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### ► To cite this version:

Lisard Torr , Diego Benites, Jean Vallance, Oscar Laurent, Brayam Ortiz-Benavente, et al.. Trace element geochemistry of sphalerite and chalcopyrite in arc-hosted VMS deposits. *Journal of Geochemical Exploration*, 2022, 232, pp.106882. 10.1016/j.gexplo.2021.106882 . hal-03436650

**HAL Id: hal-03436650**

**<https://hal.science/hal-03436650>**

Submitted on 23 Nov 2021

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1 **Trace element geochemistry of sphalerite and chalcopyrite in arc-hosted VMS**  
2 **deposits**

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13 Submitted to *Journal of Geochemical Exploration*

14

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18

19 **Abstract**

20 Trace element compositions of sphalerite and chalcopyrite have been investigated for four  
21 arc-hosted Cretaceous VMS deposits (Palma, Perubar, and María Teresa in Peru, and Cerro  
22 de Maimón in the Dominican Republic) using laser ablation inductively-coupled plasma  
23 mass spectrometry. In sphalerite, Fe, Mn, Cd, Hg, Ag, Sb, Se, In, Ge, and Ga are lattice-  
24 bound, whereas Sn, Tl, Bi, and Pb occur at least partly as mineral microinclusions.

25 Significant variations in the contents of minor and trace elements are observed in sphalerite  
26 grains from individual deposits. A strong negative correlation between Fe (Inter-Quartile  
27 Range [IQR] 44009-18168 ppm) and Zn indicates a dominant  $Fe^{2+} \leftrightarrow Zn^{2+}$  simple  
28 isovalent substitution. Regarding potential *high-tech* by-products, the contents of In (IQR  
29 42-1.6 ppm, up to 415 ppm) and Ga (IQR 31-5.9 ppm, up to 96 ppm) in the studied  
30 sphalerite are normally much higher than those of Ge (IQR 0.85-0.16 ppm, up to 9.3 ppm).  
31 Correlation trends suggest  $Cu^+ + In^{3+} \leftrightarrow 2Zn^{2+}$ ,  $Cu^+ + Ga^{3+} \leftrightarrow 2Zn^{2+}$ , and additional, more  
32 complex substitution mechanisms of Zn involving combinations of monovalent ( $Cu^+$ ,  $Ag^+$ ),  
33 divalent ( $Ge^{2+?}$ ), trivalent ( $In^{3+}$ ,  $Ga^{3+}$ , [ $Sn^{3+?}$ ]), and quadrivalent ( $Ge^{4+}$ ,  $Sn^{4+}$ ) cations.

34 As for chalcopyrite, Zn, Ag, Sn, Cd, Se, In, Ga, and Ge are interpreted to be lattice-bound,  
35 whereas Mo, Au, Tl, Sb, Pb, and Bi probably occur as microinclusions. Relative to  
36 sphalerite, chalcopyrite is depleted in In (IQR 28-8.4 ppm, up to 49 ppm) and Ga (IQR 8.3-  
37 2.9 ppm, up to 24 ppm) and enriched in Ge (IQR 5.9-0.70 ppm, up to 80 ppm). Of the  
38 other trace elements, Zn (IQR 426-190 ppm) is the most highly concentrated in the studied  
39 chalcopyrite, followed by Ag (IQR 136-23 ppm), Se (IQR 64-22 ppm), Sn (IQR 53-1.3  
40 ppm), and Cd (IQR 12-6.0 ppm). General positive correlation trends between Zn, Cd, In,  
41 Ge, and Ga in chalcopyrite suggest varied coupled substitution mechanisms of Fe and Cu  
42 with fluctuating valences due to covalent bonding.

43 Trace-element distribution patterns in sphalerite and chalcopyrite were studied for the  
44 zone-refined Sofia-D massive sulfide body in the María Teresa deposit, which comprises a  
45 core of dominant pyrite sheathed outward with zones of chalcopyrite (Cu zone), sphalerite  
46 (Zn zone) and galena + sphalerite  $\pm$  fahlore  $\pm$  barite (Pb – Zn – Ag zone). Bottom to top of  
47 the sulfide body, sphalerite records progressive depletion in In, Cu, Mn, and Se, and

48 enrichment in Ge. This distribution pattern agrees with increasing crystallization  
49 temperatures and/or volatile magmatic influx towards the core of the massive  
50 mineralization. Distribution of trace elements in chalcopyrite is rather uneven except for a  
51 sustained enrichment in Se towards the basal portion of the sulfide body. The fact that such  
52 trends are preserved in spite of extensive recrystallization during thermal metamorphism in  
53 parts of the Sofia-D massive sulfide mineralization suggests i) a closed metamorphic  
54 system and ii) that element interdiffusion was prominently local. Accordingly, we propose  
55 that sphalerite lattice-bound trace elements distribution patterns described in this article  
56 can help determine the polarity of massive sulfide bodies in VMS districts in  
57 metamorphosed and tectonized terranes.

58 **Keywords:** Sulfide; LA-ICP-MS; Coupled Substitution; Compositional Profile; Volcanic  
59 Arc; Latin America

## 60 **1. Introduction**

61 Volcanogenic massive sulfide (VMS) deposits are exhalative and/or replacive  
62 accumulations of sulfide minerals formed at or near the seafloor where uprising  
63 hydrothermal fluids driven by synchronous magmatism are entrained with cold seawater or  
64 porewaters (Doyle and Allen, 2003; Shanks and Thurston, 2012; Tornos et al., 2015). VMS  
65 mineralization occurs as sulfide mounds to tabular stratiform bodies, which are often  
66 underlain by a stockwork or feeder zone (Galley et al., 2007; Tornos et al., 2015). Their  
67 formation responds to the convergence in the time and space of anomalously high thermal  
68 gradients and extensional tectonic regimes that promote the circulation of evolved  
69 hydrothermal fluids through, and leaching of metals from, the oceanic strata (Ohmoto,  
70 1996; Tivey, 2007; Piercey, 2011). Such geological features are primarily met in oceanic  
71 spreading ridges, arc environments, and rifted continental margins.

