studies of Reno Sales and Charles Meyer at Butte, Montana. Experimental determination of the stability of sulfide minerals in terms of fS_2 and temperature followed, leading to definition of contrasts in ore-forming environments. More recent studies of vapor compositions in active volcanic and geothermal systems allow direct comparisons with geochemical environments deduced from petrologic study. In this paper, we present a compilation of oxidation and sulfidation states of fresh igneous rocks from arc environments and on sulfidation states of sulfide assemblages in calc-alkalic porphyry copper, porphyry-related base-metal veins, and epithermal gold-silver deposits. These data are contrasted with compositions of fluids from active systems by plotting vapor compositions on $\log fS_2 - 1000/T$, $R_H - 1000/T$, $R_S - 1000/T$ diagrams, where $R_H \approx \log (X H_2 / X H_2O)$, $R_S \approx \log (X H_2 / X H_2S)$ and X = mole fraction of the gas.

Oxidation states of andesitic arc magmas plot in a tight cluster between fayalite + magnetite + quartz and pyrrhotite + pyrite + magnetite. On equilibrating below the solidus, arc plutons deviate toward higher oxidation states. Sulfidation states of arc magmas are very low to

low, lying between fayalite + magnetite + quartz + pyrrhotite and pyrrhotite + pyrite. A plot of R_H values versus measured temperatures for volcanic fumaroles reveals close agreement with the isomolar SO₂ = H₂S curve (sulfur-gas buffer) to temperatures below 500 °C. Giggenbach concluded from this observation that the oxidation state of the vapors is controlled by their magmatic sulfur-gas composition, a conclusion consistent with oxidation state trajectories for cooling plutons.

Reactive magmatic-hydrothermal fluids from active systems trend toward lower R_H and R_S (higher oxidation and sulfidation states) with declining temperature, achieving minima at 200 to 100°C ($R_S = -1.5$ to -3.0). Below 200 °C, R_H and R_S both increase abruptly ($R_S = 0$) through interaction with wall rock. In contrast, geothermal liquids are relatively reduced, near neutral pH, and their sulfidation state remains low to intermediate ($R_S = 0$) throughout the range 320-100°C. This may be caused by a greater degree of fluid-rock interaction at depth, a smaller magmatic component, or a distinct magmatic component. The reduced limit of geothermal compositions has an R_H value of about -3, equivalent to Giggenbach's rock buffer, where iron-bearing minerals in fresh rock establish a "floor" to the oxidation state, just as the sulfur-gas buffer acts as a "ceiling".

The majority of porphyry copper deposits contain magnetite, either without sulfides or as part of ore-grade assemblages containing bornite and/or chalcopyrite without pyrite. In some deposits, pyrite + chalcopyrite dominates the ore zone. All of these assemblages are of intermediate sulfidation state. High-temperature volcanic fumaroles plot largely in the bornite + magnetite field, consistent with the view that porphyry copper assemblages precipitate from magmatic volatiles that cooled along the sulfur-gas buffer. Base-metal veins associated with porphyry copper deposits extend this cooling trend and display a range of sulfidation states from very high in central zones (pyrite + digenite + covellite + enargite) to intermediate and low in peripheral zones or latest stages (pyrite + tennantite + chalcopyrite).

In high-sulfidation (HS) epithermal deposits the sulfidation state ranges from high for copper-rich enargite-bearing assemblages to intermediate for the later gold-rich tennantite-tetrahedrite + pyrite assemblages, with similarities to and overlap with the base-metal veins. In intermediate sulfidation (IS) epithermal deposits the full range of intermediate sulfidation states is represented by the assemblage pyrite + chalcopyrite + tetrahedrite. The general similarity of assemblages associated with gold in HS and IS deposits suggests a closer affiliation between these two types than commonly thought. Low sulfidation (LS) epithermal deposits appear to be distinct and show little variation from low and intermediate sulfidation states. Evidence for transients in sulfidation state, due to boiling, local wall rock influence, or other factors, exists in all three types of epithermal deposits.

Sulfide mineral assemblages in porphyry copper deposits, porphyry-related base-metal veins, and HS-IS epithermal deposits, when taken together, describe a cooling path toward increasing sulfidation states from $R_S = -1$ at 600°C to $R_S = -3$ at 300°C, followed by an abrupt increase to $R_S = 0$ as equilibrium with the rock buffer is achieved. This pattern, also evident in fluid compositions from active magmatic-hydrothermal systems, suggests a continuum between these deposit types. Fluid compositions in active hydrothermal systems span the complete range of chemical and physical states that are commonly relegated to changing time in intrusion-centered ore deposits.

Introduction

Analogies have long been made between active and extinct hydrothermal systems in the study of ore deposits, first in the early part of the 20th C in the context of epithermal deposits (Ransome, 1907; Lindgren, 1933). In porphyry-type deposits, analogies evolved from dominantly geological (e.g., Sillitoe, 1973), through dominantly physical (e.g., Henley and McNabb, 1978), to integrative (White, 1981; Hedenquist and Lowenstern, 1994) as the data base on geology and fluid composition in active systems expanded (e.g., Henley and Ellis, 1983; Giggenbach, 1997).

The geochemical pathways taken by hydrothermal fluids can be described in terms of many variables. One that has long spanned the disciplines of economic geology, petrology and geochemistry is oxidation state (fO_2 compared to a standard), a variable of particular importance because of its effect on the speciation of sulfur in magmas and in aqueous fluids and its role in metal transport (Barnes and Czamanske, 1967; Burnham, 1979; Burnham and Ohmoto, 1980; Candela, 1992). Oxidation state can be estimated from mineral assemblages in igneous rocks (e.g., Wones, 1981; Whitney, 1984) and ore deposits (Holland, 1959; Barnes and Kullerud, 1961; Meyer and Hemley, 1967), and calculated from direct sampling in active systems (Giggenbach, 1987). The application of sulfur fugacity (or less precisely, partial pressure) migrated into the geologic realm from metallurgy about the middle of the last century and its application to ores (e.g., Merwin and Lombard, 1937; Sales and Meyer, 1949; McKinstry, 1959) closely paralleled the use of oxygen fugacity in petrology. Sulfidation state, a function of both fS_2 and temperature (Barton and Skinner, 1967; Barton, 1970) has had limited application to the classification of ore deposits (e.g., Heald et al., 1987) and the study of continua within classes (e.g., Barton et al., 1995) despite the consistent framework it provides for systematizing sulfide mineral assemblages and studying evolution of hydrothermal systems.

In this paper, our focus is on ores related to medium-K and high-K calc-alkaline magmatic suites of both continental and oceanic arc environments. We consider deep porphyrycopper environments as transitional into shallower porphyry-related base-metal vein and replacement deposits (Meyer and Hemley, 1967; Einaudi, 1982; Hemley and Hunt, 1992) and ultimately near the surface into epithermal gold-silver deposits (Sillitoe, 1973; Sillitoe, 1983; Hedenquist et al., 1998; Muntean and Einaudi, 2001). If taken as a whole, sulfide mineral assemblages in these deposits describe a "Lazy-L" shaped field on a log $fS_2 - 1000/T$ diagram (Fig. 1). In a given district, such a field would represent the integrated evolutionary paths of multiple batches of hydrothermal fluids, likely from different sources, as they undergo cooling, depressurization, and variable degrees of interaction with their wall rocks. The environment described is one of increasing sulfidation state on cooling from near magmatic temperatures down to 300°C, at which point an abrupt decline in sulfidation state occurs. This same Lazy-L pattern was deduced 35 years ago by Meyer and Hemley (1967), yet the analogy with fluid pathways in active volcanic and geothermal systems was not appreciated at that time due to a paucity of reliable analytical data on fluids. As discussed in this paper, fluids from active systems (e.g., compilations by Giggenbach et al., 1986, Giggenbach, 1996, and references therein) mimic this L-shaped trend in both sulfidation and redox states.

We first summarize aspects of terminology and the oxidation and sulfidation state of magmas and wall rocks, then present an overview of the fluid chemistry of active hydrothermal systems and compare this to the evidence from mineral assemblages in ore deposits. We touch on processes that control the pathways of fluid evolution based on our integration of information from these active and extinct systems.

Terminology and Conventions

Mineral assemblages, associations, and zoning

Mineral assemblages commonly are defined as groupings of minerals that occur in direct contact and that do not display evidence of reaction with one another (Barton et al., 1963). The term mineral association is applied to groups of minerals that are characteristic of a given zone of alteration or portion of a vein system, but which are not all in contact and were not necessarily all deposited at the same time. Thus, pyrite, chalcopyrite, bornite, and magnetite is a common association in potassic alteration zones in porphyry copper deposits, but the assemblages are magnetite + bornite, bornite + chalcopyrite, or chalcopyrite + pyrite, but not bornite + pyrite in this environment. In the text below minerals that are part of an association are separated by commas.

Although mineral associations commonly are summarized in maps showing mineral zoning, such maps are ambiguous because they represent patterns resulting from superposition of multiple packets of fluids and migrating flow paths through time. To serve as a basis for assessment of processes (and for ore finding), such maps need to be redrawn as a series of time frames representing the zoning of mineral assemblages at given points in time. In order to draw time frames, it is necessary to map time lines through the deposit. In porphyry deposits, successive intrusions can serve as time lines to draw successive time-frames (e.g., Gustafson and Hunt, 1975; Carten, 1986; Dilles and Einaudi, 1992) or space-time diagrams (Dilles et al., 2000; Muntean and Einaudi, 2001). In vein deposits, time lines are more difficult to establish, but in some cases these can be based on the presence of unique features such as sphalerite stratigraphy (Barton et al., 1977; Hayba, 1997).

In spite of these ambiguities in the link between space and time, some common themes are evident within the deposit types considered here. For example, in the early stages of zoned base-metal veins found in some porphyry copper districts, minerals occurring in innermost zones appear to have encroached outward onto minerals that comprised the next outer zones. In some deposits (e.g., Main Stage veins at Butte, Montana: Sales and Meyer, 1949) this contemporaneous zonal growth pattern was largely frozen in (prograde pattern). In other deposits, minerals of the outer zones encroached onto the center of the deposit as isotherms collapsed, and it is this late stage that is most prominently recorded by the temporal sequence of minerals, especially sulfides (retrograde pattern). This time sequence may be seen in outcrop as late pyrite veins cutting early quartz + chalcopyrite + bornite veins in many porphyry copper deposits (e.g., El Salvador, Chile; Gustafson and Hunt, 1975) or recognized under the microscope as late tennantite replacing enargite in many base-metal veins (e.g., McKinstry, 1963) and in some high-sulfidation epithermal gold deposits (e.g., Jannas et al., 1999; Claveria, 2001). These and other issues related to zoning and paragenesis and the use of paragenetic diagrams have been reviewed by Hemley and Hunt (1992, p. 36-40). In the present paper, we summarize these spatial and temporal patterns in tables and plot trends on phase diagrams. The plotted trends are representative of the geochemical environments within cooling hydrothermal systems and do not in general represent the trajectory of a single episode of hydrothermal fluid flow. In some districts these changing environments are spatially separated (e.g., an outward zonal pattern such as A-B-C, Fig. 1), in others they are superimposed (e.g., B superimposed on A, followed by C superimposed on B, Fig. 1), but we have not attempted to make this distinction on phase diagrams.

Sulfidation state terminology

The terms "sulfidation" and "sulfidation state", and the concepts that surround them, have a long history of development in the study of ore deposits. Table 1 serves to record part of that history and to define and clarify the terminology. The references cited are among the highlights, but are not intended as a comprehensive review. The large number of terms involved, including "high-sulfur sulfides", "sulfidation", "sulfidation reactions", "sulfur fugacity", and "sulfidation state", often are used as synonyms, whereas they have different meanings (Table 1).

We use the term sulfidation state as defined by Barton (1970) and in a manner analogous to oxidation state (e.g., Wones, 1981), where the frame of reference is temperature and the fugacity of S_2 and O_2 gas, respectively. The sulfur or oxygen fugacity of a system at any temperature can be compared to standard mineral reactions (buffers), such as

$$4 \operatorname{Fe}_{3}O_{4} + O_{2} = 6 \operatorname{Fe}_{2}O_{3}$$
(1)
magnetite hematite

for oxygen, or

$$2 \text{ FeS} + S_2 = 2 \text{ FeS}_2$$
(2)
pyrrhotite pyrite

for sulfur. The difference between the oxygen or sulfur fugacity implied by a natural mineral assemblage compared with that of a buffer reaction forms the basis for assigning relative oxidation or sulfidation states.

For ease of comparing the sulfidation state between different fluids and between fluids and mineral assemblages, we introduce here an informal terminology based on sulfidation reactions among minerals in the system Cu-Fe-As-S (reactions identified in Fig. 1) common to porphyry copper deposits, porphyry-related veins, and epithermal precious-metal deposits. From low to high, these are: "very low", "low", "intermediate", "high", and "very high" sulfidation states. The boundary between intermediate and high sulfidation states is defined by the sulfidation reaction

$$5 \text{ CuFeS}_2 + \text{S}_2 = \text{Cu}_5 \text{FeS}_4 + 4 \text{ FeS}_2$$
 (3)
chalcopyrite bornite pyrite,

(Fig. 1) which approximately coincides with the sulfidation reaction

$$0.67 \operatorname{Cu}_{12}\operatorname{As}_4 \operatorname{S}_{13} + \operatorname{S}_2 = 2.67 \operatorname{Cu}_3 \operatorname{AsS}_4$$
(4)
tennantite enargite,

and represents the transition between porphyry copper deposits (*sensu stricto*) and porphyryrelated base-metal veins (see below). Further, the sulfidation reaction

$$0.47 \text{ FeAsS} + 1.41 \text{ CuFeS}_2 + S_2 = 0.12 \text{ Cu}_{12}\text{As}_4\text{S}_{13} + 1.88 \text{ FeS}_2$$
(5)
arsenopyrite chalcopyrite tennantite pyrite,

representing a lower limit to sulfidation state in IS epithermal deposits, effectively coincides with the lower limit of intermediate sulfidation states as defined by the reaction (2).

Sulfidation state is neither a function of the sulfur content of a sulfide mineral or mineral assemblage, nor a function of proximity of a given assemblage or fluid to the sulfur condensation

curve. Regarding the first point, pyrite contains more sulfur than covellite, but its stability field extends to lower sulfidation states than that of covellite (Fig. 1). Regarding the second point, the sulfur condensation curve, which acts as a ceiling to fS_2 at any given temperature, has a flatter slope on a log fS_2 - (1000/T) diagram than the sulfidation reactions used here to index sulfidation state (Fig. 1). Because of this relation, each sulfidation state has an upper thermal limit.

We caution that the term sulfidation state is not strictly parallel to the terms "high sulfidation" (HS) and "low sulfidation" (LS) used in the classification of epithermal deposits (White and Hedenquist, 1990; Hedenquist et al., 1994a; Table 1), although sulfide assemblages that characterize the major portion of many epithermal deposits (e.g., enargite in HS deposits) are mostly consistent with the classification terms. Sulfidation state, as used here in the sense of a mineral facies or in the context of mineral assemblages and fluids, can vary widely within deposits, and, as pointed out by Barton (1970), can vary within a hand sample or crystal.

Different, hypothetical cooling paths can be drawn on a log $fS_2 - 1000/T$ petrogenetic grid: gas-buffered, rock-buffered, and mineral-buffered paths. The latter two types of cooling paths are treated at length below. A mineral-buffered path is one that is parallel to mineral reaction lines – cooling takes place at constant sulfidation state, although the absolute value of fS_2 declines along such a path. As stressed by Barton (1970), hydrothermal fluids rarely follow mineral-buffered paths, the mineral buffers being more correctly regarded as indicators of the state of the system at a given point in space and time. Barton's (1970) Main Line environment (Fig. 1), representing the common range of sulfidation states in ore-forming hydrothermal systems and containing an infinite number of possible fluid trajectories, has a lower positive slope than mineral buffers on a log $fS_2 - 1000/T$ diagram. Thus, this environment is one of gradually increasing sulfidation state (although the value of fS_2 decreases) with decreasing temperature. One of the key themes of the present paper is to investigate such trajectories in natural ores and active hydrothermal systems.

Relationship between sulfidation state, oxidation state, and pH

The valence state of sulfur was the basis used by Hedenquist (1987) in the original definition of two end member "high sulfidation" and "low sulfidation" epithermal deposits (Table 1). Aqueous and gas species in ore-forming fluids contain sulfur in valence, or oxidation, states (n) that are either greater (e.g., in $SO_4^{=}$, n = +2) or less (in H₂S, n = -2) than S₂ (n = 0). With this measure of oxidation state, two oxidation states can apply to a single sulfidation state depending on whether the predominant aqueous species is sulfate or sulfide. However, if fO_2 is used as a measure of oxidation state, as in this paper, then in H₂S-dominant fluids fS_2 covaries with fO_2 :

$$2 H_2 S + O_2 = S_2 + 2 H_2 O, (6)$$

and in SO4⁼-dominant fluids fS_2 is inversely proportional to the cube root of fO_2 at constant pH:

$$2 \text{ SO}_4^{=} + 4 \text{ H}^+ = \text{S}_2 + 3 \text{ O}_2 + 2 \text{ H}_2\text{O}.$$
 (7)

In the commonly recognized isothermal $\log fS_2$ -log fO_2 and $\log fO_2 - pH$ plots, the above relations are expressed by a maximum in fS_2 that lies along the aqueous sulfate-sulfide predominance boundaries, where fS_2 increases with increasing fO_2 and decreasing pH (see figures and discussion in Barnes and Kullerud, 1961; Barnes and Czamanske, 1967). Meyer and Hemley (1967, p. 222) use equations such as (6) and (7) above to stress the "delicate relationships" between fS_2 and fO_2 and pH, in which relatively low pH correlates with relatively

high fugacities of both sulfur and oxygen (e.g., reaction 7) at high ratios of fS_2/fO_2 , a correlation amply supported by sulfide-silicate assemblages in ore deposits.

