Tracking metal pathways in magmas using volcanic gas fingerprints

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As well as gases that regulate climate over geological time, volcanoes emit prodigious quantities of metals into the atmosphere, where they have key roles as catalysts, pollutants and nutrients. Here we compare arc basaltic volcano metal emissions measurements to those from hotspot settings. As well as emitting higher fluxes of metals (similar to those building ore deposits), these arc emissions possess a distinct compositional fingerprint, particularly rich in tungsten, arsenic, thallium, antimony and lead compared to those from hotspots. We propose volcanic metal emissions are controlled by magmatic water content and redox: hydrous arc magmas that do not undergo sulfide saturation yield metal-rich, saline aqueous fluid; shallow degassing and resorption of late-stage sulfides feeds volcanic gases at Hawai'i and Iceland. While global arc magma chemistries vary significantly, our findings suggest that volcanic emissions in arcs have a distinct fingerprint compared to other settings. A shift in global volcanic metal emissions may have occurred in Earth's past as more oxidized, water-rich magmas became prevalent, influencing the surface environment.

An important, yet poorly quantified, part of the solid Earth metal cycle is the transport of dissolved metals by magmas, followed by their sequestration as crustal ore deposits or

outgassing into the surface environment during volcanic eruptions. Metals play key roles in a number of Earth processes. The flux of volcanically-mediated metals from submarine arc volcanoes into past euxinic oceans¹ may have been of key importance for the evolution of life. Transition metals outgassed by volcanoes catalyse the aqueous oxidation of sulfur². Magmatic aqueous fluid is an important medium for transporting metals to sites of ore deposits^{3,4}. With some notable exceptions, studies of volcanic metal emissions have been dominated by studies of condensates from low temperature fumaroles⁵, generated during shallow cooling of magmatic fluids. These fluids are generally metal-poor, owing to their low temperature (below most gaseous metal condensation temperatures) and low salinity at low pressures³. There are a number of datasets from the plumes of active basaltic volcanoes⁶⁻¹⁰ however, which show that these volcanoes are emitting fluxes of metals^{6,8,11,12} of similar scale to those from large industrial smelters¹³ and those building crustal ore deposits^{3,6}, making basaltic volcanoes a significant source of metals (and semi-metals) into the atmosphere and oceans. Studying basaltic volcanic systems, instead of their more evolved counterparts, brings advantages: high metal fluxes, safer sampling, a magma less affected by differentiation (and consequent metal fractionation) and a style of volcanism more representative of that prevalent on early Earth, possibly providing insights into metal fluxing into the early ocean and atmosphere. It is expected, from metal ratios in ore deposits¹⁴ and sulfides¹⁵ and metal systematics in glasses 16,17 that mid-ocean ridge, hotspot and arc basaltic volcanoes should have different assemblages of outgassing metals. This has been noted by previous studies^{9,18}, but a systematic explanation for these differences has been lacking, until now.

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Metal outputs of basaltic volcanoes into the atmosphere

- 49 Figure 1 presents a compilation of volcanic gas and aerosol metal data^{6-10,12} (Supplementary
- Material) acquired at six active, basaltic volcanoes in a range of tectonic settings. Cu, Tl, Cd,

Zn, Pb, Se and Sn fluxes into the atmosphere can reach 10^3 - 10^4 kg/day at basaltic arc volcanoes and fluxes of a number of other metals (e.g., In) and semi-metals (Bi, As) can exceed 100 kg/day (figure 1A), similar to metal fluxes involved in crustal ore deposit development³. Metal fluxes from hotspot volcanoes Kīlauea (Hawai'i, USA; during the 2008-2018 summit eruption) and Holuhraun (Iceland; during the 2014-2015 eruption) are lower, not exceeding 100 kg/day for any metal/semi-metal species (although upper error limits for Zn and Se exceed this). Further, the data in figure 1A indicate that the metal composition of volcanic emissions differs systematically between arc and hotspot settings. The emissions from Kīlauea and Holuhraun aerosols are dominated by Se, Cd, Te, Pb, Cu and Zn and there are markedly lower fluxes of W, In, As, Cs and Tl over arc volcanic aerosols. Arc volcanic aerosols, in contrast, have the highest fluxes of Pb, Cu, Zn, Tl, As, Sn, Se. The minor silicate contributions to most of the key elements In, As, Cs and Tl inferred from the data from Stromboli, Masaya, Etna and Ambrym (Supplementary Material) mean that a variable silicate ash fraction of the plumes is unlikely to account for these differences^{6,7,36}. Further, the plume metal compositions do not mirror the composition of the lavas (figure 1B), which suggests that the gas compositions are not merely controlled by the initial compositions of the magmas in the different tectonic settings, although it is important to note that another recent study documented enrichments in Sb, As, Tl, W and Pb in Manus Basin submarine arc basalts over MORB¹⁹. Arc lavas (here exemplified by Stromboli and Etna) are, overall, richer in metals such as Sb, U, Te, W, Cs, but are not appreciably richer in the other metal species, in particular those which are so abundant in the plume, e.g., Pb, Cu, Sn and Zn. In fact, Cu, Sn and Zn have similar concentrations in erupted lavas from all of the volcanoes, even when degassing is accounted for (dashed lines in figure 1B show pre-degassing metal concentrations). The absolute flux of metals for a particular volcano is controlled by magma degassing rate and style of volcanic activity, which would elevate or depress metal fluxes, but

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is not expected to change the distinctive patterns observed in the relative metal abundances shown in **figure 1A**. The ratios X/SO_2 , where X is a metal (or semi-metal), also show differences between settings (**Supplementary Material**), with X/SO_2 being generally lower for the hotspot volcanoes ($<1 \times 10^{-4}$ for most metals and semi-metals, with many $<1 \times 10^{-6}$), extending up to 1×10^{-3} for Pb, Zn and Cu for the arc volcanoes.

We propose that the metals measured in these volcanic plumes were transported in the exsolved magmatic volatile phase prior to and during eruption. A plot of enrichment factors (see **methods**) demonstrates the tendency of the metals to exist in the volcanic plume gas and aerosol phase over the silicate melt phase, relative to Cu (**figure 1C**). Enrichment factors reach overall higher values for the hotspot volcanic systems: greater than 10² for In, Bi and Cd (for both Kīlauea and Holuhraun) and As, Tl and Pb (for Holuhraun) and reaching 10⁴-10⁵ for Te and Se, the most volatile semi-metals. For the arc plumes, enrichment factors are in general lower but follow the same overall trends (**figure 1C**). Te, Se, Cd and Bi exhibit much lower enrichment factors referenced to Cu in high-temperature these arc volcanic gases over hotspot volcanic gases, despite similar or higher concentrations of these metals in the melt prior to degassing (**figure 1B**). Kinetic effects may be important under certain conditions; it has been shown that significant variation in, e.g., Tl/Pb is possible in silicate melt during degassing, due to diffusive fractionation²⁰. These effects are poorly understood yet unlikely to explain the order of magnitude differences observed in Tl/Pb here (**figure 1A**).

Metal pathways in magmas prior to eruption

Explaining the systematic differences in trace metal emissions between these arc and hotspot volcanoes requires consideration of the processes that partition metals during magma ascent and eruption. In magmatic systems in the crust, metals partition between silicate melt, an

aqueous fluid phase (which may be a hypercritical phase or an aqueous liquid at mid-crustal pressures, or a vapor phase at low pressures, comprising CO₂, water, sulfur and halogen species), and a sulfide phase (liquid sulfide, or monosulfide solid solution, depending on temperature), should they be present. Other phases, such as silicates, oxides and phosphates, may also take up metals, but are of subordinate importance. Both chalcophile and siderophile metals partition strongly into both aqueous fluid and sulfide phases over silicate melt over a range of temperatures and pressures. Aqueous fluid-melt partition coefficients are constant regardless of whether a sulfide phase is present; therefore the presence of sulfide sequesters metals such that both the melt and aqueous fluid phases become metal-depleted. Many experimental determinations of metal partitioning have been under conditions not directly relevant to volatile-rich magmas in the crust, e.g., at mantle pressures ^{21,22}. Little attention has been paid to those metals most prevalent in volcanic gases, such as Zn, Sn, Pb and Tl. Experiments at ~200 MPa on andesites and basalts with co-existing sulfide, silicate melt and aqueous fluid²³⁻²⁶ show that aqueous fluid-melt partitioning is dependent on the composition of the aqueous fluid and on redox. High Au concentrations in silicate melt are promoted by conditions just below the sulfide-sulfate transition (which is gradual and occurs ~NNO+0.5)²³. The fluid-melt partitioning of Au is enhanced at fO₂ <NNO+0.5 and suppressed at higher fO₂ and when SO₂ dominates the aqueous fluid phase, in contrast to the partitioning of Cu, which is much less dependent on S speciation26. A recent set of experiments on aqueous fluid-melt partitioning of a wide range of metals in mafic magma, in the presence of S- and Cl-bearing aqueous fluid (Supplementary Material Table 2), showed that Cu, Se, Te and Cd are most volatile, followed by Zn, As, Ag, Sb, Cs, W, Tl, and Bi²⁴. This is broadly consistent with the volcanic emissions measurements shown in Figure 1C although W, Cs and Zn show lower enrichment factors compared to Cu than these experiments might predict.

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To understand the balance between the silicate melt-aqueous fluid and silicate melt-sulfide partitioning further, a compilation of sulfide-silicate melt, and vapor-melt partition coefficients for metals is shown in figure 2 (citations and experimental conditions in Supplementary Material). Also shown in figure 2 are the calculated emanation coefficients (see methods) of the metals (and semi-metals) in the gas/aerosol plumes of the basaltic volcanoes, plotted in order of the Holuhraun data. These data show that the hotspot basaltic volcanic emissions plumes (Kīlauea and Holuhraun) are particularly poor in those metals that partition more strongly into aqueous saline fluid than the sulfide phase, such as U, Cs, W, Zn, Mo (left hand side of the plot). The plumes from Kīlauea and Holuhraun are particularly rich (with emanation coefficients of 1-100%) in those metals and semi-metals which partition strongly into sulfide over silicate melt (with partition coefficients of >100; on the right-hand side of the plot), such as Cd, Se, Te (no data exists for Au in the gas plumes of Kīlauea or Holuhraun; Figure 2). In contrast, the data for these arc gases show that all metals have an emanation coefficient of >0.1% (with the exception of U for Stromboli). These arc plumes are richer in U, Cs, W, Cu than hotspot volcanic plumes; these elements are associated with significantly higher aqueous fluid-melt partition coefficients (up to 10¹–10²; note the aqueous fluid-melt partition coefficient for Sn is for rhyolite melt²⁷ and the partition coefficient increases with the chlorine content of the aqueous fluid²⁷) relative to their sulfide-silicate melt partition coefficients (<10, with the exception of Cu).