72 Most VMS deposits are preserved in suprasubduction zones either in back-arc, intra-arc, or  
73 forearc basins on both oceanic and continental crust (Franklin et al., 1998; Nimis et al.,  
74 2003; Herrington et al., 2005; Hollis et al., 2014; Peter et al., 2014; Torr o et al., 2016a). In  
75 arc-related systems, contrasting water depths in massive sulfide formation exert a major  
76 control on the ore and hydrothermal alteration mineral assemblages and the metalliferous

77 contents (Sillitoe et al., 1996; Dubé et al., 2007; Monecke et al., 2014). VMS deposits in  
78 back-arc and forearc basins form at water depths typically in the range between 1500 to  
79 3700 m, whereas those in arc volcanoes and arc-related rifts are normally restricted to  
80 water depths in the range between 700 to 2000 m. The latter are often associated with  
81 elevated magmatic gas contents and boiling of the hydrothermal fluids may occur in  
82 shallower arc volcanoes and rifted intra-arc submarine hydrothermal systems (Gemmell et  
83 al., 2004; Dubé et al., 2007; Monecke et al., 2014).

84 VMS deposits are important sources of Cu, Zn, Pb, Au, and Ag (Galley et al., 2007). In  
85 addition, a wealth of other metals and metalloids that are potentially economically  
86 recoverable as byproducts (e.g., Sn, Co, Ni, In, Ga, Ge, Mo, Re, Se, Te) or deleterious (i.e.,  
87 harmful or unwanted elements; e.g., Mn, As, Sb, Hg, Tl) may occur in remarkable  
88 concentrations in VMS deposits and even determine their economic viability (Benzaazoua  
89 et al., 2003; Zaccarini and Garuti, 2008; Koski and Mosier, 2012; Ridley, 2012; Carvalho  
90 et al., 2018; Frenzel et al., 2019). Minor and trace elements can either precipitate as native  
91 elements or substitute into the crystal lattice of major sulfide minerals (Cook et al., 2009,  
92 2012, 2015; Ye et al., 2011; Ridley, 2012; Murakami and Ishihara, 2014; George et al.,  
93 2015, 2018; Benites et al., 2021a,b). The understanding of the mineralogical expression of  
94 either potential by-product or deleterious minor and trace elements in a given deposit or  
95 mineralization style is essential to predict their metallurgical behavior (e.g., Cook et al.,  
96 2011).

97 This study aims at understanding the mineralogical expression of minor and trace elements  
98 in sphalerite and chalcopyrite from arc-related VMS deposits and their distribution across  
99 massive sulfide bodies. To do so, we have systematically determined the trace element  
100 budget on these ore minerals from four arc-related VMS deposits in the Andes (Palma,  
101 Perubar, and María Teresa in Peru) and the Caribbean (Cerro de Maimón in the Dominican  
102 Republic). The obtained contents are compared with published data on other VMS deposits  
103 worldwide. In addition, a comprehensive set of samples has allowed the study of ore body-  
104 scale zoning of certain trace elements in a zone-refined massive sulfide body in the María  
105 Teresa deposit. The obtained results are used to assess the incorporation mechanisms of  
106 some *high-tech* elements to the crystal lattice of major sulfides (sphalerite and

107 chalcopyrite) and contribute to revealing element distribution patterns across massive  
108 sulfide bodies in VMS districts; such patterns can be used to target the most prospective  
109 zones of zoned massive sulfide bodies for potential by-products and complementarily also  
110 to constrain sulfide body polarity during exploration in tectonized terranes.

## 111 **2. Geological Setting**

### 112 2.1. Volcanogenic massive sulfide deposits in Peru

#### 113 2.1.1. General geologic framework

114 In Peru, several VMS deposits occur along the coastal region and are hosted by rocks of  
115 Cretaceous age (Steinmüller et al., 2000; Fontboté, 2019). In northern Peru, the giant  
116 Tambogrande deposits represent the largest VMS endowment in this country and probably  
117 in South America, and include the TG1 deposit with resources of 109 million metric tons  
118 (Mt) grading 1.6 % Cu, 1.0 % Zn, 0.5 g/t Au, and 22 g/t Ag, plus 16.7 Mt grading 3.5 g/t  
119 Au and 64 g/t Ag in oxide ore, and the TG3 deposit with resources of 82 Mt grading 1.0 %  
120 Cu, 1.4 % Zn, 0.8 g/t Au, and 25 g/t Ag (Manhattan Minerals, 2002 in Winter et al., 2010).  
121 There, VMS mineralization is hosted by subalkaline (calc-alkaline and tholeiitic) basaltic  
122 pillowed and brecciated lavas and lesser amounts of felsic volcanic rocks that filled N and  
123 NW-oriented sub-basins in the lower portion of the Lancones basin (Fig. 1; Tegart et al.,  
124 2000; Winter et al., 2004, 2010). The formation of the Lancones basin is connected to the  
125 break-up of Gondwana and coincided with a period of intense extension and rifting due to  
126 trench-rollback (Winter et al., 2010 and their references). Minimum age of massive sulfide  
127 formation (and likely synchronous with or slightly postdating massive sulfide formation) is  
128 constrained by U-Pb zircon dating of felsic volcanoclastic rocks at  $104.8 \pm 1.3$  to  $100.2 \pm$   
129  $0.5$  Ma (middle Albian; Winter et al. 2010). The volcanic and sedimentary pile in the  
130 Lancones basin was intruded by Late Cretaceous to Tertiary arc-related granitoids of the  
131 Coastal Batholith (cf. Pitcher 1985; Atherton, 1990; Cobbing 1999; Hildebrand and  
132 Whalen, 2014).