R_S and R_H as alternative expressions of sulfidation and oxidation state

Given the complex relationships just described between fS_2 , fO_2 , pH, and aqueous sulfur species, and given the low concentrations of S_2 gas below 600°C, it can be argued that analysis of geochemical systems is better conducted on the basis of actual reaction participants. Thus, Giggenbach (1987) adopted the redox parameter $R_H = \log (f H_2 / f H_2O)$, and the sulfidation state parameter $R_S = \log (f H_2 / f H_2S)$. The variables R_S and $\log fS_2$ (and R_H and $\log fO_2$) are equally valid representations of phase equilibria. We use both representations and retain fS_2 because of its historical use in the study of ore deposits and its thermodynamic importance in the definition of sulfide phase equilibria. The final story is best told with a combination of both sets of variables.

Vapor compositions from active hydrothermal systems can be plotted directly on $R_H - 1000/T$, $R_S - 1000/T$, and $\log fS_2 - 1000/T$ diagrams where $R_H \approx \log (XH_2 / XH_2O)$, $R_S \approx \log (XH_2 / XH_2S)$, and X = analytical mole fraction (Appendix). We have vertically flipped the $R_H - 1000/T$ and $R_S - 1000/T$ diagrams from Giggenbach's original orientation so that the values of R_H and R_S increase downward on the y-axis. In this view, the topology is similar to the $\log fO_2 - 1000/T$ and $\log fS_2 - 1000/T$ diagrams, respectively, typically used in petrology and ore deposit studies (Fig. 2). As discussed below, vapors from active systems define a pattern similar to the Lazy-L defined by sulfide assemblages of porphyry and epithermal affiliation (compare Figs. 1 and 2).

Magmas, Rock Buffers, and Gas Buffers

Oxidation and sulfidation states of magmas

Here we briefly summarize data on oxidation and sulfidation state of magmas because these are the starting points for consideration of magmatic volatiles. The correlation between oxidation state of magmas and the types of associated mineral deposits has been addressed in the context of porphyry- and granitoid-related deposits (Burnham and Ohmoto, 1980; Einaudi et al., 1981; Thompson and Newberry, 2000), and in the broader perspective of igneous controls on metallogenesis (Ishihara, 1981; Barton et al., 1995; Barton, 1996; Jensen and Barton, 2000; John, 2001).

The oxidation state of magmas can be estimated on the basis of mineral associations and contrasts in ferric/ferrous ratios or magnetic susceptibility. These qualitative approaches to oxidation state, useful in metallogenetic analysis (e.g., Ishihara, 1981; Thompson and Newberry, 2000), are insufficient for our purpose, which is to compare quantitative estimates with analytical data from active hydrothermal systems. For quantitative estimates of oxidation states the compositions of coexisting ilmenite and titano-magnetite are particularly useful (Buddington and Lindsley, 1964; Haggerty, 1976; Spencer and Lindsley, 1981; Hildreth, 1981), but also important are assemblages containing iron-magnesium silicates and titanite (Wones, 1981; Dilles, 1987; Andersen et al., 1991). The continuing refinement of solution models for Fe-Ti-oxides, pyroxene and olivine (Andersen et al., 1993) means that past compilations of the oxidation state of volcanic rocks (e.g., Haggerty, 1976) must be recalculated for internal consistency. Table 2 summarizes the sources of data used here to calculate the oxidation state of magmas and plutonic

rocks from arc and rift environments plotted on Figure 3. The result is not fundamentally different from the well-known tendency for arc magmas to lie at higher oxidation states than rift magmas (Hildreth, 1981; John, 2001). The former include the arc-related calc-alkaline magmas associated with copper-rich porphyry and base- and precious metal vein deposits worldwide and which are the focus of this paper. These arc magmas lie 2 R_H units above the hematite + magnetite buffer where the SO₂/H₂S fugacity ratio of magmatic-hydrothermal fluids ranges from 0.1 to 10 (Burnham and Ohmoto, 1980). Rift magmas include basalt and rhyolite of the Snake River and Yellowstone hotspots (Hildreth, 1981) and the Northern Nevada Rift (John, 2001) that straddle the quartz + magmatite + fayalite buffer (Fig. 3).

The sulfidation state of magmas is more difficult to estimate and has received less attention than oxidation state because sulfides rarely preserve their magmatic identities on cooling below solidus temperatures. Sulfidation states, mostly from andesitic arc environments, and approximated by the composition of pyrrhotite in magmatic sulfide combined with temperature estimates from Fe-Ti-oxides (Whitney and Stormer, 1983; Whitney, 1984; Drexler and Munoz, 1988), are plotted on Figures 4 and 5 (Table 2). Arc magmas lie between the fayalite + magnetite + quartz + pyrrhotite buffer and the pyrrhotite + pyrite + $s_{(liq)}$ invariant point, at very low to low sulfidation states (Fig. 4). Recent study of magmatic Cu-Fe-sulfide blebs in intermediate to felsic plutonic and volcanic rocks of the Great Basin, western USA (Keith et al., 1995; Keith et al., 1997; Borrock et al., 1999) are consistent with these relatively low sulfidation states, appropriate for pyrrhotite + intermediate solid solution (which incorporates the chalcopyrite composition at high temperatures), and bornite solid solution + magnetite.

"Rock buffer"

The "rock buffer" can be described as the sum of rock-forming minerals that interact with a hydrothermal fluid at sub-magmatic temperatures and mediate its sulfidation-oxidation state and total acidity (Barton, 1970; Hemley and Hunt, 1992). Minerals that are likely to be involved as buffers are the ferromagnesian minerals with iron in different valence states: Fe⁺² in pyroxene, olivine, biotite, and magnetite; Fe⁺³ in biotite and magnetite. Giggenbach (1987) commented on the problem of defining a rock buffer in volcanic-hosted hydrothermal systems: "In the absence of detailed information on the buffer mineralogy..., it appears unjustified to presuppose the presence of any of these minerals" (p. 148). He concluded that the method used by Fudali (1965) and Carmichael and Ghiorso (1986) for estimating oxidation state of andesitic magmas can be extended to rocks, and he used fayalite and hematite as thermodynamic proxies for the rock buffer (Appendix), labeled the "Giggenbach rock buffer" on Figure 3.

Where does the Giggenbach rock buffer lie relative to mineral buffers? At 350°C, the rock buffer lies close to magnetite + pyrrhotite + pyrite; at 500°C it coincides with both Ni + NiO and K feldspar + quartz + magnetite + phlogopite ($X_{Fe}^{bio} = 0.2$), and at 800 to 1000°C it intersects the center of the field of oxidation states of subduction-related arc magmas (Figs. 2 and 3). R_H values of neutral-pH geothermal fluids lie on or slightly below this rock buffer, as discussed below. Using the above relations, we show the approximate location of the rock buffer on the log fS_2 – 1000/T and R_S - 1000/T diagrams (Figs. 2, 4 and 5), where it lies in the lowermost part of Barton's (1970) Main Line sulfidation environment.

Just as magmas ranging from MOR basalt to continental arc dacite display oxidation states that vary over 3 units R_H (Fig. 3) and 2.5 units R_S (Fig. 5), the rock buffer in any given mineral deposit can deviate from the Giggenbach rock buffer. For example, the reaction of

peridotite to serpentinite could act to buffer hydrothermal fluids to low oxidation states, resulting in low sulfidation state assemblages in the initial stages of ore deposition, as at the Mamut porphyry copper deposit, Malaysia, where chalcopyrite ore contains pyrrhotite and magnetite (Kosaka and Wakita, 1978). Alternatively, quartzite lacks buffering capacity in terms of both fO_2 and fS_2 ; a magmatic-hydrothermal fluid traveling through quartzite could evolve along the sulfur-gas buffer (see below) toward high-sulfidation states at relatively high temperatures, as at the Bingham porphyry copper deposit, Utah (Inan and Einaudi, 2002).

Magmatic sulfur-gas buffer

The SO₂/H₂S gas ratio of volatiles exsolved from relatively oxidized, I-type magmas ranges from 0.1 to 10 (Ohmoto and Rye, 1979; Burnham and Ohmoto, 1980; Whitney and Stormer, 1983; Whitney, 1984; Giggenbach, 1987). These gases may control the oxidation/sulfidation state of hydrothermal fluids to temperatures as low as 350°C if the sulfur gases dominate over H₂ and if the fluid does not react with its wall rocks due to high flow-velocities and restriction to major fissures. On the other hand, the ratio may increase on phase separation due to loss of H₂S and H₂ to the vapor or due to sulfide deposition, or may decrease as a result of chemical exchange with wall rocks containing ferrous iron or carbonaceous matter (Ohmoto and Rye, 1979). As discussed below, R_H values of volcanic fumaroles above 500°C lie close to the isomolar sulfur-gas buffer where SO₂/H₂S = 1.0. Below 500-400°C the compositions of volcanic fumaroles and modern magmatic hydrothermal fluids are modified by partial reaction with ferrous iron in wall rocks (Giggenbach, 1987). At 250°C in neutral pH epithermal environments the SO₂/H₂S gas ratios are in the range 10⁻⁶ to 10⁻⁴ (Barton et al., 1977) and the sulfur-gas buffer no longer applies.

Oxidation and sulfidation states of fluids in cooling plutons

Volatiles that exsolve from and are in equilibrium with andesitic magmas at 800°C are at a low sulfidation state, although mostly within four log units fS_2 of sulfur saturation (Figs. 2A and 4), are relatively oxidized and straddle the sulfur-gas buffer (Fig. 3). If such fluids cool in equilibrium with their host rock (roughly the $R_H = -3$ and $R_s = 0$ isopleths, Figs. 3, 4, and 5; see below), their sulfidation state increases only gradually, crossing into intermediate sulfidation states as pyrrhotite is succeeded by pyrite at around 300°C. Such a path approximates the "geothermal" path shown in these figures. In contrast, if magmatic volatiles cool along the sulfur-gas buffer they pass into intermediate sulfidation states at much higher temperatures, around 500°C, where magmatic sulfide blebs of pyrrhotite + intermediate solid solution are converted to pyrite + chalcopyrite. At lower temperatures, near 350°C, magnetite is oxidized and sulfidized to hematite + pyrite (Figs. 4 and 5). The few published data on oxidation states of plutonic rocks that have equilibrated to temperatures as low as 400°C (Dilles, 1987; Cornejo and Mahood, 1997; Core et al., 2001) suggest that sub-solidus cooling occurred along the sulfur-gas buffer (Fig. 3), but there is an uncertainty of unknown magnitude in extrapolating the Fe-Tioxide geobarometer to such oxidized conditions at these low temperatures.

Active Hydrothermal Systems

The petrologic studies discussed above used thermodynamic calculations to determine variables such as fO_2 , fS_2 , and pH from observed mineral assemblages. In contrast, fluids collected from active hydrothermal systems allow the reactants to be measured directly, thereby eliminating the need for detailed thermodynamic calculation. For example, the ratio XH_2/XH_2O of

a fumarole sample reflects the oxidation state of the fluid, as these components are the principal redox reactants. Furthermore, the ratio XH_2/XH_2S reflects the sulfidation state and can be directly related to sulfide and oxide mineral assemblages through reactions incorporating these species rather than O_2 and S_2 .

Volcanic fumaroles – magmatic discharges and the S-gas buffer

On exsolution from a magma, an aqueous fluid may consist of a single phase of intermediate salinity fluid, or, if the fluid has intersected its solvus, it will comprise a low-salinity vapor and a dense, hypersaline liquid (Henley and McNabb, 1978; Shinohara, 1994). The vapor, being more buoyant, separates and rises to the surface from its high-pressure source, eventually discharging as a fumarole if it is not condensed by absorption into ground water or by cooling at low water:rock ratios. Such expansion from high to low pressure results in adiabatic cooling, typically of the order of 100°C (Giggenbach, 1987). A plot of log (XH₂/XH₂O), or R_H, values (Table 3) versus measured temperatures (Fig. 3, black stars) reveals that volcanic fumaroles plot close to the sulfur-gas buffer, approximately one unit R_H above arc magmas and plutons at ~500 to >1000°C. The volcanic fumarole data, plotted at measured surface temperatures, could be corrected to their original higher temperatures, where they would lie on sulfur-gas buffer curves at higher pressures and overlap more closely the field of arc magmas.

The close relationship between XH_2/XH_2O ratios in volcanic fumaroles and the sulfur-gas buffer (Fig. 3) extends to temperatures below 500 °C, an observation indicating that the SO₂ and H₂S concentrations control the oxidation state of the vapors as they cool (Ohmoto and Rye, 1979; Burnham and Ohmoto, 1980; Giggenbach, 1987). Magmatic vapors do not plot at more oxidized values than the sulfur-gas buffer, suggesting that this is a significant limiting condition, at least at high temperatures. The resulting R_S values decline from 1.0 to -1.5 over the sampled temperature range of 1000 to 400°C, consistent with very low to low sulfidation states within the pyrrhotite stability field, as defined in the previous section.

Magmatic-hydrothermal and geothermal systems - departure from S-gas buffer

We consider two end-member environments of intrusion-driven, active hydrothermal systems below 500°C: magmatic hydrothermal and geothermal (Table 3). Surficial vapors discharged from boiling magmatic-hydrothermal systems range up to 400°C, whereas geothermal fluids at surface are largely <170°C. Magmatic-hydrothermal systems consist of highly reactive liquids whereas geothermal systems are near neutral pH (Giggenbach, 1984, 1997). The difference in reactivity between the two systems is related to the contribution of condensed magmatic volatiles (SO₂, HCl, HF) on one hand, and the degree of wall-rock interaction on the other (Giggenbach, 1992; Hedenquist, 1995). Despite these end-member groupings, there is a spectrum of physical and chemical characteristics that span the range from volcanic fumaroles to vapors discharged from magmatic-hydrothermal systems and in some magmatic arcs there may be a genetic relationship between the magmatic-hydrothermal systems (e.g., a vapor-cored geothermal system; Reyes et al., 1993). Such a genetic affiliation may also apply to the ancient analogues of these end members, the high-sulfidation and intermediate-sulfidation epithermal deposits (see below).

Magmatic-hydrothermal fluids are associated with many active volcanoes and discharge vapors at temperatures from 400 to 80°C (Table 3) near summit craters, adjacent to high-temperature volcanic fumaroles, or on the flank of the volcano. Chemical and isotopic evidence indicate that such fluids originate from condensation of magmatic vapors into meteoric water

(Giggenbach, 1987; Giggenbach et al., 1990; Taran et al., 1992, 1998; Hedenquist et al., 1994b). This process leads to the generation of a reactive liquid that, if sufficiently heated by magmatic vapor, will begin to boil at shallow depth, generating the vapors that were sampled (Table 3). In addition to vapor discharges, magmatic-hydrothermal liquids may reach the surface as hot springs with a pH as low as 1.0 (Hedenquist, 1995). The compositions of some magmatic-hydrothermal vapors lie close to the sulfur-gas buffer at temperatures >300°C, but most compositions are more reduced, although the rock buffer is approached only at temperatures near 100°C (Fig. 3, gray stars). The sulfidation state of these magmatic-hydrothermal systems (Fig. 4 and 5) shows an overall trend from intermediate at 400°C to high at 150°C, but in detail a wide range is observed, from low to high, even very high. This may be intuitively predicted for such reactive fluids because of the likelihood of large variations of water:rock ratio, in some cases on a very local scale. Giggenbach (1992, p. 1936) likens such transient behavior to the movement of a tennis ball between the two players, or major buffers, with the fluid ball getting caught occasionally by the intervening network of mineral buffers.

What happens to the SO₂ and H₂S as the magmatic vapors cool and condense? In some cases we see that SO₂ remains in solution as H₂SO₃ at temperatures <130°C (Southern Bottom field at Mutnovsky, Kamchatka, Table 3). Although thermodynamic calculations are hindered by the nature of the acidic brine at Mutnovsky, Taran et al. (1992) argue that the measurements indicate SO₂ and H₂S remain in equilibrium in solution at temperatures as low as 200°C. The sulfur-gas buffer for aqueous solution is similar to that of vapor, but is shifted due to differences in liquid-vapor distribution coefficients of the two gas species. The dominant process on cooling to temperatures below 350 to 300°C is disproportionation of SO₂ according to the reaction

$$4 SO_{2(g)} + 4 H_2O = 4 H_2SO_{3(aq)} = H_2S_{(aq)} + 3 H_2SO_{4(aq)}$$
(8)

followed by dissociation of H_2SO_4 which yields H^+ ion and ionized aqueous-sulfate species that depend on pH (Fig. 3). Thus, at pH 3 appropriate for reactive magmatic-hydrothermal fluids, the relevant reaction is

$$4 H_2 SO_{3(aq)} = H_2 S_{(aq)} + 3 H_2 SO_{4(aq)} = H_2 S_{(aq)} + 3 H^+ + 3HSO_4^-,$$
(9)

and at pH 6 appropriate for neutral-pH geothermal fluids the reaction is

$$4 H_2 SO_{3(aq)} = H_2 S_{(aq)} + 3 H_2 SO_{4(aq)} = H_2 S_{(aq)} + 6 H^+ + 3 SO_4^{=}.$$
 (10)

Dissociation of H_2SO_4 allows the buffering capacity of oxidized sulfur species to be consumed by interaction with wall rock, leading to the departure of fluid compositions from the sulfur-gas buffer at these lower temperatures. All of the data for the magmatic hydrothermal systems compiled here plot in the $H_2S_{(aq)}$ -predominance field at a pH of 3 (Fig. 3).

Geothermal liquids, in contrast to reactive magmatic-hydrothermal fluids, are closer to equilibrium with their host rock (e.g., Giggenbach, 1981, 1984). These near-neutral pH fluids are relatively reduced (Fig. 3, white stars) and their sulfidation state is typically low to intermediate, near $R_S = 0$ (Figs. 4 and 5) throughout the temperature range 300-100°C. All of the data for the neutral-pH systems plot well into the $H_2S_{(aq)}$ -predominance field at a pH of 6 (Fig. 3). The lower (reduced) limit of geothermal compositions has an R_H value of about –3, equivalent to Giggenbach's (1987) rock buffer, which appears to establish a redox "floor", just as the sulfurgas buffer acts as a redox "ceiling".

These apparent limiting buffer reactions lead to several questions that are relevant to understanding the evolution of ore fluids. If a fluid originates from magma at high temperature in

a state of equilibrium with the sulfur-gas buffer, what controls its evolutionary path as it cools? What is the relative importance of fluid buffers versus mineral and rock buffers, and what is the relative effect of processes such as liquid- and vapor-phase separation? Under what situations is one control more influential than another, and can the relative importance of such controls change in different positions within a hydrothermal system, and/or during its life history?