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Effect of the timing of aqueous fluid and sulfide saturation

Arc basalts contain, on average, 4 wt% H_2O^{29} , and 0.5–2.0 wt% CO_2^{30} . Aqueous fluid saturation is expected to occur in the mid-crust, into which metals will partition, strongly in some cases (**figure 2**). Hotspot basalts, in contrast, contain 0.5-1.0 wt% H_2O^{31} and 0.5 to 1.0

wt% CO_2^{32} , which results in the exsolved fluid becoming water-rich only at shallower depths. Importantly, the fluid phase will be strongly CO_2 -dominated until the magma reaches low pressures (<100 MPa) during magma ascent, when H_2O and sulfur (as H_2S or SO_2) partition into the aqueous vapor phase^{33,34}. The metal-carrying capacity of CO_2 -rich aqueous fluid has been shown to be poor³⁵.

Models of sulfide saturation^{36,37} show that while mid-ocean ridge basalts are likely saturated in sulfide on eruption, more oxidized basalts, where a greater proportion of sulfur is present as sulfate^{38,39}, must fractionate to a greater degree to achieve sulfide saturation. Some basalts may not saturate in sulfide at all^{39,40}. Sulfides are not commonly present (as quenched sulfide liquid or sulfide minerals), either in the groundmass or as inclusions, in the erupted products of Stromboli, Etna and Ambrym⁴¹. In contrast 2014–2015 Holuhraun Icelandic melts contain abundant sulfides on eruption; as do the matrix glass and melt inclusions of Laki⁴². At Kīlauea, basaltic tephra contain sulfides as inclusions in olivine. Sulfur systematics in the matrix glasses of basalts from Iceland and Kīlauea are consistent with sulfide saturation on or shortly before eruption⁴³.

If sulfide saturates prior to an aqueous fluid in the deep crust (**figure 3**, **middle**), dense sulfides may be sequestered into cumulates, thus removing metals from the silicate-melt aqueous fluid system and generating chalcophile-poor melt (and aqueous fluid) upon further ascent and eruption of the magma; this has been suggested as an explanation for the lack of ore deposits in relatively water-poor 'barren' arc magmatic systems⁴⁴. A chalcophile-poor arc volcanic gas is not a feature of the datasets presented here. If, however, an aqueous fluid phase forms before sulfide saturation, which is likely for a water-rich arc basalt (**figure 3**, **right**), metals will partition strongly into the saline aqueous fluid⁴⁵. If the silicate melt does

not saturate with sulfide¹⁶, the aqueous fluid will remain metal-rich and will be outgassed during magma ascent and eruption at the surface with the metals in proportions controlled by their respective fluid-melt partition coefficients. If the melts are very chlorine-rich, a brine phase may form at low pressure, unmixing, along with a low density vapor, from a supercritical fluid described by the solvus in the NaCl-H₂O system³. Brine formation may fractionate metals; however these effects are minimal for basalts degassing near atmospheric pressure, which are relatively poor in chlorine (compared to rhyolites) and are likely to transport a saline aqueous fluid phase up to the surface (including any minor brine phase)³. The vapor phase in the sulfide-free case will then be rich in a range of fluid-mobile metals/semi-metals, just as observed in the arc volcanic gas data presented here (**figure 1**). Note that primary melts in arc settings may also be variably enriched in fluid-mobile metals¹⁹, caused by fluid addition to the mantle wedge directly from the slab. However, the data presented here suggest that further partitioning into the aqueous fluid phase must also take place to explain the enrichment in fluid-mobile metals in the volcanic emissions over the silicate melt (**figures 1B, 2**).

If sulfide saturates after an aqueous fluid phase, the sulfides may play some role in metal partitioning during shallow eruption and degassing processes. The hotspot volcanic gases in **figure 1A** are rich in chalcophiles (e.g., Cd, Te, Se) and the emanation coefficients of the gases in these plumes broadly follows sulfide-silicate melt partition coefficients (**figure 2**), suggesting that metals are sourced from sulfides directly, either via a mantle melting process⁴⁶ or through the formation of a sulfide liquid droplet at the interface between a silicate melt and aqueous fluid bubble ^{47,48}, shown as "vapor-sulfide aggregates", in **Figure 3** (**left**). Sulfides may become unstable due to oxidation during attachment to a vapor bubble and may then be resorbed during exsolution of an aqueous vapor phase⁴⁹. In this way, metals that were bound

in sulfides¹⁷ are released to the silicate melt-vapor system, and then to the atmosphere. Congruent dissolution of sulfides within these bubble-sulfide aggregates has been proposed to explain the similarity of metal outgassing ratios to metal ratios in sulfides at some volcanoes^{47,50}.

Implications of systematic volcanic metal emissions variations

The four arc volcanoes presented here have a distinctive metal outgassing fingerprint, with higher concentrations of Cs, As, In, W and Tl over hotspot volcanoes (**Figure 1**). While global arc magma chemistries vary significantly, we propose that these characteristic metal enrichments will be a broad feature of global arc volcanic plumes due to their relatively oxidized nature, low degree of fractional crystallization and high magma water and chlorine contents. Together, these factors suppress sulfide saturation promoting partitioning of metals/semi-metal directly into aqueous/saline aqueous fluid rather than metal/semi-metal degassing being mediated by sulfides (**Figure 3**).

Saline aqueous fluids clearly play an important role in transporting large fluxes of metals from basaltic melts through the mid and upper crust, eventually either outgassing from volcanoes or being incorporated into ore deposits. Volcanic outgassing of metals and ore deposit formation may be effectively mutually exclusive in any particularly time period. In fact, the gases emitted from volcanoes are representative of the fluids that, under different conditions of magma supply or tectonic stress, might instead precipitate economically viable concentrations of metals and sulfur in shallow crustal porphyries. In arc magmatic systems, there is good evidence that mafic magmas of the type that we observe outgassing at the surface (figure 1) underplate felsic magmas in the crust⁵³. Recharging mafic magmas clearly have potential to supply extensive fluxes of volatiles to overlying felsic magmas via an

aqueous fluid phase which might migrate efficiently through crystal-rich, near-solidus magmas^{54,55}. Accumulation of aqueous fluid at the roof zones of such reservoirs may therefore trigger the shallow intrusion of stocks and then be the medium from which metals are precipitated in copper porphyry systems³, without the need to invoke reworking of magmatic sulfides to concentrate metals.

There may be broader implications of the important role of volcanoes in metal geochemical cycling. It has been proposed that arc basalts were not oxidized until the late Neoproterozoic, when extensive ocean bottom water sulfate led to subducted sulfate oxidizing the sub-arc mantle⁵⁶, which coincides with copper and gold porphyries becoming prominent in the geological record^{56,57}. Submarine arc volcanoes during the Archaean and much of the Proterozoic may hence have outgassed mixtures of metals very much like hotspot volcanoes today; with higher volcanic arc outgassing fluxes of W, Cs, As, Tl, Pb, Cu and Zn (similar to modern arc volcanoes) only becoming widespread in the earliest Phanerozoic, when shallow submarine arc vents may have become significant localized sources of these biologically important metals in the ocean⁵⁸.

Methods

Enrichment factors were calculated by normalizing gas plume metal concentrations by Cu concentrations, and lava metal compositions by Cu concentrations and then dividing the former by the latter. This normalization step is necessary because plume dilution is variable between datasets. Enrichment factor (EF) is then a quantity that is comparable between sources and is given by:

$$EF_{Cu} = \begin{bmatrix} X \\ Cu \end{bmatrix}_{gas} \begin{bmatrix} X \\ Cu \end{bmatrix}_{lava}$$
 (1)

We have used Cu as our normalizing element, rather than a more lithophile element such as a REE or Al, Mg, because the latter necessitates accounting far more accurately for the contamination of the samples by silicate material. Cu has been affected by contamination by ash, but only by up to 6%; see **supplementary material** for discussion). Cu is a poorly to moderately volatile element, which is reflected in the plot of enrichment factors; most metals plot at higher values than Cu. EF<1 means the element is less "volatile" than Cu; EF>1 means the element is more "volatile" than Cu. Ours is not the first study to use a moderately volatile element to normalize data to generate EF; Br has been used in several studies⁶.

The emanation coefficient, \mathcal{E}_x , is calculated as the percentage of the metal x that has degassed from the silicate melt, equal to:

$$\varepsilon_{x} = \frac{\left(c_{i} - c_{f}\right)}{c_{i}}$$
(2)

where c_i is the initial concentration of element x in the magma and c_f is the final concentration of element x in the post-eruptive lava (as originally defined by ⁵⁹, who used this relation to describe the degassing of radionuclides ²²²Rn, ²¹⁰Po, ²¹⁰Bi and ²¹⁰Pb).

We use two methods to calculate emanation coefficients shown in **Figure 2** (see Supplementary

Material Table 1). The emanation coefficient \mathcal{E}_x of these elements from the magma can be assessed

from their mean enrichment factor (EF, see equation 1 above) and the emanation coefficient of Pb

from molten basalt (which we assume to be constant at 0.01^{60}), as 6,8,61 :

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$$\varepsilon_{x}^{-1} = 1 + \left[\left(\frac{EF_{Pb}}{EF_{X}} \right) \left(1 - \varepsilon_{Pb} \right) \right]$$
 (3)

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We also calculate the concentration of metals and semi-metals in the silicate melt prior to degassing and then the emanation coefficient using equation 2 above. The amount of metal (or semi-metal) in the gas plume was "added" back into to the degassed lava composition by converting the mass X/SO2 ratio in the plume (Supplementary Material figure 1) to X/S and then multiplying by the mass of sulfur degassed (from melt inclusion and matrix glass sulfur concentrations 41 9 39,62). This amount, in ppm, was added to the degassed lava composition to estimate an "un-degassed" magma composition c_i . It is possible that sulfide saturation could occur prior to melt inclusion entrapment, which would cap the sulfur concentration and deplete chalcophile metal concentrations in the melt inclusions; these processes would introduce error on these estimates. The arc basalts show no evidence for sulfide saturation; there are no sulfide globules in the melt inclusions from Ambrym⁴¹, Stromboli and Etna³⁹, but Kīlauea and Holuhraun melt inclusions show occasional sulfide⁶², although it is not clear whether these formed pre- or post-entrapment. We assume that sulfur is dominantly present as SO₂ in the gas plume and H₂S is neglected. It is also possible that under some circumstances S degasses from a larger volume of magma than metals/semi metals due to its lower solubility at higher pressures or mediation of metal degassing by halogen species. In order to overcome the difficulties associated with using X/Sratios to calculate \mathcal{E}_{ω} , we repeat the calculation using X/Cl to calculate the pre-degassing metal (and semi-metal) concentrations for which data is available for most of the datasets used here (see **Supplementary Material**). We find there is good agreement between the \mathcal{E}_{v} calculated using X/S and X/Cl (see Supplementary Material).