133 In the central coastal segment of Peru, where the deposits studied in this paper are located,  
134 VMS mineralizations emplaced in the Huarmey and Cañete basins of the West Peruvian

135 Trough (WPT; [Fig. 1](#)) and are now exposed along the Peruvian Western Cordillera to the  
136 Coastal region. The opening of the WPT has been assigned to an aborted back-arc or true  
137 ocean floor extensional marginal basin (e.g., [Atherton et al., 1985](#); [Aguirre et al., 1989](#);  
138 [Atherton and Aguirre, 1992](#)) or to an intra-arc extensional pull-apart basin ([Soler, 1991](#);  
139 [Jaillard, 1994](#); [Polliand et al., 2005, 2006](#)). Stratigraphic relationships in the WPT led  
140 [Scherrenberg et al. \(2012\)](#) to delineate i) a western intra-arc basin, which is dominated by  
141 igneous rocks and that geographically coincides with the VMS mineralization; and ii) an  
142 eastern, deeper back-arc basin in which volcanic rocks overlie continental clastic rocks and  
143 marine carbonates. Major basin filling in the WPT occurred from latest Jurassic to Late  
144 Cretaceous ([Mégard, 1984](#); [Benavides-Cáceres, 1999](#)). The stratigraphy of the WPT (at  
145 least in its western portion *sensu* [Scherrenberg et al., 2012](#)) is largely composed of a  
146 volcano-sedimentary sequence including bimodal mafic-felsic volcanic rocks, limestones,  
147 shales, sandstones, and local evaporites that was intruded by the Peruvian Coastal  
148 Batholith, which in part is coetaneous with the volcano-sedimentary sequence ([Mégard,](#)  
149 [1984](#); [Atherton and Aguirre, 1992](#); [Vela, 1997](#); [Benavides-Cáceres, 1999](#); [Polliand et al.,](#)  
150 [2005, 2006](#); [Hildebrand and Whalen, 2014](#)). The stratigraphy of the WPT is complex and  
151 includes abrupt lateral facies changes greatly limiting the stratigraphic correlation and age  
152 estimates of the sedimentary units. Since the scope of this article is limited to the  
153 geochemical composition of some ore minerals in selected VMS deposits, we will not  
154 elaborate on the nomenclature of hosting units but will simply conform with ‘Casma  
155 Group’ embracing the Albian-Cenomanian volcano-sedimentary sequence versus the  
156 younger ‘Upper Cretaceous-Paleocene volcano-sedimentary sequence’ ([Romero et al.,](#)  
157 [2008](#); [Cueva et al., 2010](#); [Fontboté, 2019](#)).

158 From north to south, the most representative VMS mineralization in the Huarmey and  
159 Cañete basins are the María Teresa, Aurora Augusta, Perubar, Palma, and Cerro Lindo  
160 ([Fig. 1](#)). Geochronology has given two groups of ages for these deposits: uppermost  
161 Cretaceous and mid-Cretaceous. High-resolution U-Pb zircon geochronology performed by  
162 [Polliand et al. \(2005\)](#) on rhyolitic lavas at the bottom and top of massive ore bodies in  
163 Perubar yields dates of  $69.71 \pm 0.18$  and  $68.92 \pm 0.16$  Ma, and on intrusive rocks, dates of  
164  $67.91 \pm 0.17$  and  $67.89 \pm 0.18$  Ma. These dates ascribe the hosting rocks to the Upper  
165 Cretaceous-Paleocene volcano-sedimentary sequence and point to largely coeval

166 subsidence, submarine volcanism, and plutonic activity within the Andean magmatic arc  
167 during the Late Cretaceous (Polliand et al., 2005; Polliand, 2006). At María Teresa, a date  
168 of  $68 \pm 6$  Ma (Rb/Sr in hydrothermal sericite; Romero et al. 2008) points also to VMS  
169 mineralization in the latest Cretaceous. In contrast, the Cerro Lindo VMS mineralization is  
170 described to be hosted by a volcano-sedimentary sequence whose age has been attributed  
171 to the mid-Cretaceous (Casma Group) according to U-Pb zircon dating of host rhyolites at  
172  $105.7 \pm 0.2$  and  $106.9 \pm 0.2$  Ma (Gariépy and Hinojosa, 2003; Votorantim, 2017; Bueno  
173 Carreón, 2019; Bueno Carreón and Mendoza Mondragón, 2019). Recent U-Pb zircon  
174 dating at 101 Ma of a diorite dike crosscutting the mineralization suggests that the Palma  
175 VMS deposit is also mid Cretaceous (Choquehuanca, 2020). At Aurora Augusta,  
176 hydrothermal sericite yielded inconclusive – both mid- and late Cretaceous – ages (K-Ar;  
177 Vidal, 1987).

178 Hydrothermal alteration in the María Teresa, Perubar, and Palma typically comprises  
179 intense sericite  $\pm$  chlorite in the footwall and epidote + albite + chlorite + carbonates in the  
180 hanging wall and ‘distal’ domains of sulfide bodies (Fontboté, 2019). Sericite alteration is  
181 particularly well developed in the María Teresa deposit due to the exclusive igneous nature  
182 (i.e., plagioclase-rich) of the host rocks unlike in Perubar and Palma, where the presence of  
183 sedimentary rocks in the host sequence distorts the hydrothermal alteration halos. At María  
184 Teresa, the identification of a ‘prospective horizon’ outlined by an intense sericite  
185 alteration front has been interpreted as a paleo-horizon in which ascending hydrothermal  
186 fluids mixed with seawater in the volcanic pile in the sub-seafloor (Fontboté, 2019;  
187 Pichardo et al., 2019). Sulfur and Sr isotope data at Perubar support that mixing between a  
188 hydrothermal fluid (dominated by H<sub>2</sub>S) and seawater triggered the mineralization (Polliand  
189 et al., 1999; Polliand, 2006). The setting of the María Teresa, Palma, and Perubar in a  
190 plutono-volcanic arc and the likely involvement of acidic-oxidizing fluids indicated by  
191 alteration and ore mineral assemblages are used by Fontboté (2019) to argue for a  
192 magmatic origin of the ore-forming hydrothermal fluids.

### 193 2.1.2. Geology of the María Teresa deposit

194 The Zn-Pb-Cu-Ag María Teresa mine is located 11 km NW of Huaral and 85 km N of  
195 Lima (Fig. 1) and is owned and operated by Minera Colquisiri SA since 1985.