In order to consider these questions further, we first summarize the typical hydrothermal mineral assemblages of porphyry and epithermal ore deposits (Table 4¹). In porphyry deposits, hydrothermal assemblages span a wide thermal range, from near-magmatic temperatures to <300°C, similar to the temperature range of volcanic fumaroles. In contrast, epithermal deposits form at temperatures <300°C to as low as 100°C, similar to the lower-temperature end of magmatic-hydrothermal and geothermal systems. We then integrate information on patterns of mineral assemblages in ore deposits with the data from active hydrothermal systems to constrain the space-time evolution and processes of fluid-mineral interaction in ore deposits.

Sulfide Assemblages in Porphyry and Epithermal Deposits

Porphyry copper deposits

The review that follows is limited to porphyry copper deposits associated with medium-K and high-K calc-alkaline magmatic suites in subduction-related continental and oceanic arc environments. We have not included the so-called reduced porphyry Cu-Au deposits of Rowins (2000) because most lack the key features of porphyry-type deposits and many can be classed with gold deposits related to reduced granites (e.g., Thompson and Newberry, 2000). We make brief reference in the next section to porphyry and epithermal ores related to other magmatic suites to point out contrasts with those discussed here.

Data on mineral assemblages and associations, compiled from 30 calc-alkaline porphyry copper deposits, are generalized in Table 4. Note that this table does not achieve the ideals set out above for time-space definition of assemblages. Rather, it lists assemblages as a function of distance from the center of the deposits and generalizes that distal assemblages tend to migrate inward with time and overprint proximal assemblages. Informal "types" (Table 4), likely representing a continuum, have been identified on the basis of the opaque mineral assemblage that dominates the ore zone. Use of assemblages representing the bulk of the ore-grade portions of the deposits focuses the discussion on the dominant ore-depositional environment, but questions remain regarding the possibility that in some deposits erosion has removed high-grade ores of different character. In this context, depth of drilling is unlikely to be an issue, because the majority of these deposits are relatively well-explored by deep drilling.

In Type-1 porphyries most of the high-grade copper ore consists of the assemblage magnetite + bornite + digenite with biotite \pm K feldspar (e.g., Batu Hijau, Indonesia; Yerington Mine, Nevada). At the other end of the spectrum, in Type-4 porphyries most of the high-grade ore consists of the assemblage chalcopyrite + pyrite with either biotite + K feldspar or sericite \pm chlorite (e.g., Sungun, Iran; Sierrita-Esperanza, Arizona). The majority of the 30 deposits summarized contain magnetite, either as an early phase with or without minor bornite and/or chalcopyrite (19 deposits) or as part of pyrite-absent assemblages containing digenite, bornite,

¹ We have not included references in the text for those deposits listed in Table 4 unless we use additional sources for a key observation; sources for deposits not listed in Table 4 are referenced in the text.

and/or chalcopyrite that constitute ore grade (9 deposits). In 19 of the deposits the highest hypogene grades occur in rocks containing the assemblages bornite + magnetite, chalcopyrite + magnetite, or bornite + chalcopyrite, all without pyrite. These deposits are represented by Meyer and Hemley's (1967) "assemblage IV" (Fig. 6). Finally, 12 of the deposits (approximately 30%) contain veins with high and very-high sulfidation state assemblages, and in a small number of these deposits, the high-sulfidation veins represent a significant if not dominant part of the ore (e.g., Collahuasi and Chuquicamata, Chile; Butte, Montana).

In order to plot an average evolutionary path that integrates all packets of fluid for porphyry-copper deposits on diagrams such as Figure 1 we need constraints on temperature as well as sulfidation state. The evidence comes largely from the study of fluid inclusions, but linking a particular fluid inclusion assemblage to the ore-depositional event remains problematic. The bulk of the evidence points to temperatures ranging from 700 to 350°C for K feldspar + biotite assemblages and from 350 to 200°C for chlorite + sericite and quartz + sericite + pyrite (Eastoe, 1978; Preece and Beane, 1982; Reynolds and Beane, 1985; Dilles and Einaudi, 1992; Imai, 2001, Ulrich and Heinrich, 2001). Some recent studies (e.g., Bingham, Utah, Redmond et al., 2001; Bajo de la Alumbrera, Argentina, Ulrich and Heinrich, 2001) show that K feldspar + quartz + biotite ± magnetite formed at 700-400°C, but suggest that most of the chalcopyrite and bornite began to precipitate in equilibrium with this assemblage at 400 to 350°C. Cooler fluids were responsible for pyrite + chalcopyrite ore associated with chlorite + sericite at temperatures of 350°C at Sierrita-Esperanza, 350-275°C at Santa Rita, New Mexico and at Far Southeast, Philippines (Hedenquist et al., 1998), and 310-270°C for sericitic assemblages at Panguna, Papua New Guinea. Based on the above assemblages and temperatures, the environment of porphyry copper ores spans the range from 500°C at low sulfidation states to 270°C at intermediate sulfidation states. The main-line environment for porphyry copper deposits does not cross the reaction chalcopyrite + S_2 = pyrite + bornite (bold arrow, Figs. 1 and 8).

Based on the data presented above, ore deposition at the relatively high temperatures of potassic alteration in porphyry copper deposits took place over a range of sulfidation states from low to intermediate. The ore assemblages of Type-1 and Type-4 porphyries, magnetite + bornite and chalcopyrite + pyrite, respectively, partly overlap in sulfidation and oxidation states, but only magnetite + bornite is stable at low sulfidation states, and chalcopyrite + pyrite is stable into the higher range of intermediate sulfidation states (Figs. 6 and 7). Given the tendency for oxidation and sulfidation state to increase with decreasing temperature, these differences in ore assemblages could reflect higher temperatures of initial copper deposition in Type-1 porphyries than in Type-4 porphyries, or sulfidation of low-grade magnetite + bornite mineralization to chalcopyrite + pyrite ore in Type-4 porphyries. Additionally, differences in the composition of associated igneous rocks (granodiorite-tonalite in Type-1, quartz monzonite-granite in Type-4) could play a role, as shown to be the case by Barton et al. (1995) for porphyry copper deposits in Mexico.

A contrast – very low sulfidation states of some ores related to alkaline magmatism

Sulfide mineral assemblages in many porphyry molybdenum and tungsten deposits associated with highly evolved granites and alkaline magmatic suites (Carten et al., 1993; Jensen and Barton, 2000) are of lower sulfidation state than those associated with the subalkaline porphyry Cu-Au-Mo deposits treated here. Among the better documented molybdenum and tungsten deposits in terms of sulfide assemblages are the Mount Pleasant porphyry W-Mo-(Sn) and Burnthill W deposits, New Brunswick, and the Henderson porphyry Mo deposit, Colorado, all belonging to the high-silica rhyolite-alkalic class of Carten et al. (1993).

At Mount Pleasant the assemblages native bismuth + arsenopyrite \pm bismuthinite and loellingite + arsenopyrite (Kooiman et al., 1986) formed at temperatures above 400°C (Samson, 1990). Lower temperature polymetallic veins at Mount Pleasant contain tin-bearing sulfides, sphalerite, galena, and chalcopyrite, lesser arsenopyrite and loellingite, and minor bornite, tennantite, and pyrite (Kooiman et al., 1986). Note that the assemblage bornite + pyrite is not implied. Burnthill contains native bismuth, loellingite, and pyrrhotite (MacLelland and Taylor, 1989). These New Brunswick deposits, therefore, show a cooling trend from very low sulfidation states, whose upper limit is defined by the reaction loellingite + pyrrhotite + S₂ = arsenopyrite, through low sulfidation states (defined by pyrrhotite + native bismuth), to intermediate sulfidation states at lower temperatures implied by pyrite, tennantite, and chalcopyrite (Fig. 8, arrow labeled "Mt Pleasant").

At Henderson, pyrrhotite, indicative of low sulfidation states, is a minor mineral on the flanks of the Seriate intrusion (Seedorff, 1988), where it formed around 350-400°C (Fig. 8; arrow labeled "H") and was overprinted by pyrite at 300-350°C (E. Seedorff, unpublished data, 1987). More widespread and characteristic of Henderson is magnetite ± pyrite, possibly overlapping the porphyry copper deposits in terms of sulfidation state. Thus, Henderson lies at sulfidation states that are intermediate between the relatively lower sulfidation states of the New Brunswick tungsten deposits and the relatively higher sulfidation states of calc-alkalic porphyry-copper deposits. Parallel contrasts are evident in comparing epithermal deposits associated with alkalic igneous rocks with those associated with calc-alkalic igneous rocks. Although alkaline magmas and their hydrothermal fluids are relatively oxidized, sulfide assemblages in related epithermal ores are mostly of intermediate to low sulfidation state (Richards, 1995; Jensen and Barton, 2000; Sillitoe and Hedenquist, this volume).

Porphyry-copper related base-metal vein deposits

Associated with some calc-alkaline porphyry-copper plutons are base-metal deposits in which copper ores of very high- and high sulfidation state occur in veins, massive replacements, and sulfide-cemented breccia bodies. Zoning is generally well-developed, from central enargitebearing copper ores associated with advanced argillic and/or sericitic alteration to outer tennantite-bearing (Cu)-Pb-Zn ores associated with intermediate argillic alteration (Petersen, 1965; Meyer and Hemley, 1967; Einaudi, 1982; Bartos, 1989; Hemley and Hunt, 1992). Many are superimposed on coeval porphyry-copper deposits and constitute the majority of copper ore (e.g., Main Stage veins at Butte; Rosario and La Grande veins at Collahuasi, Chile; Main Stage veins at Chuquicamata, Chile; see Table 4). Similar copper veins may be spatially separated from their porphyry roots, such as the Magma vein (Hammer and Petersen, 1968) located 2 km north of the recently discovered Magma porphyry copper deposit, Arizona (Manske and Paul, 2002), or may have no known link to mineralized porphyries, such as the copper veins in the volcanic diatreme at Cerro de Pasco, Peru (Graton and Bowditch, 1936; Petersen, 1965; Einaudi, 1977). Sharing many of the features of porphyry-related base-metal veins are the HS epithermal Au-(Cu) deposits with enargite ore in vuggy quartz associated with advanced-argillic alteration (see below).

A common feature of porphyry-related veins is an early stage of sulfidation and silicification of wall rocks accompanied by abundant pyrite as vein fill. This early pyrite-quartz stage is followed by the deposition of Cu-Fe-As-sulfides within veins (e.g., Butte and many others) or in pyrite-matrix intrusive breccias (e.g., Bisbee, Arizona; Bryant, 1968), with or without additional pyrite. With time at a given point, the ore sulfides become increasingly

copper-rich, in the sequence chalcopyrite, bornite, digenite, and the transition in time and space from tennantite to enargite occurs at or near the transition from pyrite + chalcopyrite to pyrite + bornite (e.g., Butte; El Salvador, Chile; Magma vein, Table 4). This latter feature is so common as to constitute evidence that the thermodynamic approximation for reaction 4, tennantite + S_2 = enargite (Craig and Barton, 1973), is one or two log units too low in fS_2 , or that the presence of Sb, Fe, and Zn in natural ores stabilizes tennantite relative to enargite. Therefore, we empirically adjusted reaction 4 upward on all figures.

Butte, Montana (Table 4), remains the touchstone against which all porphyry-related veins are compared, in part because this deposit is the giant of its class, but more importantly because of the detailed descriptions of the rocks and ores and the insightful interpretations by Reno Sales and Chuck Meyer. The assemblage pyrite + digenite + covellite is present only in a restricted area of the Central zone that is generally characterized by pyrite, chalcocite, and enargite, with chalcocite the most abundant copper mineral. The Central zone is surrounded by a Transition zone containing pyrite, bornite, chalcopyrite, and tennantite with residual sphalerite, and a Peripheral zone containing Zn, Pb, and Ag sulfides and Mn carbonate. The broad patterns of relative age of sulfides indicate that inner zones encroached on outer zones, as first suggested on the basis of alteration (Sales and Meyer, 1949). For example, residual sphalerite is common in tennantite + chalcopyrite ores of the Transition zone but is mostly replaced in the Central zone. However, there are late reversals of the overall temporal sequence, interpreted by Sales and Meyer (1949) to have taken place on cooling, such as bornite replacing chalcocite and tennantite replacing enargite throughout the Intermediate zone (Meyer et al. 1968), a paragenetic sequence noted in many similar deposits by McKinstry (1963).

Many porphyry-related veins exhibit patterns similar to those at Butte. The Main stage veins at Chuquicamata and the Rosario vein, Collahuasi district, Chile are superimposed on porphyry-style assemblages of intermediate sulfidation state (Table 4). Enargite appears to join the Cu-Fe-sulfides late, and at Chuquicamata sphalerite occupies a high and/or fringe position relative to enargite, analogous to Butte, El Salvador, and Magma. The youngest veins at Chuquicamata and Rosario consist of the very-high sulfidation state assemblage covellite + digenite ± pyrite. Thus, these deposits display the same increasing sulfidation state with time seen at Butte, although zoning relations are less clear at Chuquicamata. A late reversal to lower sulfidation state is recognized at Rosario, where late chalcopyrite replaced chalcocite and digenite, and tennantite replaced enargite (Dick et al., 1994).

Chalcocite appears in some deposits as an important ore mineral in areas where wall rocks were converted to quartz \pm pyrophyllite \pm dickite, as in the Central zone at Butte and in high-level veins of the Magma porphyry deposit (Manske and Paul, 2002). Chalcocite represents a significant departure from the typically very high- and high-sulfidation states of such areas, being stable only at the transition from intermediate to low sulfidation states at temperatures near 300°C (Figs. 4 and 8). Wide fluctuations in sulfidation states are indicated, and this "anomalous" chalcocite may represent a stage of ore deposition where much of the sulfur and H⁺ ion had been consumed and the fluid had returned to its wall-rock buffer due to reaction at depth.

Epithermal gold-silver deposits

Epithermal precious-metal deposits have long been distinguished into two end-members (Table 1), with a distinct difference noted in the associated alteration halo. In high-sulfidation (HS) deposits, the host rock is a leached residue of quartz, commonly vuggy in texture, with a halo of quartz + alunite ± pyrophyllite ± dickite, indicating relatively low pH fluids. In "low-

sulfidation" ("LS") deposits, halos of illite (or chlorite) \pm adularia grade out to propylitic alteration, indicating that the pH of the fluid was near neutral. Revision of this classification of epithermal deposits into high-sulfidation, intermediate-sulfidation (IS, formerly "LS") and lowsulfidation (LS) deposits (Table 4) has been suggested recently on the basis of further examination of sulfide assemblages and magmatic-tectonic affiliations (John et al., 1999; Hedenquist et al., 2000; John, 2001; Sillitoe and Hedenquist, this volume). Here we summarize the space-time distribution of sulfide assemblages in these deposits, focusing on variability in sulfidation states within single deposits, among deposits of the same class, and between classes.

In epithermal deposits many of the sulfidation reactions already referred to apply, but the common occurrence of sphalerite in these deposits provides an additional useful measure of sulfidation state. The mole percent FeS in sphalerite coexisting with pyrite or pyrrhotite is continuously variable as a function of sulfidation state (Scott and Barnes, 1971; Czamanske, 1974): 40 to 20 mole percent FeS at low sulfidation states, 20 to 1 at intermediate sulfidation states, 1.0 to 0.05 at high, and <0.05 at very high (Fig. 5). Unfortunately, there are few studies that document sphalerite compositions in any detail and address the question of equilibrium with iron sulfides.

Residual, vuggy quartz, a common but not determining characteristic of HS epithermal deposits, typically contains disseminated pyrite and SiO_2 concentrations >95 wt percent. It forms bodies that flare out upwards and/or preferentially replace a lithologic unit. In many cases, residual quartz lacks base and precious metals and constitutes a barren lithocap of advanced argillic alteration (Sillitoe, 1995). In some cases, barren bodies of advanced argillic alteration overlie porphyry copper deposits, as in the Yerington district, Nevada (Lipske and Dilles, 2000) and El Salvador (Gustafson and Hunt, 1975; Watanabe and Hedenquist, 2001). In other cases, after the leaching stage, copper and gold were introduced to form epithermal Au-(Cu) deposits with abundant sulfides (e.g., Summitville, Colorado, and many others).

There are two typical sequences of mineral deposition in HS epithermal deposits. In one, pyrite + enargite \pm luzonite \pm famatinite is early, followed by chalcopyrite + pyrite \pm tennantite \pm sphalerite \pm galena. Also post-dating the enargite assemblage is the gold stage, consisting of electrum and gold tellurides, as at Lepanto (Claveria, 2001) as well as Goldfield, Nevada; El Indio, Chile; and Kochbulak, Uzbekistan (Kovalenker et al., 1997). However, at Goldfield electrum appears to predate tennantite, whereas at Lepanto and El Indio electrum post-dates tennantite. In silver-rich deposits of this type, Ag sulfosalts are typically associated with the tennantite ± tetrahedrite zone. At Julcani, Perú, this zone is distal from the central pyrite + enargite assemblage (Deen et al., 1994). In the second sequence, as exemplified by Summitville and La Mexicana, Argentina, there is a transition toward the surface from pyrite + tennantite + chalcopyrite + sphalerite to pyrite + enargite \pm luzonite \pm sphalerite \pm chalcopyrite. At Lahóca, Hungary, tennantite is replaced by enargite, and gold is associated with pyrite relatively late in the sequence, with or after enargite. In both sequences, barite is typically very late and shallow, and can be associated with gold, as at Tambo, Chile (Jannas et al., 1999). Hypogene covellite is rare in these deposits and where it occurs it commonly is the latest sulfide in vugs prior to filling by native sulfur, as at Nansatsu, Japan (Hedenquist et al., 1994a). Thus, covellite in HS epithermal deposits occupies the same temporal position as it does in many porphyry-related base-metal vein deposits. There is a wide range of FeS content in sphalerite in HS epithermal deposits (e.g., 0.02 to 7 mole percent at La Mexicana) suggesting a wide range of sulfidation states from intermediate to very high (Fig. 5).