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Data Availability Statement

The authors declare that the data supporting the findings of this study, and the citations detailing data sources, are available within the article and its supplementary information files.

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488		
489	Figur	es
490		
491	Figur	e 1: Metal systematics in the gas plumes of active basaltic volcanoes in a range of
492	settin	gs. A: Metal fluxes in kg/day, with uncertainties, from four arc (or "arc affinity")
493	volcar	noes (blue) and two hotspot volcanoes (red) 6,7,9,10,12,41 . X/SO_2 for each dataset are shown

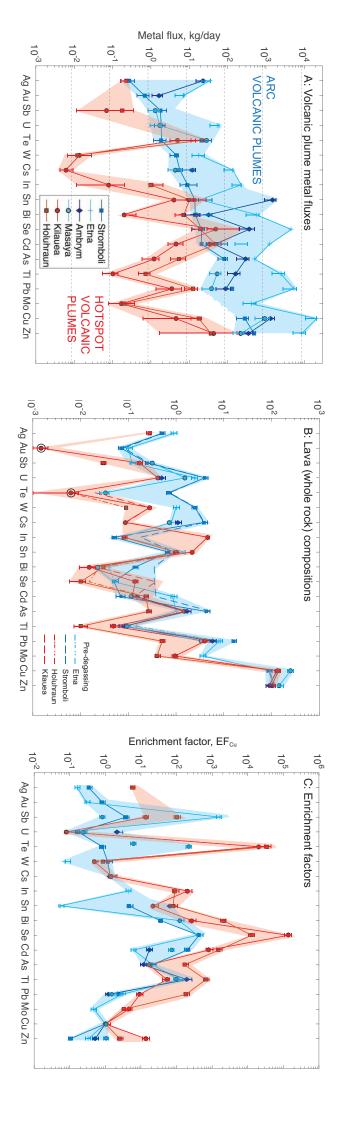
in **figure 1** of the **supplementary material**. B: Lava compositions (ppm) (a volcanic bomb for Masaya; and scoria for Stromboli and Holuhraun) for the eruptions in each setting, citations as above. Solid lines are degassed lava compositions; dashed lines show the metal concentration prior to degassing (see **supplementary material** for data and methods). Note that these dashed lines are only resolvable for some elements; for others the amount depleted by degassing is relatively small. C: Enrichment factors relative to Cu for each volcano (see Methods). Fields for each setting are shaded for ease of visual evaluation.

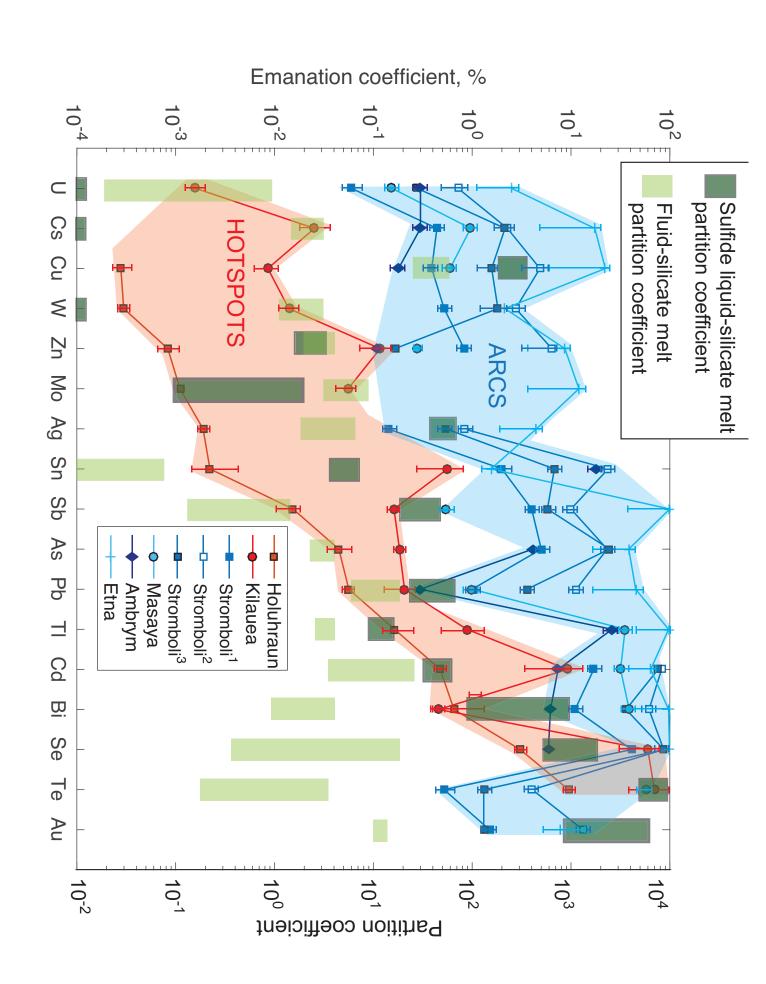
Figure 2: Volcanic gas data compared to metal partitioning between silicate melt, sulfide and aqueous fluid. Emanation coefficients for the range of metals and semi-metals are shown (see methods and Supplementary Material). Red shaded data are for volcanic gases from hotspot volcanoes Kīlauea and Holuhraun (Iceland); blue shaded data are for volcanic gases from arc basaltic eruptions Stromboli*, Etna, Masaya and Ambrym. Dark green rectangles are sulfide-silicate melt partition coefficients and light green, aqueous fluid-melt partition coefficients (see Supplementary Material for citations).

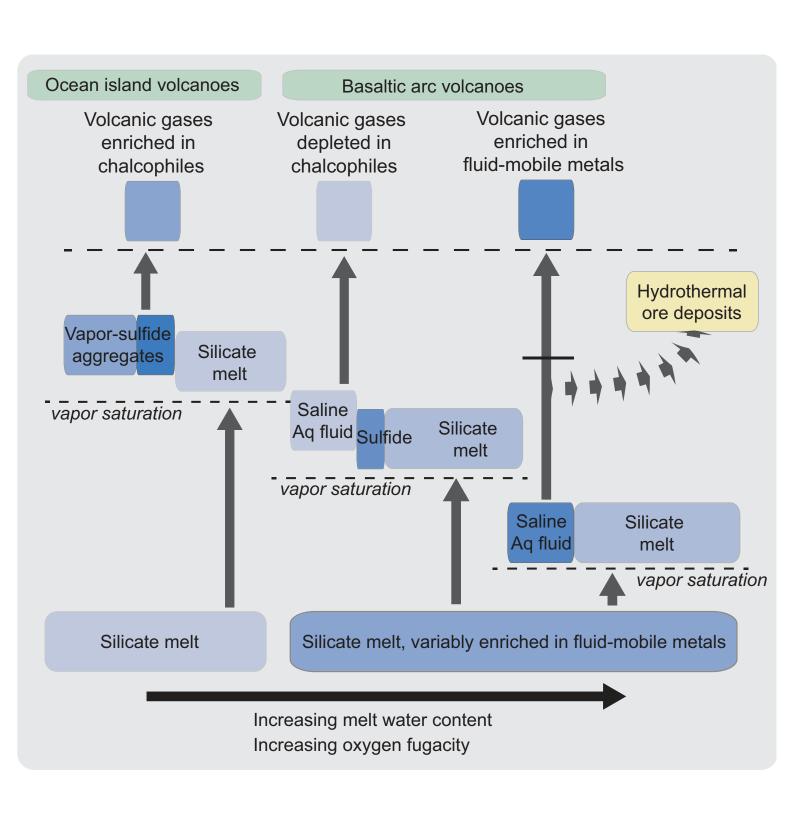
*Three emanation coefficients are shown for Stromboli (see Methods and Supplementary Material): 1. From⁶ calculated assuming $\varepsilon_{Pb} \sim 1\%^{59,60}$; 2. calculated using plume X/SO_2 and the mass of sulfur degassed. 3. calculated using plume X/Cl and the mass of chlorine degassed. Note that the X/SO_2 values for Etna aerosol (**Supplementary Table 1**) are high compared with other arc volcanoes in this dataset and with other data from Etna^{8,28}, which causes the emanation coefficients to be anomalously high when using this X/SO_2 ratio, combined with the melt inclusion sulfur contents (see **methods**), to reconstruct the metal contents of the melts.

Figure 3: Metal pathways through silicate melt, sulfide and aqueous fluid, and their impact on the metal composition of basaltic volcanic gas and aerosol. We envisage Kīlauea and Holuhraun to correspond to the far left case, where much of the magmatic water exsolves at low pressures in the crust, coinciding with late stage sulfide saturation. Arc basalts

- 520 correspond to the far right case, where deep magmatic aqueous fluid phase saturation occurs
- with no sulfide saturation, allowing metal-rich aqueous fluid to form.



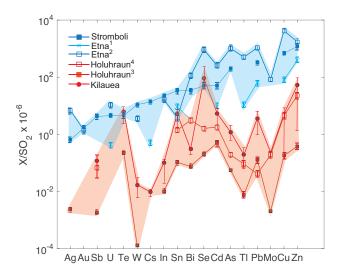




Supplementary Material

Tracking metal pathways in magmas using volcanic gas fingerprints

M. Edmonds, T. A. Mather, E. J. Liu



Supplementary figure 1: Ratios X/SO₂, in mg/kg, in the gas plumes volcanoes from arc (blue) and ocean island settings (red). Volcanoes are Stromboli (solid blue squares), Etna (light blue crosses¹ and blue open squares²), Kīlauea (red circles) and Holuhraun, Iceland (red squares³ and open squares⁴). Elements are ordered using the Stromboli dataset, as Figure 1 of the main paper. See table 1 for citations and errors on data.

Contribution of metals to the volcanic plume as silicate ash particles

Volcanic particulates sampled by filter packs in plumes are made up of both aerosols (aqueous liquids with dissolved acid gases into which metals partition from high temperature magmatic vapor, before being condensed upon quenching in the atmosphere) and also silicate particles, fragments of bubble walls typically launched into the plume by bubble bursting. The bulk analysis of the particulate fraction then inevitably results in some of the metal budget being sourced from the silicate particles rather than metals that have been volatilized from the magma forming the aerosol and gas phases. This effect is most significant for the less mobile, more lithophile elements, and less important for the more volatile metals, with high vapor-melt partition coefficients. Here we discuss the magnitude of this effect.