196 Mineralization at María Teresa (>18 Mt high-grade Zn-Pb-Ag±Cu ore, [Pichardo et al.,](#)  
197 [2019](#)) occurs as replacement of volcanic and subvolcanic rocks that were intruded by a  
198 series of subvertical, N150E-oriented andesitic and dacitic dykes and, in places, by  
199 granodiorite and diorite plutons. Volcanic rocks at María Teresa comprise basaltic  
200 andesites and andesites and less abundant rhyodacites which, akin plutonic rocks in the  
201 district, show calc-alkaline affinity ([Pichardo et al., 2019](#)). Basaltic andesites and andesites  
202 show aphanitic, porphyritic textures drawn by plagioclase and amphibole phenocrysts.  
203 Amygdaloidal and brecciated textures are also common and, towards the upper section of  
204 the sequence, pillow lavas and hyaloclastites occur. Despite that the limit between  
205 individual lava flows is often blurred, the locally observed bedding is horizontal to slightly  
206 W-dipping. The N160E-oriented El Abra strike-slip fault dismembered and displaced  
207 clockwise the mineralization in the ‘prospective horizon’ so that the Calera sector has been  
208 displaced about 500 m to the south in relation to the Mine 2 and Bubulina sectors. In  
209 addition, the N140E-oriented Sofía reverse fault has slid the western block on top of the  
210 eastern block with an approximate vertical displacement of 90 m ([Fig. 2](#)). The  
211 mineralization in the María Teresa deposit occurs mostly as N150E-oriented irregular to  
212 lenticular massive sulfide bodies and underlying stockwork structures associated with  
213 abundant alteration to sericite and quartz ([Pichardo et al., 2019](#)). Close to the contact with  
214 granodiorite intrusions chiefly to the east of the Calera zone, a mineral assemblage  
215 including cordierite and garnet porphyroblasts along with biotite, phlogopite, and tremolite  
216 is interpreted as the contact metamorphic product of the hydrothermal sericite alteration  
217 ([Pichardo et al., 2019](#)). The ore mineralization includes sphalerite, chalcopyrite,  
218 tetrahedrite-group minerals, and galena, which occur along with variable proportions of  
219 pyrite ± pyrrhotite ± barite. The Sofía-D massive sulfide body is the largest in the deposit  
220 (~ 8 Mt grading 7.1% Zn, 1.37% Cu, 0.64% Pb, 60.3 g/t Ag; [Pichardo et al., 2019](#)) and  
221 shows clear metal zoning ([Fig. 2](#)). Metal zoning, ore mineral assemblages, and textures  
222 observed in this ore body are described in detail below. Due to the well-defined metal  
223 zoning in Sofía-D, it was selected in this study to trace the distribution of trace elements in  
224 sphalerite and chalcopyrite in zone-refined massive sulfide bodies.

### 225 2.1.3. Geology of the Perubar deposit

226 The Zn-Pb-Ba Perubar deposit is located 50 km E of Lima (Fig. 1) and was exploited  
227 between 1978 and 2000 with a production of 6 Mt with an average grade of 11.8% Zn and  
228 1.4 % Pb (Polliand, 2006). The massive sulfide mineralization is hosted by an uppermost  
229 Cretaceous volcano-sedimentary sequence composed of submarine volcanic rocks and  
230 pyroclastic deposits of intermediate to felsic composition, intercalated with detrital  
231 sediments, volcanoclastic sandstones, tuffaceous mudstones, calcareous volcanoclastics, and  
232 limestones. Monzodiorite and monzogranite stocks belonging to the Coastal Batholith  
233 intruded in the area  $< \sim 1$  Myr after the mineralizing event causing contact metamorphism at  
234 the Perubar deposit (Polliand et al., 1999; see Fig. 3 in Polliand et al., 2005; Polliand, 2006).  
235 The hydrothermal alteration of the host rocks comprises a siliceous core in the stockwork  
236 zone that grades outward to chlorite + pyrite  $\pm$  sericite; extensive quartz + sericite developed  
237 in the footwall below the sulfide lenses, as well as distal quartz + K-feldspar. The deposit  
238 consists of four main massive barite-sulfide ore bodies (Graciela, Juanita, Cecilia Norte, and  
239 Cecilia Sur). According to Polliand (2006), the original (i.e., before tectonic deformation)  
240 ore zonation included a stockwork, which is now located north of the deposit, and massive  
241 mineralization that comprised pyrite(-magnetite) and pyrite-rich semi-black ores topped by  
242 a massive black ore, which graded laterally to banded barite-sulfide ores. The ore included  
243 pyrrhotite, pyrite, marcasite, magnetite, sphalerite, galena, barite, and lesser proportions of  
244 chalcopyrite, tetrahedrite, and arsenopyrite (Polliand, 2006). In addition, during our survey,  
245 tiny amounts of molybdenite and electrum were identified (Fig. S1). Representative  
246 photomicrographs of the samples containing the analyzed sphalerite from Perubar are shown  
247 in Figure S1. Polliand (2006) suggests that the conspicuously low proportion of chalcopyrite  
248 in the ore mineralization constrains the temperature of the mineralizing fluids  $< \sim 300$  °C.

#### 249 2.1.4. Geology of the Palma deposit

250 The Zn-Pb Palma deposit is located 55 km SE of Lima and 15 km south of Perubar (Fig. 1).  
251 Ongoing exploration by Volcan Cía. Minera since 2009 has revealed a resource of 14.5 Mt  
252 including 9.6 Mt indicated at 5% Zn, 0.7% Pb, and 22 g Ag/t and 4.9 Mt inferred at 6% Zn,  
253 1% Pb, and 21 g Ag/t, all reported with a cut-off of 3.5% Zn (Farfán et al., 2019). The  
254 mineralization is hosted in a volcano-sedimentary sequence consisting of sills and NNW-  
255 SSE trending dykes of basaltic to andesitic composition and local peperitic textures, organic

256 matter-rich and framboidal-pyrite-bearing pelites, and shallow-water limestones. This  
257 sequence is cut by NNW-SSE-oriented felsic dykes. The igneous suite in the district displays  
258 calc-alkaline affinity (Farfán et al., 2019). Hydrothermal alteration includes deep, poorly  
259 developed sericite alteration and much widespread albite + epidote + chlorite + pyrite  
260 alteration. The mineralization occurs as stacked massive sulfide bodies roughly parallel to  
261 the bedding and replaced chiefly limestone and subordinately pelitic and mafic and  
262 intermediate igneous rocks belonging to the lower section of the volcano-sedimentary  
263 sequence in the > 250-m-thick Palma Unit (see Fig. 3 in Farfán et al., 2019). Stringer and  
264 stockwork mineralization are also documented by these authors. Farfán et al. (2019) describe  
265 that the mineralogy of the upper semi-massive sulfide bodies comprises mainly pyrite,  
266 sphalerite, galena, and barite, with subordinate pyrrhotite and chalcopyrite, while the  
267 proportion of these two minerals increases in lower massive sulfide bodies. In addition,  
268 minor amounts of arsenopyrite and marcasite were identified during this survey. The samples  
269 analyzed in this work come from the upper semi-massive sulfide bodies. Representative  
270 microphotographs of samples containing the analyzed sphalerite from Palma are presented  
271 in Fig. S2. Ore and host rock have partly undergone contact metamorphism induced by arc  
272 plutons (Farfán et al., 2019).