Based on the above summary, the sulfidation state in HS epithermal deposits ranged from high for the enargite assemblage to intermediate for the tennantite-tetrahedrite + pyrite \pm chalcopyrite assemblage (Fig. 4). At Goldfield, Lepanto and Julcani, high sulfidation states evolved to intermediate sulfidation states with time and/or distance from source. At Summitville, La Mexicana, and Lahóca, the opposite was the case: intermediate sulfidation states evolved to high sulfidation states at more shallow depths or with time. In some deposits, very high sulfidation states were achieved at very shallow levels, but the timing relative to other assemblages is not clear.

IS epithermal deposits that are also sulfide-rich share many of the sulfide assemblages of HS deposits, except that the enargite-bearing assemblage is lacking (Table 4) and the Ag:Au ratios are higher, at least 10:1, and typically >100:1. The total sulfide content can be highly variable, from 1 to > 10 percent (Albinson et al., 2001). These features are characteristic of the base-metal + silver veins of Mexico (Pachuca, Fresnillo), USA (Comstock Lode, Nevada; Creede, Colorado), Perú, Romania and elsewhere. The veins have a halo of illite ± adularia which grades downward to sericitic and outward to propylitic assemblages. The major sulfide assemblage can be relatively simple, including combinations of sphalerite, galena, pyrite, chalcopyrite, and tetrahedrite. Silver is present as Ag sulfosalts, and in some cases a large variety of these minerals occur in trace quantities, either relatively late in the sequence (Pachuca) or down the flow path (Creede). Sphalerite compositions range from <1 to 10 mole percent FeS, locally up to 20 mole percent FeS (e.g., Creede; Barton et al., 1977), covering the full range of intermediate sulfidation states (Fig. 5) but consistent with the assemblage pyrite + chalcopyrite + tetrahedrite/tennantite (Fig. 4). Such fluctuations in the FeS content of sphalerite may be caused by intermittent events such as boiling, local wall rock interaction, or by pulses of reduced fluids of magmatic or sedimentary origin (Barton et al., 1977; Bethke and Rye, 1979).

A third class of epithermal deposit, the low sulfidation (LS) deposits, are found in association with both subalkalic (including bi-modal thoeleiitic + calc-alkalic) and alkalic igneous rocks. Here we focus on deposits related to subalkaline magmatic provinces (Table 4) rather than alkaline (Jensen and Barton, 2000; Sillitoe and Hedenquist, this volume). Subalkalic LS epithermal deposits are sulfide-poor, dominated by gold typically of bonanza grades, and can be distinguished from HS and IS deposits on the basis of the ore-mineral assemblage (John et al., 1999; Hedenquist et al., 2000; John, 2001). Alteration halos are narrow and consist of illite or chlorite, with chlorite dominant in more mafic host rocks. These deposits appear to form at relatively low temperature (< 220°C) and at shallow depths (< 250 m), in places immediately beneath hot spring sinters, as at McLaughlin, California (Sherlock et al., 1995). The very low sulfide content, <1 volume percent, is dominated by pyrite (in some cases with arsenian rims), although marcasite also is common as a result of low temperatures. Gold as electrum, in places dendritic (Saunders, 1994), is closely associated with naumannite (Ag₂Se) (e.g., ginguro or black ore at Hishikari and other bonanza deposits in Japan) or pyrite (Sleeper and Midas, Nevada) and is typically present in bands of botryoidal quartz or chalcedony. Other sulfides are present only in trace amounts, and include sphalerite, chalcopyrite, galena, Ag sulfosalts, and minor occurrences of arsenopyrite and rare pyrrhotite as at Esquel, Argentina; Mule Canyon, Nevada; Rio Blanco, Ecuador; and El Limon, Nicaragua (Sillitoe, 2002; John et al., 1999; J. Sutcliffe, pers. commun., 2002; M. Gareau, pers. commun., 2002). Within the gold ore-zone the total basemetal content is typically less than a few 100s ppm. Analyses of sphalerite from two Nevada deposits indicate a range from 1 to 15 mole percent FeS (John, 2001). Mercury, As and Sb can be highly anomalous due to the low temperature and shallow depth of formation, and orpiment,

realgar and stibnite are common minerals, typically late. At 200°C, orpiment is stable at high sulfidation states and realgar at intermediate sulfidation states (Fig. 4), higher by several log units fS_2 than the low sulfidation states implied by arsenopyrite and pyrrhotite.

Discussion and Conclusions

Sulfidation state, referenced to stability fields of sulfide mineral assemblages as a function of $\log fS_2$ and temperature, has been used to define Main Line ore-forming environments and to identify evolutionary trends in individual deposits, supported by the many examples cited here. However, this indicator, with rare exceptions (e.g., Barton et al., 1995), has seen little use in comparisons between or within deposit types. Here we suggest that sulfidation state (and its companion, oxidation state), in combination with R_S and R_H , can offer significant insight into the commonality of fluid evolution between active and extinct hydrothermal systems, as well as underscore distinctions and similarities between different classes of intrusion-centered deposits.

Looping path of porphyry-copper and related base-metal vein deposits

At the deeper and hotter levels of the porphyry environment, mineral assemblages are consistent with the fluid closely following the sulfur-gas buffer during cooling to temperatures of 500 to 400 °C, with this gas-buffered path causing pronounced increases in sulfidation and oxidation states. The high-temperature volcanic fumaroles, our proxies for the higher-temperature portion of the porphyry environment, plot largely in the bornite + magnetite and chalcopyrite + magnetite fields (Fig. 7), consistent with the view that assemblages in porphyry copper deposits are derived through exsolution and cooling of magmatic volatiles. The general environment for porphyry copper deposits (gray arrow, Fig. 8) is variably modified by excursions that result from such opposite extremes as mixing with magmatic SO₂-bearing gases in major fissures (white arrow "O", Fig. 8) or approaching the rock buffer in distal fractures (white arrow "N", Fig. 8).

Sulfide assemblages in central portions of porphyry-related vein deposits represent the peak in the trend toward increasing sulfidation state that we see in the porphyries. Mineral reaction curves sweep to higher $X H_2 / X H_2 S (R_s)$ values with declining temperature as the fluid path sweeps to lower R_s (Figs. 5 and 8), resulting in marked zonal and/or temporal changes in Cu-Fe-As-S mineral assemblages that commonly are difficult to interpret. Hydrothermal fluids in these deposits achieve their lowest value of $R_S = -3$ (highest sulfidation state, approximately equal to digenite $+S_2 = \text{covellite}$) at temperatures near 300°C (arrow labeled "T" on Fig. 8). This peak of reactivity results from cooling-induced generation of H⁺ (reactions 9 or 10) outrunning the consumption of H⁺ through hydrolysis reactions with fresh wall-rocks (Meyer and Hemley, 1967). Reactivity is enhanced by boiling (Reed, 1997) and/or by addition of new SO₂ from a degassing magma chamber at depth (Brimhall, 1979, 1980; arrow labeled "SO₂ flux", Fig. 8). The path will be accentuated further if fluids are isolated from their rock buffer, a more likely case in vein deposits than in fracture-controlled porphyry-type deposits. Thus, Sales and Meyer (1949) concluded that high sulfidation states were achieved in the Central Zone at Butte because fluids were restricted to fissures with sericite-pyrite walls. The same conclusion can be applied to the early pyrite + enargite assemblage typical of residual quartz in HS epithermal deposits. In the absence of wall rocks leached during earlier hydrothermal activity, acidic ore fluids characteristic of very high sulfidation states are unlikely to develop.

From this peak in reactivity, the sulfidation state eventually declines as equilibrium with fresh wall rock is achieved and as sulfur is lost from solution. The "isothermal looping path" shown in Figure 6 "can therefore also be construed as the trace of a more gently curved path of decreasing hydrothermal temperature" (Meyer and Hemley, 1967, p. 232), the Lazy-L pattern shown in Figures 1, 2, and 8.

Looping path of active magmatic hydrothermal systems

The composite pattern of all volcanic fumaroles and active magmatic-hydrothermal samples (Figs. 3, 4, and 5) mimics the looping path defined by porphyry-copper related ores, although it is shifted to lower temperatures. In the active systems the trend toward higher sulfidation and oxidation states on cooling is reversed at around 100-130°C. The clearest example of this reversal in R_H and R_S comes from White Island (Figs. 4 and 5), where five samples were collected from 760 to 100°C fumaroles on a north-south traverse on the same day in February, 1984 (Giggenbach, 1987). The 100°C sample has the lowest R_{SH} (X SO₂ / X H₂S) value (Table 3), indicating that SO₂ has to a large extent been reduced to H₂S through wall-rock interaction, causing an increase in R_H relative to other low-temperature samples. A similar pattern is suggested by the data from Vulcano (Fig. 5), although samples at temperatures between 300 and 100 °C are absent. The minimum R_H and R_S values for active fluids are near 115°C whereas in porphyry-copper related deposits these minima are near 300°C. This difference is an artifact of surface sampling because a vapor separated from a 250-300°C liquid at depth will reach the surface at a maximum temperature of 171°C, a function of the enthalpy of the liquid (White et al., 1971). Most of the vapors generated from such fluids cool even more and reach the surface at <130°C. Therefore, a fundamental difference in actual reversal temperatures between modern and ancient systems is unlikely.

HS deposits – variations on the theme of porphyry-related base-metal veins

HS epithermal Au-(Cu) deposits with enargite in residual quartz (e.g., Lepanto, La Mexicana, Lahóca; Table 4) have many similarities to porphyry-related base-metal deposits. These similarities extend beyond their common association with porphyry copper deposits, and include structural style, ore textures, wall-rock alteration, and sulfidation states reflected in their mineral assemblages. Many HS deposits hosted by residual quartz show a shift to lower sulfidation states with time (Fig. 8), with enargite replaced by tennantite at the time of gold deposition. In common with porphyry-related base-metal veins, this decline in sulfidation state is likely because the enargite stage is far from equilibrium with wall rock outside the silicic host, and the natural long-term tendency is for hydrothermal fluids to become reduced.

Differences between these two classes of deposits also are evident. Porphyry-related base-metal veins typically are well zoned, whereas HS deposits are highly telescoped, suggesting that the latter deposits formed at shallower depths in environments with steep thermal gradients. Base-metal veins have a more complex mineralogy and metal suite that includes copper, zinc, lead, and silver, and they tend to be associated with molybdenum-bearing, gold-poor porphyries. In contrast, many HS epithermal Au-(Cu) deposits are associated with gold-rich porphyry copper districts with minor molybdenum and other metals (cf., Barton et al., 1995).

HS epithermal deposits – affiliation with IS deposits

The gold-bearing sulfide assemblages of the HS epithermal deposits are similar to those of IS deposits. IS veins are common on the margins of HS deposits (Sillitoe, 1999; Sillitoe and Hedenquist, this volume); at Lepanto and Victoria, Philippines, these two styles of deposit are

closely related in both time and space (Claveria, 2001), although Victoria may be about 0.2 m.y. younger. Indeed there may be some affiliation between HS and IS epithermal deposits, as suggested by Reed (1997) on theoretical grounds, with differences in metal proportions (Cu-Au versus Pb-Zn-Ag-Au, respectively) likely resulting from progressive, wall-rock induced reduction and neutralization of reactive fluids, a process that affects concentrations of HCl, SO₂ and H_2S and consequently constrains metal concentrations in the fluid. A transition from HS to IS epithermal deposits would be analogous to the transition commonly seen in porphyry copper districts from proximal copper-dominated ores to distal lead-zinc-silver ores (Table 4).

Arc magmas and battle of the buffers

As noted earlier, the oxidation state of hydrothermal fluids from active systems associated with andesitic arc magmas plot within the envelope established by the sulfur-gas buffer and the rock buffer (Fig. 3). These two buffers overlap at magmatic temperatures, as gas species appear to be in equilibrium with andesitic magma, but they diverge to 3 units of R_H difference at 300 °C. Thus, a condensed magmatic fluid will become progressively more reactive relative to typical fresh host-rocks as the temperature decreases.

In contrast to volcanic fumarole compositions, where redox states are controlled by the sulfur-gas buffer, most vapor compositions related to lower temperature acidic magmatic-hydrothermal and near-neutral pH geothermal systems plot below the respective $H_2S_{(aq)} = HSO_4^-$ (pH 3) and $H_2S_{(aq)} = SO_4^-$ (pH 6) predominance boundaries (Giggenbach, 1997) within the $H_2S_{(aq)}$ field (Fig. 3), suggesting an absence of buffering by aqueous-sulfur redox equilibria at epithermal temperatures as concluded by Barton et al. (1977) for Creede. It is clear from Figure 3 that individual mineral buffers also do not match the fluid patterns.

Examination of fluid compositions in active systems on R_H and R_S diagrams (Figs. 3 and 5) highlights several features relevant to the above summary. High-temperature volcanic vapors have a linear trend that reflects the correlation between $R_H (X SO_2 / X H_2O)$ and $R_{SH} (X SO_2 / X H_2O)$ H_2S). Measured values of X SO_2 / X H_2S of about 1 to 10, (or R_{SH} of 0 to 1; Table 3) agree with the ratio predicted from X SO_2 / X H_2O . From their high-temperature magmatic origins, the cooling path of sampled magmatic-hydrothermal fluids from numerous individual localities (Table 3; tie-lines on Figs. 3 and 5) evolve to lower R_H and R_S values (higher oxidation and sulfidation states). This evolutionary trend may be a consequence of SO₂ and H₂S gases dominating other redox gases (e.g., CH_4 or H_2 ; Table 3). Where $SO_2 \approx H_2S$, the disproportionation reaction (equations 8, 9 and 10) will lead to fluids having relatively high oxidation and sulfidation states. Boiling may be important in some cases because it will result in progressive oxidation of the residual liquid due to loss of H₂ to the vapor phase. If the boiling liquid is H₂S-dominant, then the sulfidation state also will increase with a slope parallel to the boiling trend shown in Figure 5 (Appendix). Divergence from the gas buffer begins at 400°C, where SO₂ declines significantly, and below 300°C SO₂ gas joins S₂ gas as a negligible species for all reasonable hydrothermal environments. At these lower temperatures, H₂ and H₂S remain the dominant gases, and near the rock buffer $H_2 = H_2S$ ($R_s = 0$), typical of geothermal fluids (Table 3) and LS epithermal gold deposits.

Reduced geothermal systems and subalkalic LS epithermal deposits

In contrast with the acidic fluids of magmatic-hydrothermal systems, the near-neutral pH fluids from geothermal systems have generally lower sulfidation and oxidation states at 300-200°C, but overlap with the magmatic-hydrothermal fluids at 100°C near the rock buffer. As

suggested by the R_H and R_S diagrams (Figs. 4 and 5), the 300-200°C geothermal samples are all stable with respect to pyrite + magnetite, but lie close to the pyrrhotite field, consistent with the mineral assemblages in subalkalic LS epithermal deposits. Perhaps these LS deposits reflect a greater degree of water:rock interaction, or form from fluids with a smaller magmatic component. Alternatively, for some LS deposits hydrothermal fluids may be related to reduced tholeiitic magmas in rift settings (John, 2001) (Figs. 2B and 3). In reduced hydrothermal systems, simple cooling, even accentuated by boiling, is unlikely to result in any appreciable increase in sulfidation state. We interpret the geothermal fluids compiled in this paper to be analogs both for IS epithermal deposits and for subalkalic LS epithermal Au deposits of extensional arcs (e.g., Bodie, California; Hishikari, Japan: see Sillitoe and Hedenquist, this volume).

Concluding remarks

Variability of sulfidation and oxidation states of ore assemblages is a ubiquitous feature in the ore deposits considered here. This variability is a complex function of thermally prograding and retrograding hydrothermal episodes, and of intermittent events such as boiling, local wall rock interaction, or input of new hydrothermal fluids of magmatic or non-magmatic origin, all taking place as hydrothermal conduits shift location. On considering the complexities of sulfide assemblages at Butte, Sales and Meyer (1949) concluded that "the fact that a recognizably systematic district-wise pattern of mineralization and alteration has emerged is more surprising than the fact that there are frequent local divergences from that pattern" (p. 467). They conclude that microtextures between minerals are commonly ambiguous, and can only be interpreted within the context of the deposit-scale mineral-distribution patterns.

Considering individual epithermal deposits, local divergence of sulfidation state from the norm can lead to apparent inconsistencies with the overall classification into HS, IS and LS types. Although there may be problems with the classification scheme, as there are with many others, we stress that the classification terms highlight only one aspect (the sulfide mineral assemblage) of the combination of features, including magmatic, tectonic, hydrothermal, and geochemical, that should be used in classifying. Importantly, though, variability of sulfidation state within deposits blurs distinctions, suggests continua between deposit types, and indicates that common processes are active in different deposit types.

Regarding continua in terms of evolutionary paths, the sulfidation and redox environments of deposits discussed in this paper show a bimodal spectrum consisting of, on the one hand, oxidized porphyry copper and related base-metal veins and HS epithermal deposits, possibly including an affiliation with IS epithermal deposits down the flow path, and, on the other hand, relatively reduced LS epithermal gold deposits. A bimodal spectrum is also seen in active hydrothermal systems, with the first represented by fumaroles and magmatichydrothermal fluids associated with andesitic arc magmas, and the second represented by the neutral-pH geothermal systems.

The tendency for oxidized magmas to be associated with high-sulfidation deposits and for reduced magmas to be associated with low-sulfidation deposits is seen in the contrast between andesitic arc-related IS deposits and bimodal rift-related LS deposits in the western U. S. (John, 2001). However, there are many exceptions, including the relatively oxidized alkaline igneous rocks associated with deposits characterized by low to intermediate sulfidation states (Richards, 1995; Jensen and Barton, 2000). Differences among intrusion-related deposits is in part a consequence of the degassing of magmas that have different initial redox-volatile compositions,

but changes in these initial compositions due to loss of sulfur to the near-surface environment, seawater influx, or effects of local wall rocks are likely to blur the distinctions in many cases.

Samples from different portions of an active hydrothermal system provide information on the spatial evolution of fluids and the processes that influence fluid composition. The data on measured fluid compositions represent a single time frame in the life history of a given system, ranging from a single day of sampling (e.g., White Island, New Zealand; Giggenbach, 1987) to at most a half century of monitoring. By contrast, information from ore deposits provides an indication of the temporal and spatial evolution of hydrothermal systems that encompass a lifetime of several 10 thousand to a few 100 thousand years, although the link between time and space continues to be elusive in most districts. In this context, we note that samples from a variety of individual active systems span the whole time-space range seen in individual ore deposits of porphyry affiliation, reinforcing the concept of contemporaneous zonal growth proposed by Reno Sales and Charles Meyer.