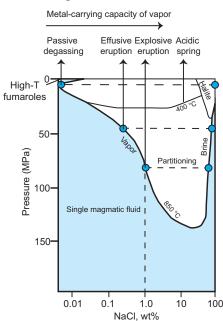
At Stromboli a minor silicate contribution was inferred for Cr, Cs, Cu, Rb and W (0-5%), but a significant one for Te and Zn (10%) and corrected accordingly³. For Etna, it has been noted that significant concentrations of Al, Si, Na, K, and REE are present in the plume, which must be sourced from silicate material². Elements such as Se, Cd and Sb, however, which are present only in trace amounts in silicate melts, are present in significant quantities in filter pack analyses (greater than more abundant lithophile trace elements such as Th, Sc), suggesting that these metals are volatile. The weight ash fraction (WAF) was calculated by these authors² using the abundance of REE. The calculated WAF is very low for Bi (0.1%), Sb (0.2%), Se and Tl (both 0.1%); slightly more significant for Pb (3%), As (2%) and Cu (6%) but significant for Mo, Zn, Te and W (33, 25, 59, 100% respectively), although Te is poorly constrained by its low abundance (close to detection, table 1) in lavas. There is abundant evidence from elsewhere that Te is highly volatile⁴. Another study used Th abundance to calculate the proportion contributed from ash and derived considerably lower estimates: zero for Tl, Bi, Cd; then 3% for Cu, 1% for Sn, 2% for Zn and 0.5% for Pb¹. At Ambrym negligible ash contributions were inferred for Ag, Au, Sn, Bi, Se, Cd, As, Tl, with a slightly larger contribution for Cs (0-15%), Pb (\sim 2%), Cu (0-17%) and Zn (7-40%)⁵. For Holuhraun the concentrations of Th and REE were used to estimate very low proportion of silicate ash (about 10 μg of ash per filter on average) and therefore the effect of ash contamination was neglected⁶. An ash correction was not calculated for the Kīlauea dataset.

Solid sulfides, oxides, halides, and oxyhydroxides might also be present in the plume, given the dramatic fall in solubility of gaseous metal chloride, sulfide and other species during the drop in temperature from the basalt liquidus temperature to ambient conditions. The presence of these phases may impact the estimation of silicate ash fraction because even nominally "immobile" high field strength elements like Sn, Th or REE may have significant volatility in the presence of abundant gaseous F, as is well known from studies of greisen-forming ore-bearing hydrothermal systems e.g.[7].

We conclude that while in some cases up to 40% of a metal or semi-metal species might be supplied from ash, in general the metals we plot here are highly volatile and their abundance is controlled dominantly by transport initially in the magmatic vapor phase and later (at and away from the vent) in liquid aerosol.

Metal-carrying capacity of magmatic vapor phase at low pressures

Experimental studies⁸ and thermodynamic modeling have shown that metals form stable

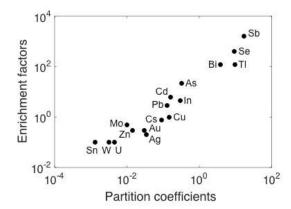


complexes with chloride, sulfate and fluoride in volcanic plumes⁹. Volcanic gas at the surface is the low density phase which is the result of separation of a brine phase from a saline decompresses supercritical fluid as and it (supplementary figure 2). Up to now, much of the literature on metals in volcanic environments has been gleaned from condensates^{10,11}, deposited during shallow cooling of magmatic fluids, which have little relevance to the question of how mafic volcanic systems outgas metals to the atmosphere¹². Little data exists for hot magmatic plumes outgassed directly from the magma column during eruptions, which are expected to contain a far higher load of metals dispersed into the atmosphere, either as gas or as sub micronaerosols, owing to their higher temperature and chloride content^{13,14}.

Supplementary figure 2: NaCl-H₂O phase diagram plotted in pressure-composition space, modified after¹⁵.

The solvus will shrink significantly at high temperatures, increasing the salinity and metal-carrying capacity of outgassing vapor. It is possible that supercritical metal-rich fluids may reach the surface, avoiding immiscibility. Until recently 16,17, it was assumed by some authors that low salinity volcanic gas would not have the capacity to dissolve large quantities of metals as chloride complexes, and the brine phase was therefore the favored ore-forming medium 18. **Figure 1** (main paper) shows that, on the contrary, volcanic gases are highly significant carriers of metals.

Table 1, below, shows the data used in this paper and its sources. We focus here on active basaltic volcanoes with data acquired using filter packs at the vent (*i.e.* not from fumaroles or lava flows, or as condensates and sublimates).



Supplementary Figure 3: Enrichment factors from **table 1** for elements in the plume of Etna (Italy), using Cu as a normalising element, plotted against "absolute" partition coefficients (defined as mass of element in the gas/mass of element in the melt phase) calculated by estimating a dilution factor for the plume¹².

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datasets presented in the main paper. Table 1: Metal/semi-metals contents of degassed lavas and gas/aerosol samples; X/SO2, metal fluxes, volatilities and enrichment factors for the

Erupted lava metal content, in ppm. Data sources: Stromboli Volcano (Italy)³, Holuhraun (Iceland)⁶, Etna (Italy)², Ambrym Volcano (Vanuatu)⁵, Kīlauea (Hawaii, USA), standard BHVO¹⁹ and Masaya Volcano (Nicaragua)²⁰. Error typically <10% for ICP-MS, IMA with internal

calibration using standards.

are typically 5-25%. Gas and aerosol metal data, in μg/m³, from Stromboli Volcano (Italy)³, Holuhraun (Iceland)⁶, Etna (Italy)¹, Ambrym Volcano (Vanuatu)⁵, Kīlauea Volcano (Hawaii, USA)¹⁴ and Masaya Volcano (Nicaragua)²⁰. Uncertainties are shown in brackets for each element within each dataset and

Flux of element X in kg per day, calculated using the mass ratio X/SO_2^* and the independently measured flux of SO_2 (data sources as above). The uncertainty on the flux measurement is 30-32%, propagated from errors on SO_2 flux (~30%) and errors on gas and aerosol measurements (5-10%).

* X/SO_2 , in mg/kg; errors on X/SO_2 are propagated from errors on analysis of S and X in gas/aerosol phase and including, where available²¹, the standard deviation of repeat analyses on the same day, amounting to 7-17% for the former and up to 95% for the latter²¹.

Emanation coefficient, ε_x , is calculated as the percentage of the metal x that has degassed from the silicate melt, equal to: $\varepsilon_x = (c_i - c_j) / c_i$

defined by²². The emanation coefficients given are estimated in a range of ways (which are indicated as footnotes for each sub-table). The emanation coefficient ε_x of these elements from the magma can be assessed from their mean enrichment factor (EF) and the emanation coefficient of Pb from molten basalt (which may be assumed to be constant at 0.01^{23}), as $t_{x}^{1,3,24}$: $t_{x}^{-1} = 1 + \left[\left(\frac{EF_{Pb}}{EF_X} \right)^{(1-\epsilon_{Pb})} \right]$. We also calculate the concentration of metals and where c_i is the initial concentration of element x in the magma and c_i is the final concentration of element x in the post-eruptive lava (as originally

are propagated from typical errors on the individual electron microprobe analyses of sulfur, which is typically 5%, and errors on X/SO₂. Total errors metal concentrations in the melt, using melt inclusion and matrix glass concentrations as shown in the footnotes. Errors on the emanation coefficients estimate a "un-degassed" magma composition c_i . For some datasets we also calculate emanation coefficients using X/Cl to estimate pre-degassing lava composition by converting the mass X/SO_2 ratio in the plume and then multiplying by the pre-eruptive melt concentration of sulfur (from melt inclusion concentrations, corrected for matrix glass sulfur concentration⁵ 6 25,26). This amount, in ppm, was added to the degassed lava composition to semi-metals in the silicate melt prior to degassing, c_i . The amount of metal (or semi-metal) in the gas plume was "added" back into to the degassed

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 EF_{Cu} is the enrichment factor relative to copper (Cu), equal to $||EF_{Cu}||_{g_{Cu}} = ||EF_{Cu}||_{g_{Cu}}$ for from 14-17% . Errors are propagated from the individual metal analyses

and range from 14-17%.

Note on propagation of errors: X/SO_2 , enrichment factors and emanation coefficients are calculated using the measured concentration data as defined above, with individual errors as defined in the paper sources (see tables below). Errors on these quantities are propagated from the error on the individual measurements, where the uncertainty on quantity z (which is a function of independent variables x and y e.g. z = xy, or z = x/y) is the sum

of the squares of the fractional errors, or error percentages, of x and y: $\frac{\delta z}{z} \sqrt{\left(\frac{\delta x}{x}\right)^2 + \left(\frac{\delta y}{y}\right)^2}$