## 273 2.2. Geology of the Cerro de Maimón deposit, Dominican Republic

274 The Cerro de Maimón VMS deposit is located 70 km northwest of Santo Domingo in central  
275 Dominican Republic (Fig. 3). It stands as the only VMS deposit currently under production  
276 in the Caribbean region. It is owned and operated by Perilya Limited through its subsidiary  
277 Corporación Minera Dominicana (CORMIDOM). The Cerro de Maimón massive sulfide  
278 body contains 4.8 Mt of proven and probable reserves grading 2.54% copper, 0.96 g/t Au,  
279 and 34.9 g/t Ag (Perilya, 2020). VMS mineralization and the metallogenic evolution of the  
280 Caribbean Greater Antilles are summarized and discussed by Nelson et al. (2011) and Torró  
281 et al. (2018a). The Cerro de Maimón massive sulfide mineralization is hosted by the Aptian  
282 Maimón Formation (Fig. 3), which is composed mostly of bimodal mafic-felsic volcanic and  
283 volcanoclastic rocks of boninitic and tholeiitic affinities emplaced in the forearc portion of  
284 the nascent Caribbean island-arc (Torró et al., 2016a, 2017, 2018b). Massive sulfides and  
285 hosting volcanic rocks were deformed and metamorphosed to peak metamorphic conditions

286 in the greenschist/blueschist facies transition (Torró et al., 2016b). The mineralization is  
287 composed of pyrite grains with both cataclastic/blow-apart and metablastic welding textures  
288 and an interstitial assemblage of chalcopyrite + sphalerite + quartz + tennantite ± galena  
289 (paragenetic sequence and photomicrographs available in Fig. 6 in Torró et al., 2016a).

### 290 3. Materials and Methods

291 The study is based on 36 ore samples from the María Teresa (n = 24), Palma (n = 4), and  
292 Perubar (n = 5) deposits in Peru and the Cerro de Maimón deposit (n = 3) in the Dominican  
293 Republic. A list of the samples, their location, and brief textural descriptions are given in  
294 Table S1 in the Supplementary Material. In the María Teresa deposit, 15 ore samples  
295 picked up at different depths along a single drill core (34-18-S) that cut the zone-refined  
296 Sofía-D sulfide body (Calera sector) have been used to monitor mineralogical and mineral  
297 geochemical variations with depth. A total of 34 thick polished sections were produced at  
298 the QEMSCAN laboratory of the Pontificia Universidad Católica del Perú (PUCP) and  
299 examined using petrographic and scanning electron (SEM) microscopy. SEM images and  
300 major element mineral compositions were acquired on an environmental SEM Thermo  
301 Fisher Quanta 650 FEI equipment with an EDAX-Octane Pro EDS microanalysis system at  
302 PUCP (Centro de Caracterización de Materiales – CAM-PUCP) and a VEGA II XMU  
303 Tescan equipment with a Bruker QUANTAX EDS microanalysis system at BIZALAB at  
304 operating conditions of 5 nA and 20 kV accelerating voltage.

305 Minor and trace element compositions of sphalerite and chalcopyrite were determined by  
306 means of electron microprobe (EPMA), at the University of Barcelona, and laser-ablation  
307 inductively-coupled plasma mass spectrometry (LA-ICP-MS), at the ETH Zürich, using  
308 the same equipment and methodology as described in Benites et al. (2021a). For samples  
309 from the María Teresa, Palma, and Perubar deposits, representative EPMA analyses of  
310 sphalerite and chalcopyrite are displayed in Table S2 in Supplementary Material. For  
311 samples from the Cerro de Maimón deposit, representative EPMA analyses of sulfide  
312 minerals are provided in Torró et al. (2016a). LA-ICPMS data are reported in Table S3 in  
313 Supplementary Material. Content statistical data for sphalerite and chalcopyrite are shown  
314 in Tables 1-2.

## 315 4. Results

### 316 4.1. Ore mineral geochemistry

317 Major and trace element compositions of sphalerite and chalcopyrite show, in general, a  
318 wide range of elements at measurable contents. Summaries of their compositions,  
319 including minimum, maximum, geometric mean, standard deviation, and interquartile  
320 range (IQR) values are displayed in [Tables 1](#) and [2](#). Content values will be hereinafter  
321 reported as IQR unless otherwise specified. Significant effort was made to report only the  
322 content of elements whose variations respond to solid solutions and not to mixed mineral  
323 analyses by analyzing mineral volumes free of obvious inclusions and by selecting only  
324 stable signal intervals in LA-ICP-MS spectra ([Fig. S3](#) in Supplementary Material).  
325 However, it should be noted that evenly distributed nano-inclusions may also yield flat  
326 ablation profiles (e.g., [Gregory et al., 2015](#)) and that, therefore, the possibility of mixed  
327 mineral analyses cannot be completely discarded due to the relatively low spatial  
328 resolution of the used methods.