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Figures



Fig. 1. Log $f_2 - 1000/T$ diagram defining the relative sulfidation state of hydrothermal fluids, the Main Line oreforming environment from Barton (1970), and the evolutionary path of fluids in porphyry copper and porphyryrelated vein deposits (see text for discussion of A, B, and C). lo = loellingite; other abbreviations in Table 4. Sulfidation reactions from Barton and Skinner (1979).





Fig. 3. $R_{\rm H}$ - 1000/T diagram showing redox states of magmas (Table 2) and fluids from active hydrothermal systems at 1 bar (Table 3), plotted over a grid of mineral reactions at 500 bars. Fluid compositions are deemed representative of their type, based on compilations or the original source, as referenced in Table 3. Dash-dot lines join fluid compositions from the same locality. Large star indicates composition of volcanic crater lakes (Giggenbach, 1992). The "rock buffer" and sulfur-gas buffer (isomolar SO_{2(g)} = H₂S_(g)) are from (Giggenbach, 1987). The predominance boundary H₂S = HSO₄⁻ at pH = 3 is relevant to magmatic-hydrothermal fluids; the boundary H₂S = SO₄⁻ at pH = 6 is relevant to geothermal fluids. Mineral abbreviations: Kspar = K feldspar; all others as in Table 4. See Appendix for further details on data sources and method of calculation.



Fig. 4. Log $fS_2 - 1000/T$ diagram, contoured for R_s , showing sulfidation state of magmas (Table 2) and fluids from active hydrothermal systems (Table 3) plotted over a grid of mineral sulfidation reactions at 1 bar. Dash-dot lines join fluid compositions from the same locality. Sulfidation states of magmas are based on pyrrhotite compositions in volcanic rocks using equations in Toulmin and Barton (1964) and Scott and Barnes (1971) combined with temperature estimates as in Figure 3. Sulfidation reactions used to define sulfidation states and those representing limits to assemblages commonly cited in the text are in bold; all sulfidation reactions are from Barton and Skinner (1979) and Myers and Eugster (1983) at 1 bar pressure with the exception of reaction tn = en which is based on natural occurrence (see text). See Table 4 for mineral abbreviations, and Appendix for details of calculations.



Fig. 5. $R_s - 1000/T$ diagram showing sulfidation state of magmas (Table 2) and fluids from active hydrothermal systems (Table 3) plotted over a grid of mineral sulfidation reactions at 1 bar. $R_s - 1000/T$ values for volcanic rocks as in Fig. 4 (sources in Table 2). Dash-dot lines join fluid compositions from the same locality. The trend expected for a liquid boiling from 300 to 150°C is based on data in Giggenbach (1980; see Appendix). SO₂ = H₂S gas buffer and Giggenbach's "rock buffer" cannot be calculated in terms of R_s and are approximated from their positions in Figure 3. Mineral reactions are from Barton and Skinner (1979) and Myers and Eugster (1983). Contours of mole percent FeS in sphalerite coexisting pyrrhotite or pyrite are from Scott and Barnes (1971) and Czamanske (1974). See Table 4 for mineral abbreviations and Appendix for log fS_2 to R_s conversion.



Fig. 6. Log $fS_2 - Log fO_2$ diagram at 250°C modified from Meyer and Hemley (1967) illustrating the looping path through time and space in porphyry and porphyry-related vein deposits. Lines of constant pH are dashed (bold where signifying predominance boundaries between aqueous sulfur species). Fields defined and labeled with roman numerals by Meyer and Hemley are: I. Covellite, digenite, chalcocite, and pyrite with advanced argillic or sericitic alteration; II. Chalcocite, bornite, pyrite, or chalcopyrite, bornite, pyrite with zoned sericitic and intermediate argillic alteration; III. Chalcopyrite + pyrite with zoned sericitic (relict K feldspar) and intermediate argillic; and IV. Chalcopyrite + magnetite, or bornite + magnetite with potassic alteration. Also shown are fields for the epithermal deposits from Heald et al. (1987) and John (2001). Mineral abbreviations as in Table 4.



Fig. 7. $R_s - R_H$ -T diagram constructed at 600, 300 and 100°C illustrating compositional relations between fluids from active systems and phase equilibria in the system Cu-Fe-S-O. Stability fields of bornite + magnetite and chalcopyrite + pyrite are highlighted by dark and medium shading, respectively. Fluid compositions from Table 3 are plotted on isothermal sections by projection along the average slope of tie-lines shown on Figures 3 and 5. Thermal windows used are: 900 to 440°C projected onto the 600°C section; 360 to 240°C onto the 300°C section; and 155 to 82°C onto the 100°C section. At 600°C the bulk fluid plots in the magnetite + bornite ± chalcopyrite field, at 300°C it straddles the chalcopyrite + magnetite and chalcopyrite + pyrite fields, and at 100°C it straddles the pyrite + chalcopyrite and pyrite + bornite ± digenite fields, a pattern that is consistent with assemblages documented in porphyry, porphyry-related vein deposits, and some epithermal deposits (Table 4). Phase relations among stoichiometric Cu-Fe-S-O minerals based on data in Barton and Skinner (1979) and Myers and Eugster (1983) recalculated to R_s and R_H as discussed in Appendix.



Fig. 8. Log $fS_2 - 1000/T$ diagram, contoured for R_s, illustrating fluid environments in porphyry copper, porphyrycopper related base-metal veins, and epithermal Au-Ag deposits in terms of a series of possible cooling paths that are influenced by processes discussed in text. Fluid environments are based on sulfide assemblages summarized in Table 4, temperatures discussed in text, and sulfidation reactions from Barton and Skinner (1979). Arrow labeled "porphyry Cu-Au-Mo" refers to early- and intermediate-age assemblages deposited at approximately 500 to 350°C and low to intermediate sulfidation states. Arrows labeled "N" and "O" represent deviations from the main-line environment. Transition to late assemblages at high sulfidation states and to porphyry-related "zoned base-metal veins" at or below 350°C is indicated by gray arrow labeled "T". Sulfide assemblages in the latter deposits describe a return from very high sulfidation to the intermediate sulfidation states of the "rock buffer". Similarly, HS epithermal deposits consist of an early Cu-rich, high sulfidation state assemblage followed by a Au-rich stage at intermediate sulfidation states. IS epithermal deposits (not shown) form at about 250 °C and mostly at intermediate sulfidation states. LS epithermal gold veins are shown near 200 °C and at low to intermediate sulfidation states near the rock buffer. Other porphyry-type deposits that form at low and even very low sulfidation states (dark, thin arrows labeled "Mount Pleasant" W-Mo and related Sn-Zn-Cu veins, and parts of Henderson Mo (labeled "H"; see text) are shown for contrast to porphyry Cu and related vein deposits.

Appendix

Calculation of Mineral and Gas Equilibria

Rock, mineral, and gas buffer curves

All redox data for volcanic and plutonic rocks (Table 2) are based on assemblages involving pyroxenes, iron-titanium oxides and titanite. Most have been recalculated in terms of T and log fO_2 for internal consistency using QUILF (Andersen et al., 1991; Andersen et al., 1993).

Giggenbach's (1997) "rock buffer" (Fig. 3) is based on the relation

$$4 \text{ FeO}_{15} = 4 \text{ FeO} + O_2,$$
 (A1)

with $a_{(\text{FeO})} / a_{(\text{FeO}1.5)} = 1$. This buffer approximates the average R_H value of subduction-related calc-alkaline magmas, and at 400 to 600°C it is close to both Ni + NiO and K feldspar + magnetite + quartz + phlogopite (mole fraction annite = 0.2).

Mineral equilibria plotted in Figures 3 and 5 and listed below (Table A) were calculated at 500 bars pressure using SUPCRT92 (Johnson et al., 1992) with the exception of hematitebearing reactions (Figs. 3 & 5) which were adjusted with equation 14 of Myers and Eugster (1983) in order to be consistent with prior R_H and R_S diagrams published by Giggenbach.

In the system Fe-S-O (Fig. 3), only the phase relations involving magnetite (or hematite) are shown for reference to magnetite-bearing igneous rocks; the stability field of $FeSO_4$ is ignored. Location of invariant point magnetite + hematite + pyrite + pyrrhotite + $s_{(1)}$ is based on the calculated intersection of the reactions pyrite = pyrrhotite + $s_{(1)}$ and pyrite = magnetite + $s_{(1)}$.

The SO_{2(g)} = $H_2S_{(g)}$ isomolar curve ("sulfur-gas buffer"),

$$2/3 H_2 S_{(gas)} + O_2 = 2/3 SO_{2(gas)} + 2/3 H_2 O,$$
 (A2)

plotted on log (f_{H_2} / f_{H_2O}) – 1000/T (Fig. 3),was calculated using standard equilibrium constants of formation at different total pressures (D. R. Stull, 1969, in Giggenbach, 1987).

The equilibrium boiling trend for residual liquid (pure water) illustrated in Figure 5 was calculated from 300°C to 100°C using vapor-liquid gas distribution coefficients for H_2 and H_2S from Table 2 of Giggenbach (1980). The boiling trend reflects the higher solubility of H_2 relative to H_2S in the vapor phase.

Recalculation of total discharge (Table 3)

We calculated R_H and R_s values for the wells listed in Table 3, which were reported for total discharge (all at temperatures <300°C), to allow comparison with vapors discharged from fumaroles. Calculations used temperature-dependent vapor-liquid gas distribution coefficients, $B_i = x_{v,i}/x_{Li}$ for H₂ and H₂S from Table 2 of Giggenbach (1980).

Justification of $R_H \approx \log (XH_2 / XH_2O)$ and $R_S \approx \log (XH_2 / XH_2S)$ for modern fluids

We follow Giggenbach (1987, p. 146-147) in making the assumption that for gases in which H_2O is the major component, the uncorrected analytical mole ratio (e.g., XH_2 / XH_2S) is approximately equivalent to the fugacity ratio (e.g., fH_2 / fH_2S). That is, the ratio of the activity coefficients do not deviate much from unity. This assumption introduces unknown errors in comparing analytical R_H and R_S values for fluids with computed R_H and R_S values for mineral reactions (Figs. 3, 4 and 5).

R_H and $log fO_2$

Log *f*O2 values calculated for rocks, mineral reaction curves, the sulfur-gas buffer, and the predominance boundaries for aqueous sulfur species (Table A) were converted to R_H , where $R_H = \log (fH_2 / fH_2O)$, using equation 15.6 of Giggenbach (1997):

$$\log fO_2 = 5.3 - 25,552/T - 2 R_{\rm H},\tag{A3}$$

for T in Kelvins.

R_s and $\log fS_2$

Log fS_2 values calculated for sulfide mineral reactions were converted to R_s , where $R_s = \log (fH_2 / fH_2S)$, in Figure 5, and R_s values for fluids given in Table 3 were converted to $\log fS_2$ values in Figure 4, on the basis of the free energy change for the reaction

$$2 H_2 S_{(g)} = 2 H_{2(g)} + S_{2(g)}$$
 (A4)

at 1 bar (Barton and Skinner, 1979), yielding the expression

$$\log f S_2 = 4.9 - 9,251/T - 2 R_8, \tag{A5}$$

for T in Kelvins.

Approximate sulfur-gas buffer on $\log fS_2 - 1/T$ and $R_S - 1/T$ plot

The relation

$$2/3 H_2 S_{(0)} + 4/3 H_2 O + S_2 = 2/3 SO_{2(0)} + 2 H_2 S_{(a0)}$$
(A6)

can be used to plot the sulfur-gas buffer on a log $fS_2 - 1000/T$ diagram at constant $aH_2S_{(aq)}$ and constant pressure (Inan and Einaudi, 2002). Reaction A6 has a moderately negative slope on a log $fS_2 - 1000/T$ diagram, indicating that if a hydrothermal fluid were to cool along the sulfur-gas buffer at constant pressure and constant $aH_2S_{(aq)}$, then the sulfidation state would increase at a significantly higher rate than is documented by active magmatic-hydrothermal systems or deduced from sulfide assemblages in ore deposits formed at T > 350°C. In order to achieve a log $fS_2 - 1/T$ trajectory such as shown on Figures 1, 2A, 4, and 8 above 350°C and retain $SO_{2(g)}/H_2S_{(g)} = 1$, hydrothermal fluids must depressurize, and/or loose sulfur. Such changes, appropriate for the high-temperature porphyry environment, could lead to the relatively flat trajectories shown on Figures 4 and 5 as "approximate $SO_{2(g)} = H_2S_{(g)}$ gas buffer".

Tables

Table 1. History of concepts involving the aqueous S-H-O system and sulfidation state as applied to fluids and mineral assemblages in porphyry Cu and epithermal Au-Ag deposits

Concept or Process	Definition or Conclusion	Sources	Comments
1906-1909	"Direct volcanic hypothesis", ore solutions	Ransome, 1907, 1909	Recognized presence of chemically
Hypogene alunite in epithe	emanated from magma, acid and then		distinct epithermal environments
ores	neutralized by reaction with rock. Later		following study of Goldfield deposit,
	changed to "solfatarism and oxidation		Nevada; interpreted close association of
	model", alunite formed by oxidation of		alunite with ore minerals as result of
	magmatic aqueous sulfur in the near-		acidic hypogene fluid related to
	surface.		fumaroles from degassing magma.
1933	Deposits classed on the basis of ore	Lindgren, 1933	Concluded that epithermal deposits
Classification of	mineralogy as either gold-, silver-, or		formed from near neutral-pH waters
epithermal deposits	base-metal rich.		similar to those discharged from hot
			springs; qualifyingly accepted Ransome's
			alunite-quartz type.
1933-1950	Magmatic-hydrothermal fluids start off	Bowen, 1933; Fenner, 1933	Acidic origins of magmatic-
	acid and become alkaline as they react		hydrothermal fluids, based on studies of
	with their wall rock.		Yellowstone hot springs, contrary to
			Graton (below).
Alkaline or acid	Magmatic-hydrothermal fluids start off	Graton and Bowditch 1036	Alkaline origins of magmatic-
magmatic volatiles?	alkaline and become acid through boiling	Graton and Downten, 1950	hydrothermal fluids based on studies of
magmatic volatiles:	and hydrolysis of H.S		hase-metal ores at Cerro de Pasco. Peru
	and nyurorysis of 1125.		contrary to Bowen and Fenner (above)
			contaily to Dowen and Penner (above).
Hot springs-epithermal	From explosive volcanic phenomena and	Schmidt, 1950	Schmidt notes that "acid sulfate" phase
analogy	fumaroles to geysers and hot springs,		of alteration at Goldfield is equivalent to
	chemical environment varies from acid &		the kaolinite-alunite-quartz alteration
	oxidizing to alkaline & reducing.		seen in active hot-springs, both the result
			of largely unmodified magmatic gases.
			Further notes clay alteration by steam-
			heated groundwaters.
1947-1963	Sulfide assemblages with a high sulfur-	Sales and Meyer, 1949;	The sulfur content of an assemblage is
Sulfur content of sulfide	content denote a high chemical potential	McKinstry, 1959, 1963	not an intensive variable. Therefore, this
assemblages:	of sulfur.		implication is correct only if metal ratio
implications for			in successive assemblages (in this case
chemical potential of S	Increasing sulfur content of assemblages	Sales and Meyer, 1949;	Cu:Fe at Butte, Montana) does not
	in paragenetic sequence implies increasing	McKinstry, 1959, 1963	change, a situation unlikely to be the
	chemical potential of S through time.		general case, as noted by Gustafson
10/7		M 111 1 10/7	(1963).
	Formation of an assemblage of higher	Meyer and Hemley, 1967	As defined, term does not require an
"Sulfidation"	sulfur-content.		increase in $f S_2$. For example, the
			reaction magnetite + 3 $S_2 = 3$ pyrite + 2
			O_2 can proceed isothermally on
			decreasing $f S_2$ if $f O_2$ decreases
			sumciently.
Path of "increasing	Assemblages at Butte record an increase	Meyer and Hemley, 1967	First integration of mineral assemblages
sulfidation"	in both $f \tilde{S_2}$ and $f O_2$ with time from early		and geochemistry to arrive at
	py-bn-cp to late py-cv-dg. ¹		evolutionary path for hydrothermal fluids
			in porphyry and vein systems in context
			of sulfidation and oxidation states.

1967 Link between $f S_2$ and $f O_2$	Relatively high acidity, high $f S_2$ and $f O_2$ at high ratios of $f S_2/f O_2$ are implied by assemblages such as py-cv-dg.	Meyer and Hemley, 1967	Solution composition commonly remains close to $H_2S = H_2SO_4$ boundary. If, on cooling, H^+ generation accelerates, the cooling path will diverge toward higher <i>f</i> S_2 and then trend back toward the rock buffer (i.e., the "looping path"). ²
1960-1970 "Sulfidation reaction" ³	A reaction that produces a new assemblage of higher sulfidation state (not necessarily one of higher sulfur-content)	Barton et al., 1960; Barton and Skinner, 1967; Barton, 1970	Term generally applied to reactions that involve sulfur in the form of hypothetical S_2 gas and that appear as univariant lines on a log $f S_2 - 1/T$ diagram. Example: 2 FeS + $S_2 = 2$ FeS ₂ . Conceptually analogous to oxidation/reduction reactions involving O_2 or H_2 ; distinct from "sulfidation" (formation of an assemblage of higher sulfur content).
"Sulfidation state " ⁴ (of sulfide assemblages and fluids)	A function of both $f S_2$ and T, the sulfidation state is defined relative to experimentally determined sulfidation reactions.	Barton et al., 1960; Barton and Skinner, 1967; Barton, 1970	Sulfide "petrogenetic grid" $(\log f S_2 - 1/T)$ introduced, describing a continuum of sulfidation states, which can increase or decrease during fluid evolution. General path on cooling is deduced to be one of increasing sulfidation state (Barton's 1970 "Main Line").
1977-1987	Mineral assemblages at Creede, Colorado, indicate predominance of SO ₂ .	Barton et al., 1977	Determined activities of S_2 and O_2 , pH, total S, in equilibrium with hm, py, Fe-chl, relatively oxidized assemblage.
SO ₂ /H ₂ S gas ratios "sulfur-gas buffer"	Ratios of dominant magmatic sulfur gases are near unity, an equilibrium that, if maintained, is an oxidizing $fO_2 - T$ cooling path relative to FMQ or HM buffers.	Burnham and Ohmoto, 1980; Giggenbach, 1987	SO ₂ /H ₂ S fugacity ratio of magmatic vapor phase andesitic magmas ranges from 0.1 to 10; in absence of wall-rock exchange, ratio may remain unchanged as vapors cool to 500 to 350°C.
	Magmatic gas ratios calculated from data on Fe-Ti-oxide and pyrrhotite compositions.	Whitney and Stormer, 1983; Whitney, 1984	SO ₂ /H ₂ S fugacity ratios for the Fish Canyon tuff, San Juan Mtns, Colorado, range from 0.6 to 2.9, average 1.6.