Volcano	Stromboli								
	Lava,	Gas, aerosol,	Flux,	X/SO_2 ,	X/Cl,	Emanation coefficient	Emanation coefficient	Emanation coefficient	EF_{Cu}
Elements	ppm	μg/m³	kg/day	mg/kg	mg/kg	$\varepsilon (\%)^a$	$\varepsilon (\%)^b$	ε (%) c	
Δ	40	4540 (+227)	$4.15 \times 10^{5} (+1.26 \times 10^{5})$			80	99	80	
٥	4	+5+0 (±227)	4.15 × 10 (±1.20 × 10)			20	77	90	
C1	550	$2910 (\pm 146)$	$1.40 \times 10^5 (\pm 4.25 \times 10^4)$			74	79	71	
Ag	0.51	$6.0 \times 10^{-3} \ (\pm 0.0003)$	$0.27 (\pm 0.08)$	$0.65 (\pm 0.05)$	$2.06 (\pm 0.22)$	0.14	0.82	0.54	$0.3 (\pm 0.045)$
Au		$2.0 \times 10^{-3} \ (\pm 2.0 \times 10^{-4})$	$0.68 (\pm 0.22)$	$1.64 (\pm 0.18)$	$0.69 (\pm 0.08)$	1.5	13	1.3	$0.8 (\pm 0.14)$
Sb		$3.3 \times 10^{-2} (\pm 3.3 \times 10^{-3})$	$1.77 (\pm 0.56)$	$4.27 (\pm 0.48)$	$11.3 (\pm 1.2)$	4	10	5.8	$3.8 (\pm 0.65)$
U		$2.4 \times 10^{-2} \ (\pm 1.2 \times 10^{-3})$	$1.85 (\pm 0.56)$	$4.45 (\pm 0.31)$	$8.25 (\pm 0.90)$	0.06	0.72	0.28	$0.2 (\pm 0.03)$
Te		$2.0 \times 10^{-2} \ (\pm 1.0 \times 10^{-3})$	$1.89 (\pm 0.57)$	$4.55 (\pm 0.32)$	$6.87 (\pm 0.76)$	0.52	4.1	1.3	$0.8 (\pm 0.12)$
¥		$1.0 \times 10^{-1} \ (\pm 1.0 \times 10^{-2})$	4.53 (±1.43)	$10.9 (\pm 1.22)$	34.4 (±3.8)	0.52	2.8	1.8	$1.2 (\pm 0.21)$
Cs	3.8	$1.8 \times 10^{-1} \ (\pm 1.8 \times 10^{-2})$	5.66 (±1.79)	$13.6~(\pm 1.52)$	$61.9 (\pm 6.8)$	0.44	2.3	2.2	$1.4 (\pm 0.24)$
Sn		$1.1 \times 10^{-1} \ (\pm 5.5 \times 10^{-3})$	$14.0 (\pm 4.26)$	$33.6 (\pm 2.38)$	$37.8 (\pm 4.2)$	2.0	24	6.8	$4.6 (\pm 0.69)$
Bi		$1.7 \times 10^{-1} \ (\pm 8.5 \times 10^{-3})$	$14.3 (\pm 4.35)$	$34.5 (\pm 2.44)$	$58.4 (\pm 6.4)$	11	62	36	35.2 (±5.3)
Se		$7.4 \times 10^{-1} (\pm 7.4 \times 10^{-2})$	$20.8 (\pm 6.58)$	$50.0 (\pm 5.59)$	$254 (\pm 28)$	41	87	87	429 (±74)
Cd		$4.8 \times 10^{-1} \ (\pm 2.4 \times 10^{-2})$	$21.1 (\pm 6.42)$	$50.9 (\pm 3.60)$	$165 (\pm 18)$	17	83	76	$199 (\pm 30)$
As		$3.0 (\pm 0.3)$	83.0 (±26.3)	$200 (\pm 22.4)$	$1030 (\pm 110)$	5	23	24	$20.2 (\pm 3.5)$
TI									
РЬ	16.2	$1.3~(\pm 0.065)$	$132 (\pm 40.2)$	318 (±22.5)	447 (±49)	1	11	3.6	$2.3 (\pm 0.35)$
Mo									
Cu	87	$3.0 (\pm 0.15)$	283 (±86.1)	682 (±48.2)	$1030 (\pm 110)$	0.39	4.8	1.6	$1.0 (\pm 0.15)$
Zn	112	$0.40~(\pm 0.04)$	491 (±155)	$1180 (\pm 132)$	137 (+15)	0 83	64	0 17	$0.1(\pm 0.017)$

^a Emanation coefficient calculated assuming $\epsilon_{Pb} \sim 1\%^{22,23}$ (see table caption), from ³.

^b Emanation coefficient calculated using plume X/SO_2 and the mass of sulfur degassed (from matrix and melt inclusion glass analysis; see table caption and methods). Melt inclusion and matrix glass data from ^{3,27} (0.34 and 0.004 wt% sulfur).

^c Emanation coefficient calculated using plume X/CI and the mass of chlorine degassed (from matrix and melt inclusion glass analysis; see table caption and methods). Melt inclusion and matrix glass data from ^{3,27} (0.19 and 0.055 wt% chlorine).

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Volcano	Holuhraun						
	Lava	Gas, aerosol ^a	Flux,	X/SO_2	Emanation coefficient	Emanation coefficient	EF_{Cu}
Elements	ppm	µg/m³	kg/day	mg/kg	ε (%) ^b	£ (%)°	
S	400^{28}	185 (±19)				74	500 (±85)
Cl		$3.64 (\pm 0.36)$		$1.97 \times 10^4 (\pm 2.2 \times 10^3)$		32	42 (±7.1)
Ag	0.28	$1.54 \times 10^{-3} \ (\pm 1.5 \times 10^{-4})$	$0.23~(\pm 0.04)$	$2.41 \times 10^{-3} \ (\pm 2.7 \times 10^{-4})$	1.8×10^{-3}	1.9×10^{-3}	$5.9 (\pm 1.0)$
Au							
Sb	0.026	$2.47 \times 10^{-3} \ (\pm 2.5 \times 10^{-4})$	$0.18 (\pm 0.06)$	$1.82 \times 10^{-3} \ (\pm 2.0 \times 10^{-4})$	1.5×10^{-2}	1.5×10^{-2}	$103 (\pm 18)$
U							
Te	0.0048	$1.58 \times 10^{-1} \ (\pm 1.6 \times 10^{-2})$	$22.3 (\pm 7.1)$	$2.28 \times 10^{-1} \ (\pm 2.5 \times 10^{-2})$	9.2	9.5	$35800 (\pm 6200)$
₩	0.093	$7.30 \times 10^{-5} \ (\pm 7.3 \times 10^{-6})$	$1.2 \times 10^{-2} (\pm 3.8 \times 10^{-3})$	$1.26 \times 10^{-4} \ (\pm 1.0 \times 10^{-5})$	2.9×10^{-4}	3.0×10^{-4}	$0.9 (\pm 0.16)$
Cs							
Sn	0.99	$7.28 \times 10^{-2} \ (\pm 7.3 \times 10^{-3})$	$10.2 (\pm 3.2)$	$1.04 \times 10^{-1} \ (\pm 1.1 \times 10^{-2})$	2.2×10^{-2}	2.2×10^{-2}	$80.0 (\pm 14)$
Bi	0.025	$4.80 \times 10^{-2} \ (\pm 4.8 \times 10^{-3})$	$7.28 (\pm 2.3)$	$7.5 \times 10^{-2} (\pm 8.4 \times 10^{-3})$	6.4×10^{-1}	6.6×10^{-1}	$2080 (\pm 360)$
Se	0.014	$1.68 \times 10^{-1} \ (\pm 1.7 \times 10^{-2})$	$19.6~(\pm 6.2)$	$2.01 \times 10^{-1} \ (\pm 2.2 \times 10^{-2})$	2.9	3.1	$12800 (\pm 2200)$
Cd	0.23	$3.30 \times 10^{-1} \ (\pm 3.3 \times 10^{-2})$	$48.7 (\pm 15.4)$	$4.99 \times 10^{-1} \ (\pm 5.6 \times 10^{-2})$	4.5×10^{-1}	4.8×10^{-1}	$1540~(\pm 270)$
As	0.27	$4.09 \times 10^{-2} \ (\pm 4.1 \times 10^{-3})$	$5.38 (\pm 1.7)$	$5.50 \times 10^{-2} \ (\pm 6.2 \times 10^{-3})$	4.4×10^{-2}	4.5×10^{-2}	$167 (\pm 30)$
Π	0.01	$6.07 \times 10^{-3} \ (\pm 6.1 \times 10^{-4})$	$0.73~(\pm 0.23)$	$7.46 \times 10^{-3} \ (\pm 8.3 \times 10^{-4})$	1.6×10^{-1}	1.6×10^{-1}	658 (±114)
Рb	0.51	$8.79 \times 10^{-2} \ (\pm 8.8 \times 10^{-3})$	$12.8 (\pm 4.1)$	$1.31 \times 10^{-1} \ (\pm 1.5 \times 10^{-2})$	5.4×10^{-2}	5.7×10^{-2}	$186 (\pm 32)$
Mo	0.40	$1.26 \times 10^{-3} \ (\pm 1.3 \times 10^{-4})$	$0.2~(\pm 0.06)$	$2.01 \times 10^{-3} \ (\pm 2.2 \times 10^{-4})$	1.1×10^{-3}	1.1×10^{-3}	$3.4 (\pm 0.6)$
Cu	142	$1.30 \times 10^{-1} \ (\pm 1.3 \times 10^{-2})$	$17.7 (\pm 5.6)$	$1.80 \times 10^{-1} \ (\pm 2.0 \times 10^{-2})$	2.7×10^{-4}	2.8×10^{-4}	$1.0 (\pm 0.2)$
Zn	100	$2.37 \times 10^{-1} \ (\pm 2.4 \times 10^{-2})$	$36.9 (\pm 11.7)$	$3.79 \times 10^{-1} \ (\pm 4.2 \times 10^{-2})$	8.0×10^{-4}	8.3×10^{-4}	$2.6 (\pm 0.5)$

a sample name BAR-A from⁶
b Emanation coefficient (expressed in terms of a %) from⁶, calculated using a dilution factor (calculated using at-vent and downwind plume sulfur dioxide concentrations). Error propagated from gas measurements are in the range 10-20%.
c Emanation coefficient calculated assuming the degassing of 1100 ppm sulfur, from 1500 ppm to 400 ppm²⁸ and using the X/SO₂ ratio of the plume to reconstruct

the concentration of element X prior to degassing.