#### 329 4.1.1. Sphalerite

330 Iron, Mn, Cd, Hg, and Cu ([Fig. S3](#) in Supplementary Material) yield smooth and constant  
331 ablation profiles. Studied sphalerite shows a relatively wide range of Fe contents in samples  
332 from the María Teresa deposit (40675-17122 ppm), which are higher than in samples from  
333 the Cerro de Maimón (585-535 ppm) and generally lower than in samples from the Perubar  
334 (67570-39965 ppm) and Palma (63423-30015 ppm) deposits ([Table 1](#)). Manganese also  
335 shows a relatively wide range of contents in sphalerite from María Teresa (2343-326 ppm),  
336 which overlaps with the composition of grains from Cerro de Maimón (482-446 ppm) and  
337 are in general lower than in grains from Perubar (5964-907 ppm) and Palma (7751-6734  
338 ppm). The contents of Cd are constantly high in all analyzed sphalerite samples (7373-5290  
339 ppm in María Teresa, 6665-4297 ppm in Perubar, 7109-4782 ppm in Palma, and 3829-3574  
340 ppm in Cerro de Maimón). Mercury contents are higher in sphalerite from the Palma (667-  
341 593 ppm) and María Teresa (448-326 ppm) deposits than in sphalerite from the Perubar  
342 (107-18 ppm) and Cerro de Maimón (26-22 ppm) deposits. The content of Cu is generally  
343 higher in sphalerite samples from the Cerro de Maimón (138-113 ppm) and María Teresa

344 (98-36 ppm) deposits than in sphalerite samples from the Perubar (41-25 ppm) and Palma  
345 (31-11 ppm) deposits.

346 At generally lower contents and also showing smooth ablation spectra are Ag, In, Ge, and  
347 Ga (Table 1; Fig. S3 in Supplementary Material). Silver content ranges mostly overlap in  
348 sphalerite grains from the four studied deposits (12-4.2 ppm in María Teresa, 7.1-3.6 ppm  
349 in Perubar, 17-3.8 ppm in Palma, and 4.2-3.1 ppm in Cerro de Maimón). The contents of In  
350 are in general much higher in sphalerite from the María Teresa deposit (68-4.4 ppm, up to  
351 415 ppm) than in sphalerite from the Cerro Maimón (10-9.5 ppm), Perubar (2.3-0.78 ppm)  
352 and Palma (0.22-0.0091 ppm; Table 1) deposits. Gallium contents are higher in sphalerite  
353 from the Cerro de Maimón deposit (95-78 ppm, up to 96 ppm) than in sphalerite from the  
354 María Teresa (32-11 ppm, up to 94 ppm), Perubar (30-4.7ppm), and Palma (2.4-1.9 ppm)  
355 deposits. Germanium contents are in general very low in analyzed sphalerite grains (0.65-  
356 0.17 ppm in María Teresa, 0.19-0.15 ppm in Perubar, 3.5-0.16 ppm in Palma, and 1.2-0.68  
357 ppm in Cerro de Maimón).

358 The content of other elements are in general very low (Pb: 9.3-1.1 ppm; Sb: 3.7-0.31 ppm;  
359 Tl: 0.66-0.045; Sn: 0.49-0.23 ppm; Bi: 0.60-0.0055 ppm; detailed results for individual  
360 deposits are available in Table 1). Some analyses show slightly ragged ablation spectra  
361 (Fig. S1 in Supplementary Material).

#### 362 4.1.2. Chalcopyrite

363 Chalcopyrite grains from the María Teresa and Cerro de Maimón VMS deposits have been  
364 analyzed. Zinc, Ag, Sn, Cd, Se, In, Ge, and Ga show relatively constant ablation profiles  
365 (Fig. S3 in Supplementary Material). Zinc is the trace element most highly concentrated in  
366 studied sphalerite grains; its contents are higher in chalcopyrite from the María Teresa  
367 deposit (425-194 ppm) than in chalcopyrite from the Cerro de Maimón deposit (a single  
368 analysis >d.l. yields a concentration of 71 ppm; Table 2). Silver, Sn, and Cd are relatively  
369 enriched in chalcopyrite grains from the María Teresa deposit (150-71 ppm Ag, 77-33 ppm  
370 Sn, 12-6.2 ppm Cd) relative to grains from the Cerro de Maimón deposit (23-15 ppm Ag,  
371 1.6-1.4 ppm Sn, 0.53-0.41 ppm Cd). Selenium is more concentrated in chalcopyrite from  
372 the Cerro de Maimón deposit (89-79 ppm) than in chalcopyrite from the María Teresa

373 deposit (31-19 ppm). Indium values are relatively constant in analyzed grains from the  
374 studied deposits and are systematically higher in chalcopyrite from María Teresa (32-12  
375 ppm) than in chalcopyrite from Cerro de Maimón (8.1-7.3 ppm). The contents of Ga and  
376 Ge are in general lower than those of In (9.2-2.6 ppm Ga and 4.3-0.66 ppm Ge in María  
377 Teresa; 3.6-3.3 ppm Ga and 6.1-5.3 ppm Ge in Cerro de Maimón).

378 Downhole ablation spectra reveal very irregular profiles for Mo and Au and in some  
379 samples, slightly irregular profiles for Tl, Sb, and Bi (Fig. S3 in Supplementary Material).  
380 Molybdenum yields relatively erratic contents, which are much higher in analyzed  
381 chalcopyrite from Cerro de Maimón (83-23 ppm) than in chalcopyrite from María Teresa  
382 (0.034-0.021 ppm, and mostly <d.l.). The contents of Tl (< 11 ppm), Sb (< 31 ppm), Bi (<  
383 10 ppm), and Au (< 0.40 ppm) are relatively low (Table 2).

#### 384 **4.2. Metal zonation, mineralogy and trace-element distribution patterns in the zone-** 385 **refined Sofía-D body**

386 The metal zonation in Sofía-D massive sulfide body includes, bottom to top, i) a Cu zone  
387 composed dominantly of chalcopyrite and lesser amounts of sphalerite grading downward  
388 to massive pyrite; ii) a Zn zone in which sphalerite is the most abundant phase and appears  
389 along with lesser amounts of galena and chalcopyrite and minor fahlore; and iii) a Pb – Zn  
390 – Ag zone with abundant galena and sphalerite, and lesser proportions of fahlore and barite  
391 (Fig. 2).

392 On the hand sample scale, sphalerite occurs as mm- to cm-sized grains of yellow-pale  
393 brown (Fig. 4A-C) and red-brownish (Fig. 4D-E) color. In places, the massive sphalerite  
394 aggregates are cut by irregular, chalcopyrite-rich veinlets (Fig. 4A, D-E). In the eastern part  
395 of the María Teresa deposit, i.e., close to the granodiorite stock as is the case for the Sofía  
396 D ore body, coarse-grained pyrrhotite ± magnetite is observed and, similarly to in Perubar  
397 (see Vidal, 1987; Polliand et al., 1999), interpreted as the metamorphic product after pyrite.  
398 An example is shown in Figure 4F coming from a 20-cm-wide band composed of coarse-  
399 grained pyrrhotite ± magnetite at the contact with and andesitic dyke. In volcaniclastic  
400 rocks in the immediate host of the Sofía-D body, chalcopyrite selectively replaced some of

401 the clasts and the matrix (Fig. 4G). Silicified dykes that cut the Sofia-D body typically  
402 contain patches and irregular veinlets of chalcopyrite (Fig. 4H).