1981-1994 Classification of porphyry-related and epithermal deposits	Classification of porphyry-related systems based on sulfidation & oxidation state of mineral assemblages combined with ore morphology.	Einaudi et al., 1981; Einaudi, 1982	Oxidized (porphyry-Cu, Cu-skarns.) vs reduced (Sn-W porphyries, W skarns) based on mineral assemblages; correlation with magma types. Sulfidation state contrasts between endmember porphyry-Cu (intermediate sulfidation state) and porphyry-related base-metal vein/replacement deposits (high and very high sulfidation states).
	Epithermal deposits classed by mineralogy and deduced fluid chemistry into "Adularia-Sericite" and "Acid Sulfate".	Bethke, 1984; Hayba et al., 1985; Heald et al., 1987	Endmember chemical environments proposed for epithermal deposits with implications for sulfidation & oxidation states: adularia-sericite relatively lower sulfidation-oxidation state than acid- sulfate deposits.
	Epithermal deposits classed by high (HS) vs low (LS) oxidation (valence) states of sulfur (n) in the fluid (e.g., $n = +4$ in SO ₂ and $n = -2$ in H ₂ S).	Hedenquist, 1987; White and Hedenquist, 1990	Proposed that an intrinsic characteristic of the fluid of the two end-members be used for classification: acidic and oxidized vs neutral pH and reduced. Terminology does not relate directly to sulfide mineralogy or sulfidation state, yet it results in classification of epithermal deposits similar to acid- sulfate (HS) and adularia-sericite (LS).
	Epithermal deposits classed by dominant sulfides and deduced fluid chemistry into "high sulfidation" (HS) and "low sulfidation" (LS) types.	Hedenquist et al., 1994a	Noted that typical sulfide assemblages in these two end-member systems consisted of pyrite-enargite and pyrite±pyrrhotite with a variety of base metal sulfides, i.e., relatively high and low sulfidation states respectively, as defined by Barton and Skinner (1967).
High (HS), intermediate (IS), and low sulfidation (LS) epithermal deposits	Classification of epithermal deposits based on tectonic settings, igneous associations, sulfidation state, and oxidation state.	John et al., 1999; Hedenquist et al., 2000; John, 2001	Integration of tectonic settings, intrinsic oxidation state of magmas, and mineralogy of deposits leads to a proposed continuum of sulfidation states in HS and IS epithermal deposits, contrasted with distinct LS deposits.

¹ for mineral abbreviations, see Table 4.

² See Barton and Skinner (1967), p. 231-232 and their Fig. 6.9. ³ See Barton and Skinner (1967, 1976) for an authoritative review of the geochemistry of sulfide minerals.

⁴ Definition of sulfidation state used in present paper, with respect to mineral assemblages and fluids. This table is restricted to the porphyry and epithermal environments and to processes & concepts related to volatile components and their link to sulfidation state; it thereby leaves out important contributions on topics of ore-depositional processes, evidence from light stable isotopic and fluid inclusion studies, and classifications based on morphology or metal content.

Name in Figure keys	Location	Geologic Province/Tectonic Setting	Magma Series	Rock Type	Source for oxidation State (Rh)	QUILF used to recalculate T and fO_2^{-1}	Source for sulfidation state (Rs) ²
Volcanic Rocks							
Pinatubo	Western Luzon, Philippines	Active Island arc	CA	D	Imai et al. (1993)	No; authors use QUILF	no data
Rabaul	New Britain, Papua New Guinea	Active Island arc	CA	A, D, R	Heming and Carmichael (1973)	No; authors use QUILF	Heming and Carmichael (1973)
Satsuma- Iwojima	Kikai caldera, Japan	Active Island arc	CA	D	no data		Ueda and Itaya (1981)
Nevado del Ruiz	Central Cordillera, Colombia	Active Continental arc	CA	A, D	Fournelle (1990); Melson et al. (1990)	Yes	no data
St. Helens	Mount St. Helens, Cascades, Washington, USA	Active Continental arc	CA	D	Melson and Hopson (1981)	Yes	Nordlie et al. (1983)
Bishop Tuff	Long Valley caldera, California, USA	Continental, transtensional	CA	R	Hildreth (1977)	Yes	Hildreth (1977)
Fish Canyon	La Garita caldera, Colorado, USA	Central San Juan volcanic Field	СА	QL	Xirouchakis et al. (2001)	Authors provide lower limit based on QUILF	Whitney and Stormer (1983)
Julcani	Peru	Continental arc	KCA	D, R	Drexler and Munoz (1988)	Yes	Drexler and Munoz (1988)
Western Andesite	Great Basin, Nevada (Tonopah, Paradise Range and Kate Peak) and Walker Lane, California (Bodie Hills), USA	Continental arc, transtensional	KCA	A, D, R	John (2001)	Author uses QUILF	no data
Northern Nevada Rift	Great Basin, Nevada & Oregon, USA	Continental rift	KT	B,BA, R	John (2001)	Author uses Quilf	no data
Taupo & Coromandel	North Island, New Zealand	Taupo Volcanic Zone, Coromandel peninsula	CA	RD	Rutherford and Heming (1978)	Yes	Rutherford and Heming (1978)
Azores – Canary Islands	Mid-Atlantic Ridge, Spain	MOR related fracture zones	А	В	Wolff and Storey (1983)	Yes	Wolff and Storey (1983)
Plutonic Rocks							
La Gloria	Central Chilean Andes, Chile	Continental magmatic arc	KCA	GD, QM	Cornejo and Mahood, (1997)	Yes	no data
Yerington	Yerington batholith, Nevada, USA	Continental magmatic arc	CA	GD, QM	Dilles (1987)	Yes	no data
Wasatch-Oquirrh	Wasatch (Park	Continental back-	KCA	М	Core (2001)	Yes	no data

Table 2. Sources of data on the oxidation and sulfidation states of magmas plotted on Figures 3, 4, and 5

	Premier stock) and Oquirrh (Last Chance stock) Mountains, Utah	arc					
Oslo Rift	Central & Southern Oslo Rift, Norway	Continental rift	А	М	Neumann (1976)	Yes	no data

¹QUILF (Andersen et al., 1991; Andersen et al., 1993) was used to recalculate temperature and log fO_2 from data on assemblages involving pyroxenes, iron-titanium oxides and titanite given in the references listed. Log fO_2 values were then converted to Rh (Rh \approx log [X H₂ / X H₂O]) using equation 15.6 of Giggenbach (1997). ² Log/S₂ – 1/T values for volcanic rocks are taken from the references listed, without recalculation, and converted to Rs (Rs \approx log [X H₂ / X H₂S

]) using the relations $2 H_2S_{(g)} = 2 H_{2,(g)} + S_{2,(g)}$ from SUPCRT92 (Johnson et al., 1992). Magma series abbreviations: A = alkalic, CA = calcalkalic, KCA = high-K calcalkalic, KT = high-K tholeiitic

Rock type abbreviations: see Table 3.

Table 3. Measured compositions of vapors discharged from volcanic fumaroles, magmatic-hydrothermal and geothermal fumaroles,