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Volcano	Etna ²							
	Lava	Gas, aerosol ^a	Flux,	X/SO_2^e ,	Emanation Coefficient	Emanation	Emanation	EF_{Cu}
	ppm	$\mu g/m^3$	kg/day	mg/kg	ε (%) ^b	ε (%) ^c	$\varepsilon (\%)^d$	
Elements:								
S	150	$181 (\pm 19)$	$5.0 \times 10^6 (\pm 1.7 \times 10^6)$					71 (±9)
Cl	1700	67 (±7)	$9.2 \times 10^5 (\pm 3.0 \times 10^5)$	$1.85 \times 10^5 \ (\pm 2.0 \times 10^4)$	39.9	56.4	0.80	$2.3 (\pm 0.3)$
Ag	0.9	$2.5 \times 10^{-3} \ (\pm 2.5 \times 10^{-4})$	35 (±12)	$6.91 (\pm 0.77)$	4.47	8.36	0.07	$0.2 (\pm 0.035)$
Au	0.1	$5.1 \times 10^{-4} \ (\pm 5.1 \times 10^{-5})$	$7.0~(\pm 2.3)$	$1.41 (\pm 0.16)$	7.91	14.3	0.10	$0.3~(\pm 0.05)$
Sb	0.12	$3.32 (\pm 0.33)$	$4.5 \times 10^4 (\pm 1.5)$	$9.17 \times 10^3 \ (\pm 1.0 \times 10^3)$	99.8	99.9	85	1630 (=
U	2.5	$3.8 \times 10^{-3} \ (\pm 3.8 \times 10^{-4})$	52.5 (±17.3)	$10.5~(\pm 1.2)$	2.50	4.75	0.035	$0.1 (\pm 0.012)$
Te								
W	1	$1.3 \times 10^{-3} \ (\pm 1.3 \times 10^{-4})$	$18.0~(\pm 5.94)$	$3.59 (\pm 0.4)$	2.14	4.09	0.035	$0.1 (\pm 0.017)$
Cs	0.8	$0.01~(\pm 0.001)$	$138 (\pm 46)$	27.6 (±3.1)	17.4	29.1	0.26	$0.74 (\pm 0.13$
Sn	1.5	$1.4 \times 10^{-3} \ (\pm 1.4 \times 10^{-4})$	$19.3~(\pm 6.8)$	$3.87 (\pm 0.43)$	1.55	2.97	0.035	$0.1 (\pm 0)$
Bi	0.02	$4.0 \times 10^{-2} \ (\pm 4.0 \times 10^{-3})$	552 (±182)	$110 (\pm 12)$	97.1	98.5	29	118 (土
Se	0.05	$3.3 \times 10^{-1} \ (\pm 3.3 \times 10^{-2})$	$4560 (\pm 1500)$	$912 (\pm 101)$	99.1	99.5	58	388 (±
Cd	0.85	$9.0 \times 10^{-2} \ (\pm 9.0 \times 10^{-3})$	$1243\ (\pm 435)$	249 (±27)	64.1	77.7	2.1	6.2 (±
As	_	$3.7 \times 10^{-2} \ (\pm 3.7 \times 10^{-3})$	511 (±179)	$102 (\pm 11)$	38.4	54.9	7.12	22 (±
T1	0.09	$1.8 \times 10^{-1} \ (\pm 1.8 \times 10^{-2})$	$2486 (\pm 870)$	497 (±55)	97.1	98.5	29	118 (土
Pb	~	$3.9 \times 10^{-1} \ (\pm 3.9 \times 10^{-2})$	$5387 (\pm 1890)$	$1077 (\pm 120)$	45.1	61.5	1.0	2.9 (±0
Mo	3.7	$3.0 \times 10^{-2} \ (\pm 3.0 \times 10^{-3})$	$414 (\pm 145)$	82.9 (±9.2)	12.0	21.0	0.17	$0.5 (\pm 0.09)$
Cu	90	$1.53 (\pm 0.15)$	$2.11 \times 10^4 (\pm 7.4 \times 10^3)$	4227 (±470)	22.3	35.8	0.35	$1.0 (\pm 0.17)$
Zn	110	$6.2 \times 10^{-1} (\pm 0.062)$	$8564 (\pm 3000)$	$1710 (\pm 1.90)$	8.67	15.6	0.10	$0.3 (\pm 0.05)$

ET1 of

methods). Melt inclusion and matrix glass data from (0.32 and 0.015 wt% sulfur)²⁶. b Emanation coefficient calculated using plume X/SO2 and the mass of sulfur degassed (from matrix and melt inclusion glass analysis; see table caption and

^e Emanation coefficient calculated using plume X/Cl and the mass of chlorine degassed (from matrix and melt inclusion glass analysis; see table caption and methods). Melt inclusion and matrix glass data from $(0.28 \text{ and } 0.17 \text{ wt}\% \text{ chlorine})^{26}$.

^d Emanation coefficient calculated assuming $\varepsilon_{Pb} \sim 1\%^{22,23}$ (see table caption and methods).

e Note that these X/SO₂ values are high compared with other arc volcanoes in this dataset and with other data from Etna^{1,29}, which causes the emanation coefficients to be anomalously high when using the X/SO₂ ratio, combined with the melt inclusion sulfur contents, to reconstruct the metal contents of the melts (main paper,

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Zn	Cu	Mo	Pb	T1	As	Cd	Se	Bi	Sn	In	Cs	W	Te	U	Sb	Au	Ag	Cl	S	Elements			Volcano
144	251		3.36	0.092		0.113		0.023			0.711		0.033	1.55	0.321				150		ppm	Lava	Masaya ²⁰
$1.69 (\pm 0.17)$	$6.32 (\pm 0.063)$		$0.142 (\pm 0.014)$	$0.21 (\pm 0.021)$		$0.22 (\pm 0.022)$		$0.061 (\pm 0.006)$			$0.029 (\pm 0.0029)$		$0.19 (\pm 0.019)$	$0.01 (\pm 0.001)$	$0.0074 (\pm 0.00074)$				$0.11 (\pm 0.006)$		μg/m ³	Gas, aerosol,	
160 (±18)	600 (±67)		$13.5 (\pm 1.5)$	$20.0 (\pm 2.2)$		$20.9 (\pm 2.3)$		$2.90 (\pm 0.32)$			$1.38 (\pm 0.15)$		$9.0 (\pm 1.0)$	$0.48 (\pm 0.05)$	$0.35 (\pm 0.04)$				$5.0 \times 10^{530} \ (\pm 6.0 \times 10^4)$		mg/kg	X/SO_2 ,	
80 (±26)	$300 (\pm 99)$		6.7 (±2.2)	$10 (\pm 3.4)$		$11 (\pm 3.6)$		$1.5~(\pm 0.5)$			$0.69 (\pm 0.23)$		$4.5 (\pm 1.5)$	$0.24 (\pm 1.2)$	$0.18 (\pm 0.06)$				$5.0 \times 10^5 (\pm 1.65 \times 10^4)$		kg/day	Flux,	
$0.47 (\pm 0.06)$	$1.0~(\pm 0.33)$		1.7 (±0.22)	91 (±12)		77 (±10)		$110 (\pm 14)$			$1.6 (\pm 0.21)$		230 (±29)	$0.26 (\pm 0.034)$	$0.92 (\pm 0.12)$				$2.8 \times 10^3 (\pm 0.4 \times 10^3)$			EF_{Cu}	