403 Under the petrographic microscope, pyrite grains, which have sizes typically in the range  
404 between 100 and 600  $\mu\text{m}$ , draw a variety of textures including anhedral isolated grains with  
405 roundish and embayed outlines, aggregates of several grains often with foam equilibrium  
406 textures (i.e., contacts at  $120^\circ$ ) and, in some cases, internal irregular fractures (Fig. 5A-D).  
407 Arsenopyrite is relatively scarce in the studied samples and appears as isolated euhedral  
408 and subhedral fractured grains and sizes in the range between 50 and 200  $\mu\text{m}$  (Fig. 5C-D).  
409 The other sulfides, fahlore (tennantite-(Fe)  $\pm$  argentian tetrahedrite), and traces of other  
410 sulfosalts such as bournonite, boulangerite, and polybasite (petrographic observations  
411 complemented with EDS analyses) occupy interstitial space between, and line porosity and  
412 fractures within, pyrite  $\pm$  arsenopyrite grains (Fig. 5). According to these observations,  
413 pyrite and arsenopyrite crystallized early in the paragenetic sequence (Fig. 6) and, in places,  
414 underwent thermal re-equilibration connected to dyke and granodiorite intrusions (Pichardo  
415 et al., 2019). Internal fractures of pyrite and arsenopyrite (Fig. 5C) also point to fragile  
416 deformation, probably due to movements along the Sofia reverse fault (Fig. 2). Sphalerite  
417 grains in the Zn zone show clean surfaces (Fig. 5B,E), whereas in the Cu zone they have  
418 been pervasively replaced by chalcopyrite and show dense chalcopyrite disease textures  
419 (Fig. 5D, F-G). Chalcopyrite floods the interstitial space between pyrite  $\pm$  arsenopyrite in  
420 the Cu zone along with relicts of sphalerite and in places shows sharply bounded  
421 exsolutions of cubanite laths up to 150  $\mu\text{m}$  in length (Fig. 5G). Auriferous silver (73 wt.%  
422 Ag, 27 wt.% Au, EDS analysis) has been observed in one sample and appears as minute  
423 grains ( $< 2 \mu\text{m}$ ) in fractured arsenopyrite along with chalcopyrite (Fig. 5H).

424 The systematic sampling developed along the drill core 34-18-S from 72.85 to 128 m  
425 (Table S1 in the Supplementary Material) allows for the assessment of vertical trends in  
426 trace-element contents in sphalerite and chalcopyrite. Bottom to top in sphalerite, there is a  
427 relatively sustained decrease in the contents of Mn, Cu, Se, Bi, and In and an increase of  
428 Ge, and the contents of elements such as Cd and Ga show irregular distributions (Fig. 7A-  
429 G). The contents of In significantly decrease from sample 2019-MT-015 (128 m, 167-158  
430 ppm) to sample 2019-MT-001 (72.85 m, 4.4-4.1 ppm), although the maximum contents



431 occur in sample 2019-MT-004 (81.4 m, 409-399 ppm; [Fig. 7G](#)). The distribution of Cu  
432 contents almost mimics that of In, with a decrease from sample 2019-MT-015 (128 m, 203-  
433 160 ppm) to sample 2019-MT-001 (72.85 m, 27-24 ppm), although with maximum  
434 contents at intermediate depths in samples 2019-MT-004 (81.4 m, 335-321 ppm) and 2019-  
435 MT-011 (105.5 m, 597-115 ppm; [Fig. 7C](#)). Germanium contents show a moderately  
436 irregular distribution pattern with a general increase in its content from sample 2019-MT-  
437 015 (128 m, 0.33-0.23 ppm) to sample 2019-MT-002 (74.6 m, 1.3-0.73 ppm; [Fig. 7I](#)). As  
438 for chalcopyrite, vertical distribution patterns are mostly irregular ([Fig. 8](#)). The most  
439 remarkable exception is Se, which content tends to decrease steadily at shallower levels (45  
440 ppm – single value – in sample 2019-MT-015 at 128 m, 31-27 ppm in sample 2019-MT-  
441 012 at 109.4 m, 11-10 ppm in sample 2019-MT-007 at 90.5 m; [Fig. 8E](#)). An opposite trend  
442 is defined by the content of Ge, with general enrichment from deeper (2.2 ppm – single  
443 value – in sample 2019-MT-015 at 128 m, 0.84-0.60 ppm in sample 2019-MT-012 at 109.4  
444 m,) to shallower (80-74 ppm in sample 2019-MT-007 at 90.5 m) chalcopyrite samples ([Fig.](#)  
445 [8J](#)). Although probably hosted as sub-microscopic mineral inclusions, the content of Bi in  
446 chalcopyrite also shows significant variation and tends to increase with depth ([Fig. 8G](#)).

## 447 **5. Discussion**

### 448 **5.1. Trace-element budget in sphalerite and chalcopyrite from arc-hosted VMS** 449 **deposits**

450 The minor and trace element contents measured in sphalerite and chalcopyrite from the  
451 Palma, Perubar, María Teresa (Perú), and Cerro de Maimón (Dominican Republic) VMS  
452 deposits will be compared along the following lines with available data from other VMS  
453 deposits, and particularly with those formed in arc systems. We have compared our results  
454 with the comprehensive compilation by [Monecke et al. \(2014\)](#) of whole-rock and mineral  
455 geochemical data from both active seafloor massive sulfide (SMS) and fossil equivalent  
456 deposits emplaced in a variety of tectonic settings worldwide.