Receive Reprint T.CC F.Q.* R.J.*	plus total discharge from some geothermal wells.													
Valencie functorieValencie valueValue	System	Location	Rock type ¹	T (°C)	H ₂ O ²	CO ₂ ²	SO_2^2	H ₂ S ²	HCl ²	HF ²	H ₂ ²	$R_{\rm H}^{3}$	$R_{s}^{4}R_{sH}^{5}$	Refs ⁷
Erra'Ale Ethiopia AB 1130 794000 794000 7800 8415 752 716 62 230 721 0.30 10.1 733 Galeras Colombia D 642 91680 900 00 1200 200 221 2.47 1.65 a. Momotombo Nicaragna BA 844 90500 1600 1200 600 200 0.70 0.23 0.81 3.0 3.00 2.22 2.47 1.65 a. Mutusa SE Hokaido D 600 9000 16050 1200 3.0 1.00 2.22 2.67 0.70 2.30 0.65 0.13 0.55 0.13 0.55 0.13 0.55 1.81 1.60 1.00 0.00 1.00 0.00 2.00 1.00 0.00 2.10 0.50 0.10 0.50 0.10 0.50 0.10 0.10 0.00 1.00 0.10 0.00 1.00 0	Volcanic fumaroles													
Galeras Colombia D 642 914800 59800 8415 5725 7160 620 250 -2.51 -0.30 0.17 ag Klyuchevskoy Kamchafka B 1100 97700 100 300 300 300 200 -2.64 0.25 1.5 a.h Munctorus Felokkaido D 640 96000 1600 1200 6800 2.00 1.420 2.83 -0.68 0.37 a.g St.Helens* Washington D 650 960000 1500 150 150 150 2.76 0.70 2.0 1.20 2.83 0.680 1.00 1.00 2.00 1.20 2.80 1.20 2.80 1.20 2.80 1.20 2.80 1.20 2.80 1.20 2.80 1.20 2.20 1.10 2.0 2.0 0.00 2.0 2.90 1.20 2.40 2.80 1.20 2.80 1.00 1.00 1.00 <td>Erta'Ale</td> <td>Ethiopia</td> <td>AB</td> <td>1130</td> <td>794000</td> <td>104000</td> <td>67800</td> <td>6200</td> <td>4200</td> <td>na</td> <td>14900</td> <td>-1.73</td> <td>0.38 1.04</td> <td>a,b</td>	Erta'Ale	Ethiopia	AB	1130	794000	104000	67800	6200	4200	na	14900	-1.73	0.38 1.04	a,b
Kiyachevskoy Kamchauka B 1100 977300 1200 900 20 1200 200 220 2.47 1.65 a. Momotombo Nicurangua BA 844 95000 200 120 100 020 300 200 120 020 120 0.210 0.250 1.5 1.	Galeras	Colombia	D	642	914800	59800	8415	5725	7160	620	2850	-2.51	-0.30 0.17	a,g
Momotombo Nicaragua BA 844 95050 2500 7000 4900 5000 2500 12.0	Klyuchevskoy	Kamchatka	В	1100	977300	1200	900	20	14200	2500	5900	-2.22	2.47 1.65	a,k
Mit. Usu SE Hokado D 690 93000 320 180 340 60 200 1.00 <th< td=""><td>Momotombo</td><td>Nicaragua</td><td>BA</td><td>844</td><td>950500</td><td>23800</td><td>7000</td><td>4900</td><td>3500</td><td>290</td><td>8700</td><td>-2.04</td><td>0.25 0.15</td><td>a,h</td></th<>	Momotombo	Nicaragua	BA	844	950500	23800	7000	4900	3500	290	8700	-2.04	0.25 0.15	a,h
NganuhoeNew ZealandA6409600016050160086002702808701403.41.430.73aPapandayanMadmensiaN806072000700200600700200600700200600700200600700200600700200600700200600700200600700200600700200600	Mt. Usu	SE Hokkaido	D	690	993000	3600	320	180	340	60	2000	-2.70	1.05 0.25	a,f
Papandayan Indonesia A 400 95000 29300 2750 2120 3.3 1.40 -1.32 0.13 0 St. Helens ^k Washington D 650 9800 8700 670 670 200 6707 2.21 0.57 0.57 0.51 650 5180 100 1500 650 100 1500 650 100 1500 650 100 1500 200 5180 300 2.24 0.05 2.2 d White Island, 84 New Zealand D 470 27000 2700 1500 6000 1600 1600 1700 2.20 1.10 0.20 1.10 0.50 570 2.21 1.10	Ngauruhoe	New Zealand	А	640	960000	16050	10200	6800	2500	na	1420	-2.83	-0.68 0.18	a
St. Helens ⁶ Washington D 650 986000 8800 670 900 700 200 400 -2.39 d.6.5 -1.1 0 Satsuma Ivojima S Kyushu R 877 97300 5095 5180 1640 530 201 6300 21.9 32.0 3.3 Vulcano Italy TB 620 86100 19000 5200 150 450 150 20.0 300 21.0 3.20 2.40 0.09 0.20 2.40 18.0 0.00 0.00 0.00 0.00 0.00 2.000 2.40 2.30 0.10 0.00	Papandayan	Indonesia	А	400	959000	29360	6750	2900	1250	33	140	-3.84	-1.32 0.37	a,e
Satsuma Iwojima S Kyushin R 877 973600 5095 8180 1640 500 277 6070 -2.19 3.20 2.03 aj Showa Shinzan?? E Hokknido D 8000 52000 430 480 4530 100 50.00 5.20 0.10 5.00 2.19 3.02 2.03 aj White Island, 84 New Zealand D 457 941000 2010 1200 3.00 1245 na 5.40 -3.24 -0.10 0.22 af Augustine* Alaska D 457 91000 1000 800 10.00 800 10.00 1.00 -1.0 0.22 af Augustine* Alaska D 870 92100 27000 2300 800 1000 1000 1000 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00 1.00	St. Helens ⁶	Washington	D	650	986000	8800	670	900	700	200	4000	-2.39	0.65 -0.13	0
Showa Shinzari? SE Tokkaido D 800 980000 12000 430 4 530 240 6300 -2.10 3.20 2.03 a,j Vulcano Italy TB 620 861000 19000 6800 4800 430 101 150 3.00 3.02 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 2.40 0.30 0.40 3.20 2.01 0.00 0.20 0.00 0.20 2.00 0.00 0.00 0.00 0.20 0.00	Satsuma Iwojima	S Kyushu	R	877	973600	5095	8180	1640	5306	277	6070	-2.21	0.57 0.70	a,I
Vulcano Indy TB 620 86100 19000 6800 4800 4500 1010 1510 2.76 6.00 0.50 2.76 White Island, 84 New Zealand D 760 927000 5200 300 1240 na 540 -2.49 0.90 0.52 cd White Island, 84 New Zealand D 457 944000 2200 61800 6000 1000 860 5400 -2.20 -1.12 na na 0 Augustine * Alaska D 870 92000 2000 2300 500 600 1000 -1.12 na na 0 Kudryavy, 91 S Kurils BA 585 951000 2000 7700 9200 1200 500 500 500 500 500 500 500 2.25 1.25 0.35 0.43 9 Lewotolo Lomben, NTT A 90 94000 200 100<	Showa Shinzan??	SE Hokkaido	D	800	980000	12000	430	4	530	240	6300	-2.19	3.20 2.03	a.i
White Island, 84 New Zealand D 760 92700 8150 8150 710 300 302 2.49 0.09 0.52 c.1 White Island, 84 New Zealand D 457 941000 2000 1200 1300 1240 18 400 -324 -0.70 6.0 1 300 1200 8400 -324 -0.70 6.0 1 300 100 600 100 600 100 600 100 600 100 600 100 600 100 700 2.21 1.8 0.22 1.0 0 100 100 100 100 100 100 100 100 1.00	Vulcano	Italv	ТВ	620	861000	119000	6800	4800	4530	1010	1510	-2.76	-0.50 0.15	a
White Island, 84 New Zealand D 457 94100 29100 3200 3300 12450 na 540 -3.24 -0.79 0.60 d Minte Island, 88 New Zealand D 457 92000 6200 6000 200 280 0.30 -3.26 -1.18 0.22 a,f Augustine * Italy AB 928 91000 14002 2800 6000 600 500 700 -2.72 -0.36 0.48 m Kudryavy, 91 S Kurils BA 585 951000 2400 1800 500 500 700 -2.72 -0.36 0.48 m Lewotolo Lomblen, NTT A 900 94000 200 1100 300 100 500 2.22 1.26 0.56 v Unzen N Kyushu A 81 9200 200 100 300 100 200 2.50 a.23 0.50 2.23 1.26 0.56 v 1.12 1.26 0.50 2.20 1.12 1.20 <td>White Island, 84</td> <td>New Zealand</td> <td>D</td> <td>760</td> <td>927000</td> <td>52900</td> <td>8150</td> <td>2450</td> <td>5180</td> <td>300</td> <td>3020</td> <td>-2.49</td> <td>0.09 0.52</td> <td>c.d</td>	White Island, 84	New Zealand	D	760	927000	52900	8150	2450	5180	300	3020	-2.49	0.09 0.52	c.d
White Island, 88 New Zealand D 495 92000 6100 1000 6000 2400 28 400 -2.20 -1.18 0.22 a,1 Augustine ^ Ialaka D 8470 2100 2200 1200 1200 1200 100 860 600 -0.00 -2.12 ana na	White Island, 84	New Zealand	D	457	941000	29100	13200	3300	12450	na	540	-3.24	-0.79 0.60	d
Augustine 6 Alaska D 870 847700 22700 61800 6800 1010 860 5400 -2.20 -0.10 0.96 x Enne 6 Italy AB 928 919000 14000 28000 na 25005 5000 7000 -2.12 na na o Kudryavy, 91 S Kurils BA 855 95100 24000 1700 9200 1260 369 540 -3.15 -1.23 0.93 y Merapi Java A 900 940000 4300 5000 5000 200 100 3000 5000 2.27 1.26 0.56 v Unzen N Kyushu A 818 95000 2700 430 2610 na 140 -3.85 -0.23 -1.15 n El Chichon, 83 Mexico TA 115 992400 7100 170 240 250 na 140 -3.85 -0.23 -1.15 n a E higui wojama, 81 Styushu R <	White Island, 88	New Zealand	D	495	920000	60100	10000	6000	2400	28	400	-3.36	-1.18 0.22	a.f
	Augustine ⁶	Alaska	D	870	847700	22700	61800	6800	10100	860	5400	-2.20	-0.10 0.96	x
Kudryavy, 91 S Kurils BA 870 921000 27300 2330 580 6600 610 1300 -1.85 0.35 0.60 m Kudryavy, 91 S Kurils BA 585 951000 2400 11800 3000 5000 570 1700 -2.75 -0.36 0.48 m Lewotolo Lomblen, NT A 490 766000 12000 7700 9200 160 500 -2.27 1.00 0.00 w Tolbalchik Kamchatka B 1020 97400 200 1010 300 1040 300 5.00 -2.27 1.02 0.00 w Unzen N Kyushu A 818 955000 2400 1300 1010 170 120 250 na 140 -3.85 0-2.3 - 1.5 n El Chichon, 85 Mexico TA 115 992400 7100 170 170 420 250 na 140 -3.85 0-2.3 - 1.5 n a.2 El Chichon, 85 Mexico	Etna ⁶	Italv	AB	928	919000	14000	28000	na	25000	5000	7000	-2.12	na na	0
Kudryavy,91 S Kurils BA 585 95100 2490 1800 3900 5000 570 170 2.75 0.36 0.48 m Lewotolo Lomblen, NTT A 490 766000 12000 7700 9200 120 369 540 3.15 -1.23 0.93 y Merapi Java A 900 94000 4300 5000 2000 100 300 500 2.25 1.26 0.56 v Tolbalchik Kamchatka B 1020 979400 200 470 240 250 na 140 -3.85 -0.23 -0.15 n Magmatic hydroth. EI Chichon, 95 Mexico TA 101 841600 14260 400 220 400 6 104 -3.91 -1.5 n EI Chichon, 95 Mexico TA 101 841600 14260 400 1080 200 100 6 108 4.5 -2.00 -0.19 p Kuju I	Kudrvavv, 91	S Kurils	BA	870	921000	27000	23300	5800	6600	610	13000	-1.85	0.35 0.60	m
Lewotolo Lomblen, NTT A 490 766000 12000 7770 920 1260 369 540 -3.15 -1.23 0.93 y Merapi Java A 900 940000 43000 5000 5000 5000 5000 5000 2.27 0.00 0.00 w Tolbalchik Kamchatka B 1020 97400 200 100 300 2430 2610 na 740 -2.09 0.50 0.23 0.74 n Magmatic hydroth. EI Chichon, 85 Mexico TA 101 841600 14260 400 6 1040 -3.85 -0.23 -0.74 n EI Chichon, 95 Mexico TA 101 84100 142600 400 970 100 -2.99 -1.15 -3.1 p Kuju Iwoyama, 61 S Kyushu R 165 992000 1800 2800 1200 1200 160 15	Kudryavy, 91	S Kurils	BA	585	951000	24900	11800	3900	5000	570	1700	-2.75	-0.36 0.48	m
Merapi Java A 900 94000 4300 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5000 5200 2.27 0.00	Lewotolo	Lomblen, NTT	А	490	766000	120000	77700	9200	1260	369	540	-3.15	-1.23 0.93	y
Number Kamchatka B 1020 979400 200 100 300 1400 500 52.05 1.26 0.56 v Unzen N Kyushu A 818 95500 2700 4370 2430 2610 na 740 2.29 0.50 0.25 u Magmatic hydroth. EI Chichon, 83 Mexico TA 115 992400 7100 170 240 250 na 140 -3.85 0.23 - 0.15 n EI Chichon, 95 Mexico TA 101 81400 12600 400 200 400 70 100 6104 -2.91 0.33.7 n Kuju Iwoyama, 61 Skyushu R 320 99300 910 1800 2800 1200 420 28 4.55 2.00 -1.9 Kuju Iwoyama, 84 Skyushu R 165 99200 1400 2800 100 155 20 4.68 2.11 0.55 70	Merapi	Java	А	900	940000	43000	5000	5000	2000	100	5000	-2.27	0.00 0.00	w
Unzen N Kyushu A 818 95500 2700 4370 2430 2610 na 7740 -2.09 0.50 0.25 u Magmatic hydroth. E1 Chichon,83 Mexico TA 115 992400 7100 170 240 250 na 140 -3.85 -0.23 -0.7 n E1 Chichon,95 Mexico TA 101 841600 142600 400 2800 400 610 2.40 61 140 -2.35 -0.33 -0.7 n E1 Ruiz Colombia A 82 95200 2400 1800 170 280 120 420 -2.03 -2.01 9.3 -2.1 0.93 -2.1 0.95 -2.1 0.95 -2.10 9.5 -2.10 9.5 -2.00 -2.01 9.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.00 2.01 9.01 9.00 2.00 2.00 2.00 2.00 2.01 9.00 2.01 9.00 <	Tolbalchik	Kamchatka	В	1020	979400	200	1100	300	10400	3000	5500	-2.25	1.26 0.56	v
Magmatic hydroth. El Chichon, 83 Mexico TA 115 992400 7100 170 240 250 na 140 -3.85 -0.23 - 0.15 n El Chichon, 95 Mexico TA 101 841600 142600 400 200 400 6 1040 -0.33 -0.74 n El Ruiz Colombia A 82 95500 23400 19800 1760 59 <10	Unzen	N Kyushu	А	818	955000	27000	4370	2430	2610	na	7740	-2.09	0.50 0.25	u
El Chichon, 83 Mexico TA 115 992400 7100 170 240 250 na 140 -3.85 -0.23 -0.15 n El Chichon, 95 Mexico TA 101 841600 142600 400 2200 400 6 1040 -2.91 -0.33 -0.74 n El Ruiz Colombia A 82 955000 23400 19800 1760 59 <10 86 4.05 -1.31 1.05 a.e Kuju Iwoyama, 61 S Kyushu R 400 96900 4500 6900 14100 4800 970 1000 -2.99 -1.15 -0.31 p Kuju Iwoyama, 84 S Kyushu R 155 99200 1800 2200 2600 1800 67 5 -5.30 -2.72 -0.07 p Kudryay, 91 S Kurils BA 130 95000 22500 2300 2600 110 5 20 4.68 -2.11 0.95 m Manza karabuki NE Honshu A 94 99200 1460 220 6290 20 na 0.54 -6.26 -4.07 -1.46 z Mutnovsky funnel, 83 Kamchatka BA 360 98700 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a,] Mutnovsky S bottom. 83 Kamchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky N bottom. 83 Kamchatka BA 272 98800 9820 290 1020 160 10 350 -3.45 -0.46 -0.55 1 Mutnovsky N bottom. 83 Kamchatka BA 106 981000 2880 280 120 160 10 350 -3.45 -0.46 -0.55 1 Mutnovsky N bottom. 83 Kamchatka BA 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 99890 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 285 91800 69000 2980 120 500 na 0.14 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 107 94700 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 107 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 107 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 107 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 4220 170 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, 104 94200 54000 nd 1820 2080 60 na 122 -3.00 -1.15 -0.81 d Zao NE Honshu A, 104 94200 54000 nd 2440 87 na 945 -3.00 -0.41 na c Krafla Leeland B 300 98500 13150 nd 350 nd na 360 -3.44 0.01 na c	Magmatic hydroth													
Licensel, os Inclusion Inclusion <td>El Chichon 83</td> <td>Mexico</td> <td>ТА</td> <td>115</td> <td>992400</td> <td>7100</td> <td>170</td> <td>240</td> <td>250</td> <td>na</td> <td>140</td> <td>-3 85</td> <td>-0 23 -0 15</td> <td>n</td>	El Chichon 83	Mexico	ТА	115	992400	7100	170	240	250	na	140	-3 85	-0 23 -0 15	n
Ele Ruiz Colombia A 82 955000 23400 19800 1760 59 <10 86 4.05 -1.31 1.05 ac El Ruiz Colombia A 82 955000 23400 19800 1760 59 <10	El Chichon 95	Mexico	ТА	101	841600	142600	400	2200	400	6	1040	-2.91	-0.33-0.74	n
Entrain Go Discost Discost <thdiscost< th=""> <thdiscost< th=""> <thdiscost< th=""> <th< td=""><td>El Ruiz</td><td>Colombia</td><td>Δ</td><td>82</td><td>955000</td><td>23400</td><td>19800</td><td>1760</td><td></td><td><10</td><td>86</td><td>-4.05</td><td>-1 31 1 05</td><td>и ае</td></th<></thdiscost<></thdiscost<></thdiscost<>	El Ruiz	Colombia	Δ	82	955000	23400	19800	1760		<10	86	-4.05	-1 31 1 05	и ае
Rajn Moyama, 84 S Kyushu R 320 993000 910 1800 2000 1200 42 28 4.55 -2.00 0.19 p Kuju Iwoyama, 84 S Kyushu R 165 992000 1800 2200 2600 1800 42 28 -4.55 -2.00 0.19 p Kuju Iwoyama, 84 S Kyushu R 165 992000 1800 2200 2000 1800 67.5 5 -5.30 -2.72 0.07 p Kudravy, 91 S Kurils BA 130 950000 2200 2000 1600 97.5 260 884 -3.05 -0.27 0.39 a,1 Mutnovsky funnel, 83 Karnchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky piper, 83 Karnchatka BA 272 98800 9820 290 1020 160 10 3.05 -1.76 -0.22 1 Mutnovsky piper, 83 Karnchatka	Kuju Iwoyama 61	S Kyushu	R	400	969000	4500	6900	14100	4800	970	1000	-2.99	-1 15 -0 31	n
Kapit Mojana, 64 S Kyushu R 165 992000 1800 2000 1800 600 67 5 5.30 -2.72 0.07 p Kudryavy, 91 S Kurils BA 130 992000 1800 2200 2600 1800 67 5 5.30 -2.72 0.07 p Manza karabuki NE Honshu A 94 992000 1460 220 6200 20 na 0.54 -6.26 -4.07 1.46 z Mutnovsky funnel, 83 Kamchatka BA 360 987000 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a,1 Mutnovsky funnel, 83 Kamchatka BA 272 988000 9820 290 102 160 10 350 -0.46 -0.55 1 Mutnovsky Notom, Kamchatka BA 97 978000 12400 2400 6310 550 9 110 -3.95 -1.76 -0.42 1 Nasu mugen NE Honshu A	Kuju Iwoyama, 84	S Kyushu	R	320	993000	910	1800	2800	1200	42	28	-4 55	-2 00 -0 19	P D
Kugu novyma, ov S Kushu R 130 950000 22000 2600 100 5 5.5.0 1.2 0.35 p Kudryavy, 91 S Kurils BA 130 950000 22500 23000 2600 110 5 2.0 -4.68 -2.11 0.95 m Manza karabuki NE Honshu A 94 992000 1460 220 6290 20 na 0.54 -6.26 -4.07 -1.46 z Mutnovsky funnel, 83 Kamchatka BA 360 987000 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a.1 Mutnovsky bottom, Kamchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky Nottom, Kamchatka BA 97 978000 12400 2400 6310 550 9 110 -3.95 -1.76 -0.42 1 Nasu mugen NE Honshu A 149	Kuju Iwoyama, 84	S Kyushu	R	165	992000	1800	2200	2600	1200	67	20	-5 30	-2 72 -0 07	P n
Manza karabuki NE Honshu A 94 992000 1460 220 6290 20 na 0.54 -6.26 -4.07 -1.46 z Mutnovsky funnel, 83 Kamchatka BA 360 987000 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a,I Mutnovsky funnel, 83 Kamchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky upper, 83 Kamchatka BA 272 988000 9820 290 1020 160 10 350 -3.45 -0.46 -0.55 1 Mutnovsky Nbottom, BA 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.46 -2.45 0.14 z z Nasu mugen NE Honshu A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.42 1 Vul	Kudrvavy 91	S Kurils	RA RA	130	950000	22500	23000	2600	110	5	20	-4.68	-2 11 0.95	P m
Mutnovsky funnel, 83 Kamchatka BA 360 987000 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a.l Mutnovsky S bottom, Kamchatka BA 106 987000 5190 4020 1640 975 260 884 -3.05 -0.27 0.39 a.l Mutnovsky S bottom, Kamchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky N bottom, BA 272 988000 9820 290 1020 160 10 350 -3.45 -0.46 -0.55 1 Mutnovsky N bottom, A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 100 9	Manza karabuki	NE Honshu	Δ	94	992000	1460	22000	6290	20	na	0.54	-6.26	-4.07 -1.46	7
Mutnovsky Shotton, Star 500 50000 5100 4020 1040 5100 604 5100 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 6021 6031 500 6021 6031 6021 6031 500 6021 6021 6021 6031 500 6021<	Mutnovsky funnel 83	Kamchatka	BA	360	987000	5190	4020	1640	975	260	884	-3.05	-0.27 0.39	2 a 1
83 Kamchatka BA 106 981000 13585 1045 2740 60 1 106 -3.97 -1.41 -0.42 1 Mutnovsky upper, 83 Kamchatka BA 272 988000 9820 290 1020 160 10 350 -3.45 -0.46 -0.55 1 83 Kamchatka BA 97 978000 12400 2400 6310 550 9 110 -3.95 -1.76 -0.42 1 Nasu mugen NE Honshu A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106	Mutnovsky S bottom,	Ivamenatka	DA	500	207000	5170	4020	1040	715	200	004	5.05	0.27 0.59	и,1
Mutnovsky upper, 83 Kamchatka BA 272 988000 9820 290 1020 160 10 350 -3.45 -0.46 -0.55 1 83 Kamchatka BA 97 978000 12400 2400 6310 550 9 110 -3.95 -1.76 -0.42 1 Nasu mugen NE Honshu A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano beach Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 106<	83	Kamchatka	BA	106	981000	13585	1045	2740	60	1	106	-3.97	-1.41 -0.42	1
Muthovsky N bottom, 83 Kamchatka BA 97 978000 12400 2400 6310 550 9 110 -3.95 -1.76 -0.42 1 Nasu mugen NE Honshu A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano beach Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 106 951000 4420 1820 2080 60 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A,	Mutnovsky upper, 83	Kamchatka	BA	272	988000	9820	290	1020	160	10	350	-3.45	-0.46 -0.55	1
Nasu mugen NE Honshu A 149 996000 2980 120 520 60 na 26 -4.58 -1.30 -0.64 z Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.64 z Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano beach Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 197 947000 40600 3760 840 7260 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d Zao NE Honshu A, D <t< td=""><td>83</td><td>, Kamchatka</td><td>BA</td><td>97</td><td>978000</td><td>12400</td><td>2400</td><td>6310</td><td>550</td><td>9</td><td>110</td><td>-3.95</td><td>-1.76-0.42</td><td>1</td></t<>	83	, Kamchatka	BA	97	978000	12400	2400	6310	550	9	110	-3.95	-1.76-0.42	1
Showa Shinsan SE Hokkaido D 336 998900 420 34 45 260 na 340 -3.47 0.88 -0.12 c Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano crater Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 197 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, D </td <td>Nasu mugen</td> <td>NE Honshu</td> <td>A</td> <td>149</td> <td>996000</td> <td>2980</td> <td>120</td> <td>520</td> <td>60</td> <td>na</td> <td>26</td> <td>-4.58</td> <td>-1.30-0.64</td> <td>7</td>	Nasu mugen	NE Honshu	A	149	996000	2980	120	520	60	na	26	-4.58	-1.30-0.64	7
Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano crater Italy TB 285 918000 69000 2950 2850 6700 na 10 -4.96 -2.45 0.01 j Vulcano beach Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 197 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, D<	Showa Shinsan	SE Hokkaido	D	336	998900	420	34	45	260	na	340	-3 47	0.88-0.12	c
Vulcano beach Italy TB 100 910760 86430 nd 1920 nd nd 94 -3.99 -1.31 na q White Island, 84 New Zealand D 197 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, D 96 977000 19800 2400 120 nd na 4.3 -5.36 -1.45 1.30 z Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A	Vulcano crater	Italy	TB	285	918000	69000	2950	2850	6700	na	10	-4.96	-2.45 0.01	i
White Island, 84 New Zealand D 197 947000 40600 3760 840 7260 na 21 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.65 -1.60 0.65 c,d White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, D 96 977000 19800 2400 120 nd na 4.3 -5.36 -1.45 1.30 z Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java <	Vulcano beach	Italy	TB	100	910760	86430	nd	1920	nd	nd	94	-3.99	-1.31 na	a
White Island, 84 New Zealand D 106 951000 44200 1820 2080 60 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 19 -4.70 -2.04 -0.06 c,d White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 -0.81 d Zao NE Honshu A, D 96 977000 19800 2400 120 nd na 4.3 -5.36 -1.45 1.30 z Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A 104 942000 54000 nd 2440 87 na 945 -3.00 -0.41 na c Krafla Iceland B <td< td=""><td>White Island, 84</td><td>New Zealand</td><td>D</td><td>197</td><td>947000</td><td>40600</td><td>3760</td><td>840</td><td>7260</td><td>na</td><td>21</td><td>-4 65</td><td>-1.60.0.65</td><td>r c.d</td></td<>	White Island, 84	New Zealand	D	197	947000	40600	3760	840	7260	na	21	-4 65	-1.60.0.65	r c.d
White Island, 84 New Zealand D 100 960000 36300 270 1730 260 na 122 -3.90 -1.15 0.81 d Zao NE Honshu A, D 96 977000 19800 2400 120 nd na 4.3 -5.36 -1.45 1.30 z Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A 104 942000 54000 nd 350 na 945 -3.00 -0.41 na c Krafla Iceland B 300 985000 13150 nd 350 nd na 360 -3.44 0.01 na c Krafla Iceland B 300 985000 -3700 nd 1350 nd 360 -3.44 0.01 na c	White Island, 84	New Zealand	D	106	951000	44200	1820	2080	60	na	19	-4 70	-2 04 -0 06	c d
Zao NE Honshu A, D 96 977000 19800 2400 120 nd na 4.3 -5.36 -1.45 1.30 z Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A 104 942000 54000 nd 350 nd a 360 -3.44 0.01 na c Krafla Iceland B 300 985000 13150 nd 350 nd a 360 -3.44 0.01 na c	White Island, 84	New Zealand	D	100	960000	36300	270	1730	260	na	122	-3.90	-1.15-0.81	d
Geothermal Campi Flegrei Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A 104 942000 54000 nd 2440 87 na 945 -3.00 -0.41 na c Krafla Iceland B 300 985000 13150 nd 350 nd na 360 -3.44 0.01 na c	Zao	NE Honshu	A.D	96	977000	19800	2400	120	nd	na	4.3	-5.36	-1.45 1.30	z
Conternal Italy 155 845000 153000 na 1780 31 na 226 -3.57 -0.90 na c Darajat Java A 104 942000 54000 nd 2440 87 na 945 -3.00 -0.41 na c Krafla Iceland B 300 985000 13150 nd 350 nd na 360 -3.44 0.01 na c	Geothermal		, =											
Darajat Java A 104 942000 54000 nd 1760 51 nd 220 -5.57 -0.90 nd c Darajat Java A 104 942000 54000 nd 2440 87 na 945 -3.00 -0.41 na c Krafla Iceland B 300 985000 13150 nd 350 nd na 360 -3.44 0.01 na c Krafla Iceland A 102 95600 27000 -1 1200 2.07 0.11	Campi Elegraj	Italy		155	8/15000	153000	ne	1780	21	ne	226	_3 57	-0.00 ====	C
Sarajar Sarajar <t< td=""><td>Darajat</td><td>Tava</td><td>Δ</td><td>104</td><td>942000</td><td>54000</td><td>nd</td><td>2440</td><td>91 87</td><td>no</td><td>0/15</td><td>-3.00</td><td>-0.41 no</td><td>c</td></t<>	Darajat	Tava	Δ	104	942000	54000	nd	2440	91 87	no	0/15	-3.00	-0.41 no	c
Manual D Job (J0000) J1000 J1000 J000 J0000 J0000 J0000	Krafla	Iceland	R	300	985000	13150	nd	350	o/ nd	no	360	-3.44	0.01 no	c
Ketetani New Zealand A 136 956400 3/800 nd 1300 nd na 1020 -2.97 -0.11 na c	Ketetahi	New Zealand	A	136	956400	37800	nd	1300	nd	na	1020	-2.97	-0.11 na	c