Ag 3.32 × 10 ⁻¹ Au 1.5 × 10 ⁻³ Au 1.5 × 10 ⁻³ Sb 1.7 × 10 ⁻¹ 2.5 × 10 ⁻¹ U 4.2 × 10 ⁻¹ 3.6 × 10 ⁻¹ Te 6.3 × 10 ⁻³ 1.34 × 10 W 2.80 × 10 ⁻¹ 1.46 × 10 Cs 8.6 × 10 ⁻² 1.38 × 10 In 4.6 1.00 × 10 Sn 2.15 5.1 × 10 ⁻¹ Bi 1.5 × 10 ⁻¹ 4.3 × 10 Se 1.4 × 10 ⁻¹ 1.03 × 10 As 1.5 2.73 × 10 As 1.5 2.73 × 10 Pb 4.0 4.0 × 10 ⁻² 2.8 × 10 ⁻¹ Mo 9.50 × 10 ⁻¹ 4.7 × 10 Ch 137 1.5 × 10 ⁻¹	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.2 \times 10^{-1} \\ 1.5 \\ 4.9 \times 10^{-2} \\ 4.0 \\ 9.50 \times 10^{-1} \end{array}$	3.32×10^{-1} 1.5×10^{-3} 1.7×10^{-1} 4.2×10^{-1} 6.3×10^{-3} 2.80×10^{-1} 8.6×10^{-2} 4.6 2.15 1.5×10^{-1} 1.4×10^{-1} 1.2×10^{-1} 1.5×10^{-1} 1.5×10^{-1} 1.5×10^{-1} 1.4×10^{-1} 1.5×10^{-1} 1.5×10^{-1} 1.6×10^{-2} 1.7×10^{-1} 1.9×10^{-2} 1.9×10^{-2}	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.2 \times 10^{-1} \\ 1.5 \\ 4.9 \times 10^{-2} \\ 4.0 \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.5 \\ 4.9 \times 10^{-2} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.2 \times 10^{-1} \\ 1.5 \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.2 \times 10^{-1} \\ 1.5 \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \\ 1.2 \times 10^{-1} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \\ 1.4 \times 10^{-1} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 4.2 \times 10^{-3} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \\ 1.5 \times 10^{-1} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \\ 2.15 \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \\ 4.6 \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \\ 8.6 \times 10^{-2} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \\ 2.80 \times 10^{-1} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \\ 6.3 \times 10^{-3} \end{array}$	$\begin{array}{c} 402 \\ 3.32 \times 10^{-1} \\ 1.5 \times 10^{-3} \\ 1.7 \times 10^{-1} \\ 4.2 \times 10^{-1} \end{array}$	3.32×10^{-1} 1.5×10^{-3} 1.7×10^{-1}	3.32×10^{-1} 1.5×10^{-3}	3.32×10^{-1}		S 12	Elements:	h wdd	Lava ¹⁹ Gas,	Volcano Kīlauca
$2.5 \times 10^{-3} (\pm 0.4 \times 10^{-4})$ $3.6 \times 10^{-5} (\pm 5.0 \times 10^{-6})$ $1.34 \times 10^{-1} (\pm 1.5 \times 10^{-2})$ $1.46 \times 10^{-4} (\pm 2.6 \times 10^{-5})$ $1.48 \times 10^{-4} (\pm 3.0 \times 10^{-5})$ $1.00 \times 10^{-3} (\pm 1.0 \times 10^{-5})$ $5.1 \times 10^{-2} (\pm 9.0 \times 10^{-3})$ $4.3 \times 10^{-3} (\pm 1.9 \times 10^{-4})$ $1.49 (\pm 0.15)$ $1.03 \times 10^{-1} (\pm 1.2 \times 10^{-2})$ $2.73 \times 10^{-2} (\pm 2.0 \times 10^{-3})$ $2.8 \times 10^{-3} (\pm 0.4 \times 10^{-3})$ $4.0 \times 10^{-2} (\pm 0.3 \times 10^{-2})$ $4.7 \times 10^{-3} (\pm 0.5 \times 10^{-3})$ $4.7 \times 10^{-3} (\pm 0.5 \times 10^{-3})$															<u></u>	$)^{-3} (\pm 0.4 \times 10^{-4})$ $)^{-5} (\pm 5.0 \times 10^{-6})$	$)^{-3} (\pm 0.4 \times 10^{-4})$			$16 (\pm 1.5)$			$\mu g/m^3$	Gas, aerosol ¹⁴	
$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 45)$ $3.8 (\pm 2.3)$ $1.1 (\pm 0.5)$ $0.13 (\pm 0.07)$ $3.1 (\pm 2.3)$ $1.7 \times 10^{-1} (\pm 0.1)$ $4.5 (\pm 4.0)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 4.5)$ $3.8 (\pm 2.3)$ $1.1 (\pm 0.5)$ $0.13 (\pm 0.07)$ $3.1 (\pm 2.3)$ $1.7 \times 10^{-1} (\pm 0.1)$	4.8 (±3.0) 1.4 × 10 ⁻² (±8.0 × 10 ⁻³) 6.0 × 10 ⁻³ (±2.0 × 10 ⁻³) 8.0 × 10 ⁻² (±7.0 × 10 ⁻²) 4.5 (± 2.3) 0.21 (±0.07) 62.5 (±45) 3.8 (±2.3) 1.1 (±0.5) 0.13 (±0.07) 3.1 (±2.3)	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 4.5)$ $3.8 (\pm 2.3)$ $1.1 (\pm 0.5)$ $0.13 (\pm 0.07)$ $3.1 (\pm 2.3)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 4.5)$ $3.8 (\pm 2.3)$ $1.1 (\pm 0.5)$ $0.13 (\pm 0.07)$	4.8 (±3.0) 1.4 × 10 ⁻² (±8.0 × 10 ⁻³) 6.0 × 10 ⁻³ (±2.0 × 10 ⁻³) 8.0 × 10 ⁻² (±7.0 × 10 ⁻²) 4.5 (± 2.3) 0.21 (±0.07) 62.5 (±45) 3.8 (±2.3) 1.1 (±0.5)	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 4.5)$ $3.8 (\pm 2.3)$ $1.1 (\pm 0.5)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 45)$ $3.8 (\pm 2.3)$	$\begin{array}{c} 4.8 \ (\pm 3.0) \\ 1.4 \times 10^{-2} \ (\pm 8.0 \times 10^{-3}) \\ 6.0 \times 10^{-3} \ (\pm 2.0 \times 10^{-3}) \\ 8.0 \times 10^{-2} \ (\pm 7.0 \times 10^{-2}) \\ 4.5 \ (\pm 2.3) \\ 0.21 \ (\pm 0.07) \\ 62.5 \ (\pm 4.5) \end{array}$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$ $62.5 (\pm 4.5)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$ $0.21 (\pm 0.07)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$ $4.5 (\pm 2.3)$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$ $8.0 \times 10^{-2} (\pm 7.0 \times 10^{-2})$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$ $6.0 \times 10^{-3} (\pm 2.0 \times 10^{-3})$	$4.8 (\pm 3.0)$ $1.4 \times 10^{-2} (\pm 8.0 \times 10^{-3})$	4.8 (±3.0)			$0.07 (\pm 0.06)$		$448 (\pm 45)$	$7 \times 10^{814} (\pm 65 \times 10^6)$		kg/day	Flux,	
0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-2}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-2}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1) 0.19 (± 0.02) 3.6 (± 0.4) 0.23 (± 0.03) 5.1 (± 0.06)	0.3 (±0.7) 0.017 (±2 × 10 ⁻³) 9.5 × 10 ⁻³ (±1 × 10 ⁻³) 9.5 × 10 ⁻³ (±1 × 10 ⁻²) 0.1 (±1 × 10 ⁻²) 5.3 (±6 × 10 ⁻¹) 0.3 (±3 × 10 ⁻²) 91 (±10) 5.3 (±0.6) 1.2 (±0.1) 0.19 (±0.02) 3.6 (±0.4) 0.23 (±0.03)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10 ⁻³ ($\pm 1 \times 10^{-2}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1) 0.19 (± 0.02) 3.6 (± 0.4)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-2}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1) 0.19 (± 0.02) 3.6 (± 0.4)	0.3 (± 0.7) 0.017 $(\pm 2 \times 10^{-3})$ 9.5 $\times 10^{-3} (\pm 1 \times 10^{-3})$ 9.5 $\times 10^{-3} (\pm 1 \times 10^{-2})$ 0.1 $(\pm 1 \times 10^{-2})$ 5.3 $(\pm 6 \times 10^{-1})$ 0.3 $(\pm 3 \times 10^{-2})$ 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1) 0.19 (± 0.02)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10 ⁻³ ($\pm 1 \times 10^{-2}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-3}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10) 5.3 (± 0.6) 1.2 (± 0.1)	0.3 (± 0.7) 0.017 $(\pm 2 \times 10^{-3})$ 9.5 $\times 10^{-3} (\pm 1 \times 10^{-3})$ 9.1 $(\pm 1 \times 10^{-2})$ 5.3 $(\pm 6 \times 10^{-1})$ 0.3 $(\pm 3 \times 10^{-2})$ 91 (± 10) 5.3 (± 0.6)	0.3 (±0.7) 0.017 (±2 × 10 ⁻³) 9.5 × 10 ⁻³ (±1 × 10 ⁻³) 0.1 (±1 × 10 ⁻²) 5.3 (±6 × 10 ⁻¹) 0.3 (±3 × 10 ⁻²) 91 (±10)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-2}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$) 91 (± 10)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-3}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$) 0.3 ($\pm 3 \times 10^{-2}$)	0.3 (± 0.7) 0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10^{-3} ($\pm 1 \times 10^{-3}$) 0.1 ($\pm 1 \times 10^{-2}$) 5.3 ($\pm 6 \times 10^{-1}$)	$0.017 (\pm 2 \times 10^{-3})$ $0.017 (\pm 2 \times 10^{-3})$ $9.5 \times 10^{-3} (\pm 1 \times 10^{-3})$ $0.1 (\pm 1 \times 10^{-2})$	0.017 ($\pm 2 \times 10^{-3}$) 9.5 × 10 ⁻³ ($\pm 1 \times 10^{-3}$)	0.017 $(\pm 2 \times 10^{-3})$	0.5 (±0.7)	くっ (±0 7)	$0.0029~(\pm 3\times 10^{-4})$	$0.12 (\pm 0.01)$		$640 (\pm 45)$			mg/kg	X/SO_2	
9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴) 6.4 × 10 ³ (±1 × 10 ³) 1.7 × 10 ³ (±3 × 10 ³) 1.8 × 10 ² (±30) 2.5 × 10 ³ (±400) 2.9 × 10 ² (±50) 9.4 × 10 ³ (±1 5 × 10 ³)	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴) 6.4 × 10 ³ (±1 × 10 ³) 1.7 × 10 ³ (±3 × 10 ³) 1.8 × 10 ² (±30) 2.5 × 10 ³ (±400) 2.9 × 10 ² (±50)	9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 6.3 (±0) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴) 6.4 × 10 ³ (±1 × 10 ³) 1.7 × 10 ³ (±3 × 10 ³) 1.8 × 10 ² (±30) 2.5 × 10 ³ (±400)	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴) 6.4 × 10 ³ (±1 × 10 ³) 1.7 × 10 ³ (±3 × 10 ³) 1.8 × 10 ² (±30) 2.5 × 10 ³ (±400)	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴) 6.4 × 10 ³ (±1 × 10 ³) 1.7 × 10 ³ (±3 × 10 ³) 1.8 × 10 ² (±30)	$\begin{array}{c} 9.1 & (\pm 1.3) \\ 9.1 & (\pm 1.3) \\ 8.6 & (\pm 1.3) \\ 8.6 & (\pm 1.3) \\ 63 & (\pm 10) \\ 3.2 \times 10^3 & (\pm 5 \times 10^2) \\ 2.7 \times 10^2 & (\pm 40) \\ 9.3 \times 10^4 & (\pm 1 \times 10^4) \\ 6.4 \times 10^3 & (\pm 1 \times 10^3) \\ 1.7 \times 10^3 & (\pm 3 \times 10^3) \\ 1.7 \times 10^3 & (\pm 3 \times 10^3) \end{array}$	9.1 (± 1.3) 9.1 (± 1.3) 8.6 (± 1.3) 8.6 (± 1.3) 63 (± 10) 3.2 \times 10 ³ $(\pm 5 \times 10^2)$ 2.7 \times 10 ² (± 40) 9.3 \times 10 ⁴ $(\pm 1 \times 10^4)$ 6.4 \times 10 ³ $(\pm 1 \times 10^3)$ 1.7 \times 10 ³ $(\pm 3 \times 10^3)$	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10^3 (±5 × 10^2) 2.7 × 10^2 (±40) 9.3 × 10^4 (±1 × 10^4) 6.4 × 10^3 (±1 × 10^3)	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²) 2.7 × 10 ² (±40) 9.3 × 10 ⁴ (±1 × 10 ⁴)	9.1 (±1.3) 9.1 (±1.3) 8.6 (±1.3) 8.3 (±10) 3.2 × 10^3 (±5 × 10^2) 2.7 × 10^2 (±40) 9.3 × 10^4 (±1 × 10^4)	9.1 (± 1.3) 9.6 (± 1.3) 8.6 (± 1.3) 63 (± 10) 3.2 \times 10 ³ $(\pm 5 \times 10^2)$ 2.7 \times 10 ² (± 40)	9.1 (±1.3) 9.6 (±1.3) 8.6 (±1.3) 63 (±10) 3.2 × 10 ³ (±5 × 10 ²)	9.1 (±1.3) 9.6 (±1.3) 63 (±10)	$9.1 (\pm 1.3)$ $9.1 (\pm 1.3)$ $8.6 (\pm 1.3)$	$9.1 (\pm 1.3)$	0.4 ~ 10 (±1 ~ 10)	$8.4 \times 10^{3} (+1 \times 10^{3})$	$2.3 (\pm 0.3)$	$1.6 \times 10^2 \ (\pm 20)$					mg/kg	X/Cl	
70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18 0.88 0.21 0.056	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18 0.88 0.21	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18 0.88 0.21	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18 0.88	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18 0.88	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18	70 0.014 0.025 0.0050 0.56 0.46 60 9.2 0.18	70 0.014 0.025 0.0050 0.56 0.46 60 9.2	70 0.014 0.025 0.0050 0.56 0.46 60	70 0.014 0.025 0.0050 0.56 0.46	70 0.014 0.025 0.0050 0.56 0.46	70 0.014 0.025 0.0050 0.56	70 0.014 0.025 0.0050	70 0.014 0.025	70 0.014	70		0.0016	0.16		0.42	89		ε (%) a	Emanation coefficient	1
99 0.97 2.92 0.41 31 35 35 99 94 25.4 52 16	99 0.97 2.92 0.41 31 35 39 94 25.4 52 16	99 0.97 2.92 0.41 31 35 39 94 25.4	99 0.97 2.92 0.41 31 35 35 99 94 25.4	99 0.97 2.92 0.41 31 35 39 99 94 25.4	99 0.97 2.92 0.41 31 35 39 99	99 0.97 2.92 0.41 31 35 39 99	99 0.97 2.92 0.41 31 35 39 99	99 0.97 2.92 0.41 31 35 99	99 0.97 2.92 0.41 31 35	99 0.97 2.92 0.41 31 35	99 0.97 2.92 0.41 31	99 0.97 2.92 0.41	99 0.97 2.92	99 0.97	99	0.10	n 16	0.22		30	89		ε (%) b	Emanation coefficient	1
0.0087 96 0.053 0.16 18 2.3 22 99 46 1.8 5.5 1	0.0087 96 0.053 0.16 18 2.3 22 29 99 46 1.8 5.5	0.0087 96 0.053 0.16 18 2.3 22 99 46 1.8 5.5	0.0087 96 0.053 0.16 18 2.3 22 99 46 1.8 5.5	0.0087 96 0.053 0.16 18 2.3 22 99 46 1.8 5.5	0.0087 96 0.053 0.16 18 2.3 22 29 99	0.0087 96 0.053 0.16 18 2.3 22 99 46	0.0087 96 0.053 0.16 18 2.3 22 99	0.0087 96 0.053 0.16 18 2.3 22 99	0.0087 96 0.053 0.16 18 2.3 22	0.0087 96 0.053 0.16 18 2.3 2.2	0.0087 96 0.053 0.16 18 2.3	0.0087 96 0.053 0.16 18	0.0087 96 0.053 0.16	0.0087 96 0.053	0.0087 96	0.0087		1.46		3.3	44		ε (%) c	Emanation coefficient	1
$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$ $53.3 (\pm 11)$ $9.32 (\pm 1.6)$ $4.61 (\pm 0.86)$ $1.00 (\pm 0.16)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 4.3)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$ $53.3 (\pm 11)$ $9.32 (\pm 1.6)$ $4.61 (\pm 0.86)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$ $53.3 (\pm 11)$ $9.32 (\pm 1.6)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$ $53.3 (\pm 1.1)$ $9.32 (\pm 1.6)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{4} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$ $53.3 (\pm 11)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{4} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$ $17.0 (\pm 2.9)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$ $800 (\pm 150)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{4} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$ $1.39 \times 10^{5} (\pm 2.5 \times 10^{4})$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$ $267 (\pm 43)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$ $22.1 (\pm 5.2)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$ $202 (\pm 37)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$ $1.50 (\pm 0.4)$	$1.98 \times 10^{4} (\pm 3.8 \times 10^{3})$ $4.86 \times 10^{-1} (\pm 0.11)$	$1.98 \times 10^4 \ (\pm 3.8 \times 10^3)$	0.000 (-0.01)	$0.080 (\pm 0.02)$	$13.7 (\pm 2.1)$						EF_{Cu}	