System	Location	Rock	T (°C)	H_2O^2	CO_2^2	SO_2^2	H_2S^2	HCl ² F	IF ²	H_2^2	R., ³	R_{e}^{4}	R ⁵ I	Refs ⁷
Broadlands well 25	New Zealand	R/D	290	994900	4890	nd	47	nd	nd	9	-2.88	0.23	na	r
Kawerau well 19	New Zealand	R, A	275	998300	1560	nd	65	nd	nd	5	-2.93	-0.10	na	s
Mokai well 6	New Zealand	R	319	998900	1015	nd	43	nd	nd	1.1	-4.21	-0.76	na	s
Rotokawa well 6	New Zealand	R	301	987600	2120	nd	120	nd	nd	9	-3.04	-0.22	na	s
Waiotapu well 7	New Zealand	R	242	999000	920	nd	53	nd	nd	5	-2.47	0.13	na	t
Wairakei well 72	New Zealand	R	260	999800	250	nd	10	nd	nd	0.7	-3.57	-0.08	na	r

¹ Abbreviations for igneous rocks: A = andesite, AB = alkali basalt, B = basalt, BA = basaltic andesite, D = dacite, GD = granodiorite, M = monzonite, QL = quartz latite, QM = quartz monzonite, R = rhyolite, TA = trachyandesite, TB = trachybasalt.

² µmols/mol

 3 R_H = log (XH₂ / XH₂O), where X = analytical mole fraction

 ${}^{4}R_{s} = \log (XH_{2} / XH_{2}S)$, where X = analytical mole fraction

 ${}^{5}R_{SH} = \log (XSO_2 / XH_2S)$, where X = analytical mole fraction

⁶ composition recalculated ("restored") by original author

⁷ References: a, Giggenbach (1996); b, Giggenbach and LeGuern (1976); c, Giggenbach et al. (1986); d, Giggenbach (1987); e. Giggenbach et al. (1990); f, Giggenbach and Matsuo (1991); g, Martini (1993), h, Menyailov et al. (1986); i, Shinohara et al. (1993); j, Mizutani and Suguira (1982); k, Taran et al. (1991); 1, Taran et al. (1992); m, Taran et al. (1995); n, Taran et al. (1998); o, LeGuern (1988); p, Mizutani et al. (1986); q, Chiodini et al. (1995); r, Giggenbach (1980); s, Giggenbach (1995); t, Hedenquist and Browne (1989); u, Ohba et al. (1994); v, Menyailov et al. (1984); w, Allard (1983); x, Symonds et al. (1990); y, Poorter et al. (1991), z, Kiyosu (1983)

Note: vapors from magmatic hydrothermal systems interact with acidic fluid; geothermal vapors derive from near-neutral pH fluid. R_H and R_S values for geothermal wells calculated using vapor-liquid distribution coefficients (Giggenbach, 1980) to allow comparison of total well discharge compositions to vapors discharged from fumaroles.

na, not available; nd, not detected

Table 4: Summary of alteration and sulfide mineral assemblages in subalkaline porphyry-Cu¹, porphyry-related base-metal vein², and epithermal Au-Ag deposits.

Porphyry Cu Left to r	ight: <i>time toward p</i>	resent	and/or	decrea	sing dept	h and/or a	distance increasin	g from center ³					
examples	early age ³ , higher	·Т	inter	mediat	e age ³	late ve	eins ³ , lower T,	references					
· · · · ·					0								
Type 1 porphyry: major mt-dg-bn in highest grade Cu ore													
Batu Hijau;	early mt (B, M)	bn+c	ep, ep	cp+p	y, mt	py+bn,	sl+gl+tn (CC,	Clode et al., 1999;					
Panguna;	⁴ ; <u>mt+dg+bn,</u>	(K)		(or h	m)+cp	CS)		Eastoe, 1978; Proffett,					
Yerington mine	<u>mt+bn+cp⁵</u>			(CS,	QSP)			1979					
	<u>(K)</u>												
Type 2 porphyry: majo	or mt-bn-cp in high	est gra	de Cu c	ore									
Bajo de l'Alumbrera;	early mt (M,	bn+c	сp	cp+p	y,	py+gl+	sl (P); py+en	Ulrich & Heinrich, 2001;					
Dos Pobres;	K); <u>mt+bn,</u>	(CC))	hm(c	or	(AA)		Langton & Williams,					
Rosario Porphyry	<u>mt+bn+cp (K)</u>			mt)+	cp (CS,	Rosa	rio vein	1982; Dick et al., 1994					
(under Rosario vein)				QSP)								
Type 3 porphyry: major to moderate dg-bn or bn-cp, mt absent or minor in highest grade Cu ore													
Bingham;	early mt (M,	<u>cp+j</u>	<u>oy,</u>	cp+p	y,	(po+cp	at Morrison);	John, 1978; Redmond et					
Chuquicamata;	K), rare po+cp;	mt+	<u>cp+p</u>	cp+p	y+bn	others:	py+bn+en,	al., 2001; Ossandon et al., 2001: Gustafson & Hunt					
El Salvador;	dg+bn+cp;	<u>y</u> ° (F	<u>8)</u>	(QSI	P); later	py+en-	+cv (AA);	1975: Gibbins 2000.					
Grasberg;	<u>bn+cp (EDM°</u> ,			cp+p	y+tn+	latter	is ore at	Pollard & Taylor, 2002:					
Morrison	<u>K)</u>			sl (14	A)	Chuo	uicamata	Carson & Jambor, 1976					
Type 4 nornhyry, mai	or cn-ny minor cn-	hn or i	nt_cn_n	v in hi	ahest arad	le Cu ore							
Butte ⁷ Magma	early mt (M	bn_c	n cn	y m mg cn⊥r	w	$\frac{10 \text{ Cu OIC}}{\text{pv}}$	P): $cp \pm py(C)$	Manske and Paul. 2002:					
Pornhyry (deen	B): $bn \pm cn$	cn+i	р, <u>ср.</u> ъv (К	$\frac{cp+1}{(OS)}$	P CAS	cn+nv-	r = 0, cp + py (C),	West and Aiken, 1982:					
lateral to Magma	$c_{p+p_{y}}$ (FDM ⁷	$\frac{cp_1}{R}$	<u>y (11,</u>	$\frac{(0)}{S(1)}$, CAD,	ore at	Butte &	Preece and Beane, 1982;					
Vein): Sierrita.	K B)	<u>D</u>]		<u>501</u>		Magm	a.	Graybeal, 1982;					
Feberanza: Silver	к, b)					nv±ce	a. Lda±hn (AA)•	Hezarkhani and					
Bell: Sungun						nv+hn	+cn (OSP)	Williams-Jones, 1998;					
Den, Sungun						pyron							
Porphyry-Cu Related	Base-Metal Veins ²	, very	high to	interm	ediate sul	fidation s	states						
examples	central or deep	;	int	ermed	iate ³	dis	stal or shallow ³	references					
Butte Main Stage,	py, cc, en, cv; py	, cc,	py, br	n, cc,	py, cp, t	n, (bn),	sl, (py, gl, rho);	Sales and Meyer,					
(superimposed on	en		en; re	licts	relicts o	f sl	rho, (sl, py, gl);	1949; Meyer et al.,					
porphyry-style ore)			of sl				Ag, (gl, sl, rho,	1968; Proffett, 1979					
							py)						
Chuqicamata Main	Early: py, cp, bn;	py,			py+cp+l	bn; py,	tn	Lopez, 1939;					
Stage,	bn, dg, (en); <u>Late</u>	: ру,			bn, dg, ((en);		Ossandón et al., 2001					
(superimposed on	dg, cv, (en); en, (py,			py, dg, d	cv, (en)							
early porph-style bn-	sl); en, sl, cv, dg;	cv,											
cp, mt-cp)	dg, (py, hm)												
	<i>.</i>							11					
Kosario vein,	py-cp-(tn), py-bn	-			cp, cp-p	У	ру	Hunt, 1985; Dick et					
Collahuasi (on top of	(en), cc-dg-cv							al., 1994, Clark et al., 1998					
kosario porphyry)								1770					
Epithermal Au-Age lef	ft to right: <i>time tow</i>	ard pr	esent an	nd/or d	ecreasino	depth an	d/or distance incr	easing from center					
examples	alteration		sulfide	e temn	oral seque	ence	variation	references					
r				p									
High-Sulfidation epith	ermal: py-en veins	, AA I	nalos, C	u-rich.	some aff	iliated wi	th porphyry Cu de	posits					
Lepanto (over porphyry), VQ-AA en	+lz+p	y cp	+tn+sl	+g late	e Au te	w/ time, or H	Iedenquist et al., 1998;					

Jannas et al., 1999; El Indio, Goldfield QSP, l ±py or Ag sf on margin Ransome, 1909 ру Losada-Calderón and La Mexicana, Lahóca, AA-VQ py+tn+cp tn±en±fm, w/ dec py+en+ McPhail, 1996; Gatter et al., Summitville (all over depth or ±Au Au+py fm, Au te ру

porphyry)			or Au+ba	time	1999; Stoffregen, 1987	
Intermediate-Sulfidation	epithermal	: base-metal su	ılfide & Ag-ricl	n veins		
Pachuca, Fresnillo,	adul,	sl+gl+py+	Ag sf, ±cp,	sl FeS 1-	w/ time or	Geyne et al., 1963; Gemmell
Creede, Comstock Lode	QSP,	cp+tt	Au	25 mole%	distance	et al., 1988; Barton et al.,
	IA, P					1977
Low-Sulfidation epitherm	al: sulfide	poor, Au (Ag)	bonanza veins			
Sleeper, Midas,	IA	py-mc	py+Au±na	late or	w/ time,	Saunders, 1994; John et al.,
McLaughlin, Hishikari			Ag sf, ch	shallow	multiple	1999; Sherlock et al., 1995;
-			(sl, tt, cp)	op, rl, cn	veins	Izawa et al., 1990

¹ in porphyry-type deposits, the position of molybdenite is not included in this table, although in most deposits it post-dates the main stage of veinlet copper ores and predates the late base-metal veins

² additional examples and summaries of these deposits or ore-styles can be found in Einaudi (1982) and Bartos (1989)

³ Asemblages listed are representative of the types of deposits. but not every deposit within a type contains all of the assemblages listed for that type. Only time-space diagrams can accurately portray evolution; in general, the early (high-T)-to-late(low-T) progression summarized here is from the perspective of the highly mineralized zone, but "late" or "distal" assemblages may be forming at the same time as "early" or "central" assemblages on the system scale. Also, early or high-temperature assemblages in porphyries commonly are repeated many times; high temperature = approx 550 to 400°C; intermediate temperature = approx 350°C; low temperature = approx 300 to 250°C; epithermal deposits form around 250 to 200°C.

⁴ alteration types temporally associated with the sulfide assemblages in the porphyry and vein deposits are listed in parentheses; see abbreviations below.

⁵ sulfide assemblages in bold represent the highest grade of copper in most of the porphyry deposits listed under a given type

⁶ in the deposits listed under Type 3, EDM is recognized only at Bingham

⁷ refers to pre-Main Stage at Butte; in the deposits listed under Type 4, EDM is recognized only at Butte

Sulfide abbreviations: asp = arsenopyrite; bn = bornite; cc = chalcocite; cn = cinnabar; cp = chalcopyrite; dg = digenite; en = enargite; fm = famatinite; gl = galena; hm = hematite; lz = luzonite; mc = marcasite; mt, = magnetite; na = naumannite; op = orpiment; po = pyrrhotite; py = pyrite; rl = realgar; sb = stibnite; sf = sulfosalts; sl = sphalerite; te = tellurides; tn = tennantite, tt = tetrahedrite

Gangue abbreviations: adul = adularia; ch = chalcedony; qz = quartz, rho = rhodocrosite; dol = dolomite

Alteration abbreviations: (AA) = "advanced argillic"; (AC) = albite-chlorite-calcite; (B) = biotite; (CAS) = chlorite-albite-calcite; (CC) = carbonate-chlorite-clay); (CS) = chlorite-sericite; (EDM) = "early dark micaceous", biotite, phengite, K-feldspar, andalusite, corundum; (IA) = "intermediate argillic", montmorillonite-kaolinite-illite, biotite chloritized, K-feldspar stable/metastable; (K) = "potassic", quartz, phlogopite, rutile, K-feldspar, anhydrite; (M) = amphibole-albite-magnetite; (P) = "propylitic", chlorite, epidote, calcite, albite; (QSP) = "phyllic, sericitic", quartz-sericite-pyite; (SCC) = sericite-chlorite-carbonate, (VQ) = vuggy quartz.

Table A. Reactions shown on R_H and R_S diagrams

R_H diagram (Fig. 3) $3 \text{ Fe}_2 \text{SiO}_4 + \text{O}_2 = 2 \text{ Fe}_3 \text{O}_4 + 3 \text{ SiO}_2$ fayalite mt qz $2 \text{ Ni} + \text{O}_2 = 2 \text{ NiO}$ 2 KFe₃AlSi₃O₁₀(OH)₂ + O₂ = 2 KAlSi₃O₈ + 2 Fe₃O₄ + 2 H₂O annite K feldspar mt $3 \text{ FeS} + \text{O}_2 = 3/2 \text{ FeS}_2 + 1/2 \text{ Fe}_3\text{O}_4$ ро ру mt $3/2 \text{ FeS} + \text{O}_2 = 1/2 \text{ Fe}_3\text{O}_4 + 3/2 \text{ S}$ mt sulfur ро $4/3 \text{ FeS}_2 + \text{O}_2 = 2/3 \text{ Fe}_2\text{O}_3 + 8/3 \text{ S}^{-1}$ hm sulfur ру $4 \text{ Fe}_3\text{O}_4 + \text{O}_2 = 6 \text{ Fe}_2\text{O}_3$ $\frac{1}{2}$ mt hm $20/3 \text{ CuFeS}_2 + \text{O}_2 = 4 \text{ FeS}_2 + 4/3 \text{ Cu}_5 \text{FeS}_4 + 2/3 \text{ Fe}_2 \text{O}_3^{-1}$ ср ру bn hm $1/2 \ H_2 S_{(aq)} + O_2 = 1/2 \ SO_4^{-2}{}_{(aq)} + H^{+-2}$ $1/2 H_2 S_{(aq)} + O_2 = 1/2 HSO_{4 (aq)} + 1/2 H^+$ 3 $2/3 \text{ H}_2\text{S}_{(\text{gas})} + \text{O}_2 = 2/3 \text{ SO}_{2(\text{gas})} + 2/3 \text{ H}_2\text{O}^{-4}$ R_s diagram (Fig. 5) $Fe + H_2S_{(gas)} = FeS + H_2$ iron ро $2 \operatorname{Fe}_2 \operatorname{SiO}_4 + \operatorname{H}_2 \operatorname{S}_{(\text{gas})} = \operatorname{Fe}_3 \operatorname{O}_4 + \operatorname{FeS} + 2 \operatorname{SiO}_2 + \operatorname{H}_2$ fayalite mt ро qtz $\mathrm{FeS} + \mathrm{H}_2\mathrm{S}_{(\mathrm{gas})} = \mathrm{FeS}_2 + \mathrm{H}_2$ ру ро $3 \text{ Fe}_3\text{O}_4 + 2 \text{ H}_2\text{S}_{(\text{gas})} = 4 \text{ Fe}_2\text{O}_3 + \text{Fe}\text{S}_2 + 2 \text{ H}_2$ mt hm py $5 \text{ CuFeS}_2 + 2 \text{ H}_2\text{S}_{(\text{gas})} = \text{Cu}_5\text{FeS}_4 + 4 \text{ FeS}_2 + 2 \text{ H}_2$ ср bn ру $H_2S_{(gas)} = S + H2^{-5}$ Sulfur Calculated at 500 bars pressure, unless noted

¹ these reactions are adjusted in Figs. 2 - 8 to be compatible with magnetite + hematite equilibrium data from Myers and Eugster (1993)

² defines the predominance boundary for aqueous sulfur species at pH = 6 (this pH is approximately neutral in the epithermal temperature range) ³ defines the predominance boundary for aqueous sulfur species at pH = 3⁴ used to define the "sulfur-gas buffer" at 1, 30, and 1000 bars based on data in Giggenbach (1987)

⁵ defines saturation with respect to liquid sulfur, "sulfur condensation"

Mineral abbreviations as in Table 4.