methods). Melt inclusion and matrix glass data: 0.13 and 0.015 wt% sulfur³¹. ^a Emanation coefficient calculated using plume X/SO₂ and the mass of sulfur degassed (from matrix and melt inclusion glass analysis; see table caption and

b Emanation coefficient calculated using plume X/Cl and the mass of chlorine degassed (from matrix and melt inclusion glass analysis; see table caption and methods). Melt inclusion and matrix glass data: 0.05 and 0.035 wt% chlorine²⁵. Note that the uncertainty on this emanation coefficient is high owing to the low concentration of Cl in the plume and the uncertainty associated with estimating the chlorine budget from melt inclusions and matrix glasses^{21,32} ^e Emanation coefficient calculated assuming $\varepsilon_{Pb} \sim 1\%^{22.23}$ (see table caption and methods).

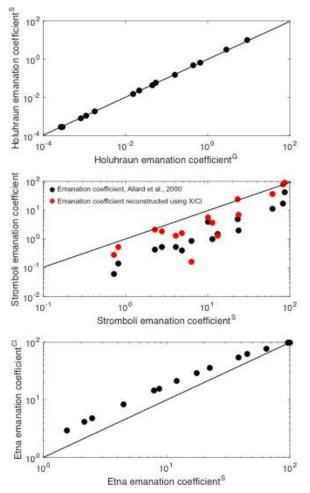
Zn	Cu	Mo	Pb	Tl	As	Cd	Se	Bi	Sn	Cs	₩	Te	U	Sb	Au	Ag	CI	S	Elements			Volcano
89.5	129.00		5.9	0.08	1.7	0.15			_	1.1			0.5				0.07	0.015		ppm	$Lava^5$,	Ambrym ⁵
$0.58 (\pm 0.035)$	$1.60~(\pm 0.10)$		$0.09 (\pm 0.004)$	$0.195 (\pm 0.012)$	$0.25 (\pm 0.025)$	$3.2 \times 10^{-2} \ (\pm 3.2 \times 10^{-3})$	$3.1 \times 10^{-1} (\pm 0.03)$	$4.1 \times 10^{-2} \ (\pm 1.6 \times 10^{-3})$	$0.8 (\pm 0.08)$	$1.8 \times 10^{-2} \ (\pm 1.1 \times 10^{-3})$			$1.3 \times 10^{-2} \ (\pm 5.2 \times 10^{-4})$		$3.1 \times 10^{-3} \ (\pm 1.9 \times 10^{-4})$	$3.20 (\pm 0.19)$				$\mu g/m^3$	Gas, aerosol ^s ,	
$350 (\pm 110)$	$1300 (\pm 400)$		90 (±27)	$160~(\pm 49.0)$	$300 (\pm 95)$	$46 (\pm 15)$	$360 (\pm 114)$	32 (±9.7)	$1500 (\pm 470)$	12 (±3.7)					$1.6 (\pm 0.5)$	$23 (\pm 4.0)$				kg/day	Flux,	
0.11	0.18		0.3	26	4.1	7.4	5.9	6.2	18	0.3			0.3				30	95		E (%)a	Emanation coefficient	
$0.52 (\pm 0.086)$	$1.00 (\pm 0.16)$		$1.2 (\pm 0.19)$	$197 (\pm 32.4)$	$11.9 (\pm 2.2)$	$17.2 (\pm 3.2)$			$64.5 (\pm 11.8)$	$1.3 (\pm 0.21)$			$2.1 (\pm 0.33)$								EF_{Cu}	

^a Emanation coefficient calculated using measurements of ε_{Pb} from⁵ (see table caption and methods).

Comparison of different ways to calculate emanation coefficients

The tables above show various different ways to estimate the emanation coefficient, $\varepsilon_x = \frac{\left(c_i - c_j\right)}{c_i}$ where c_i is the initial concentration of element x in the magma and

concentration of element x in the post-eruptive lava (as originally defined by 22). The **supplementary figure 4** show how they compare to one another. The emanation co calculated using the various methods are broadly consistent with one another in tern volatility, with the methods using X/SO_2 and X/Cl to "add back" metals in the propo these ratios to calculate c_i and thus emanation coefficient yielding higher emanation than the methods based on measurement of the emanation coefficient of $^{210}\text{Pb}^{23}$, by



Supplementary Figure 4: top: e1 coefficient for metals and semi-m Holuhraun calculated by by usin factor calculated from their measi sulfur in vent and plume gases (x emanation coefficient calculated using the mass of sulfur outgasse (from melt inclusion and matrix g microanalysis²⁸) and X/SO₂ ratios to calculate c_i (y axis). Middle: e coefficient for Stromboli metals a calculated in this study using X/Sreconstruct c_i (x axis) plotted ver coefficient from³ using measuren emanation coefficient of ²¹⁰Pb²³ (using X/Cl (this study) to calculat table caption and methods for det comparison between emanation c calculated using X/SO_2 (x axis) w calculated using X/Cl (y axis) to α Mt Etna (Italy). Black solid line i

Metal partition coefficients

Table 2: Vapor-melt ($D_{vapor-melt}$) and sulfide-silicate melt ($D_{sulfide-silicate\ melt}$) partition coefficients used in this paper. Sources are shown either in column header or in individual cells.

	$D_{vapor-melt}^{33}$	$D_{vapor-melt}$	$D_{sulfide ext{-}silicate\ melt}$	$D_{\it sulfide-silicate melt}^{\it 34}$	$D_{sulfide\text{-}silicate\ melt}^{35}$
U		0.002 to 0.45 ³⁶	0.001^{37}		
Cs	2.3 to 5.9	$0.5 \text{ to } 10^{38}$			
W	1.8 to 4.5	3 to 16 ³⁹		0.003 to 0.1	
Cu	33 to 70	1 to 80 ⁴⁰	260 to 447 ⁴¹	540 to 1040	1042 to 1624
Zn	2.3 to 5	9 to 70	1.9 to 3.1 ⁴¹	0.28 to 0.94	2.4 to 4.7
Mo	5.5 to 9.8	9.3 to 523 ³⁹		0.1 to 2.58	
Ag	1.9 to 7	7 to 10^{38}	360 to 617 ⁴¹	300 to 700	853 to 1528
Sn		$0.002 \text{ to } 0.078^{40}$		2.4 to 6.0	9 to 13
Sb	0.14 to 2.6	1 to 10 ³⁸	15 to 34 ⁴¹	1.4 to 11.2	
As	3 to 6.2	1 to 2.5 ⁴²		0.3 to 19.7	
Pb	8.4 to 15	2 to 14 ³⁹	26 to 54 ⁴¹	13 to 48	45 to 71
Tl	3.1 to 6	3.5 to 12.4 ³⁹	10 to 17 ⁴¹		
Cd	6.5 to 30		45 to 82 ⁴¹		61 to 193
Bi	1 to 5.3	7 to 30^{38}		110 to 1130	
Se	0.59 to 12		790 to 1500 ⁴¹		
Au		12 to 15 ⁴²		930 to 5500	
Te	0.22 to 6.7		6000 to 16000 ⁴¹		

Notes on experimental conditions.

³³ Partitioning of metals between vapor and basalt silicate melt at 850 °C, 2 kbar, Cl- and H₂O-bearing fluids.

³⁴ Partitioning between basanite melt, sulfide liquid and monosulfide solid solution, at 1175 to 1300 °C, 1.5-3.0 GPa, 3 units below to 1 unit above FMQ buffer, relevant to mantle conditions.

³⁵ Mid-ocean ridge basalts at FMQ to FMQ-1.

³⁶ Haplogranite melt composition, aqueous fluid with HCl and HF, 2 kbar, 750 C.

 $^{^{37}}$ 1-10 GPa, 1750–2100 °C, 0–28 wt% S, and fO_2 2 log units below IW (core conditions).

³⁸ Granitic and peralkaline melts, melts with high chlorinities (1-14 mole/kg), $\log fO_2 = NNO-1.7$ to NNO+4.5.

³⁹ Haplogranite melt composition, with H_2O -HCl and H_2O -HF vapor phase. For Cu, lowest $D_{vapor-melt}$ for H_2O -only and highest for Cl-rich case; for Sn highest $D_{vapor-melt}$ for Cl-rich case, but poorly constrained.

⁴⁰ Haplogranite melt composition, aqueous fluid with HCl and HF, 2 kbar, 750 C.

⁴¹ The silicate constituent was either a MORB or a composition close to the 1.5 GPa eutectic composition in the system anorthite–diopside–forsterite (An₅₀Di₂₈Fo₂₂). FeO was added to vary FeO activity.

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