457 Sphalerite from the four investigated deposits yields fairly variable Fe contents ([Table 1](#)),  
458 the content of the other analyzed trace elements being largely subordinated to that of Fe. The  
459 content of Fe in sphalerite samples from the Cerro de Maimón deposit analyzed in this study

460 is particularly low (Table 1); a more complete geochemical dataset based on EPMA analyses  
461 on sphalerite samples from this deposit reported by [Torró et al. \(2016a\)](#) indicates average Fe  
462 contents of 1.16 wt.%. Iron contents in the studied deposits are compatible with sphalerite  
463 crystallized in intermediate to high-intermediate sulfidation environments in equilibrium  
464 with pyrite ([Barton and Toulmin 1964](#); [Hannington and Scott, 1989](#)). Co-crystallization of  
465 fahlore (tennantite and tetrahedrite) along with sphalerite in the studied ore samples is a  
466 further indication of intermediate-sulfidation state of the hydrothermal mineralizing fluids.  
467 Similar Fe contents to those in the studied Andean VMS deposits are reported in sphalerite  
468 samples from the Zinkgruvan, Marketorp, and Sauda VMS deposit (Table 3 and references  
469 therein). Iron contents similar to those found in sphalerite from the Cerro de Maimón deposit  
470 are reported in the Vorta, Eskay Creek, and Çayeli VMS deposits. In contrast, sphalerite  
471 from the VMS Laochang deposit (South China) systematically yields higher Fe contents.

472 The relatively high content of Cd in sphalerite from the studied deposits (Table 1) are  
473 comparable with those in sphalerite from other VMS deposits such as Vorta, Eskay Creek,  
474 Zinkgruvan, Marketorp, and Laochang, and higher than in Sauda (Table 3). Cadmium  
475 contents of several hundreds of parts per million are typical of SMS emplaced in arc settings  
476 compared to contents in the range of some tens of parts per million in those emplaced in  
477 mid-ocean ridge systems, sphalerite and wurtzite being the main Cd hosts in VMS deposits  
478 according to [Monecke et al. \(2014\)](#). According to our (Table 1) and previous (Table 3) data,  
479 Cd in sphalerite from VMS deposits in arc systems typically yields contents of several  
480 thousands of parts per million. The content of Hg varies between some tens (Perubar and  
481 Cerro de Maimón) and some hundreds (María Teresa and Palma) of parts per million in the  
482 studied sphalerite samples; the content of Hg is not provided for sphalerite from other VMS  
483 deposits in arc systems in the revised literature. [Petersen et al. \(2014\)](#) reported contents  
484 between <0.02 and 7.4 wt.% Hg (median = 0.37 wt.% Hg; EPMA data) in sphalerite from  
485 the Palinuro SMS deposit in the Aeolian arc (transitional arc setting). [Monecke et al. \(2014\)](#)  
486 found out that the content of Hg in SMS is higher in mineralization emplaced in arc  
487 volcanoes and rifted arcs (some tens of ppm) relative to mineralization emplaced in back-  
488 arc spreading centers and mid-ocean ridges; also, that Hg usually shows a strong positive  
489 correlation with Sb in mineralized samples from SMS which normally results from the chief  
490 concentration of Hg in tetrahedrite and tennantite over other ore minerals. The contents of

491 Sb in the studied sphalerite are mostly in the 1-10 parts per million range or below ([Table](#)  
492 [1](#)), similar to sphalerite from other VMS mineralization such as Vorta, Zinkgruvan, and  
493 Laochang, and lower than in Eskay Creek and Çayeli ([Table 3](#)). The content of Tl in the  
494 studied sphalerite samples is normally below 1 ppm ([Table 1](#)), similar to sphalerite from  
495 other VMS deposits with the exception of Çayeli, in which the content of Tl is much higher  
496 (13-4.9 ppm; [Table 3](#)). The contents of Sb and Tl are usually higher in sulfide samples from  
497 SMS mineralization emplaced in arc volcanoes and rifted arcs than in SMS mineralization  
498 emplaced in back-arc spreading centers and mid-ocean ridges according to data compiled by  
499 [Monecke et al. \(2014\)](#).

500 The studied sphalerite samples yield In contents largely in the range between <1 ppm and  
501 some tens of parts per million ([Table 1](#)), similar to sphalerite from Vorta, Eskay Creek,  
502 Zinggruvan, Marketorp, and Sauda, and lower than in Laochang ([Table 3](#)). The measured In  
503 contents are also much lower than in sphalerite from the Neves Corvo VMS deposit (up to  
504 2.1 wt.%, EPMA data: [Carvalho et al., 2018](#)). The contents of Ga in the sphalerite samples  
505 analyzed here are largely of the same order of magnitude (from a few ppm to ~100 ppm) as  
506 contents reported for sphalerite from other VMS deposits ([Table 3](#)), whereas the contents of  
507 Ge (mostly <1 ppm) are generally lower than those reported in the literature. The contents  
508 of In, Ge, and Ga are mostly indistinguishably low in SMS sulfides irrespective of their  
509 tectonic environment of emplacement ([Monecke et al., 2014](#)).

510 In the studied chalcopyrite samples from the Andean VMS deposits, the content of Zn is of  
511 some hundreds of parts per million ([Table 2](#)), similar to chalcopyrite from Vorta, Sulitjelma,  
512 and Kutlular, and lower than in chalcopyrite from Çayeli ([Table 4](#) and references therein);  
513 Zn content in sphalerite from the Cerro de Maimón deposit is significantly lower. The  
514 contents of Ag are mostly in the range between a few to some hundreds of parts per million  
515 in the analyzed chalcopyrite ([Table 2](#)), like chalcopyrite from VMS reported in the literature  
516 ([Table 4](#)). Selenium contents in the studied chalcopyrite samples are of some tens of parts  
517 per million ([Table 2](#)), and therefore lower than Se contents in chalcopyrite from other  
518 subduction-related VMS deposits, which are mostly in the hundreds of parts per million  
519 range ([Table 4](#)). According to [Monecke et al. \(2014\)](#), chalcopyrite is the main host of Se in  
520 VMS and SMS deposits along with pyrite and, typically, higher Se contents are found in

521 sulfide samples collected in SMS formed in mid-ocean ridges relative to those in subduction  
522 related systems. Cadmium shows contents of a few to some tens of parts per million in the  
523 analyzed chalcopyrite samples (Table 2), similar to chalcopyrite from other VMS deposits  
524 (Table 4); therefore, as expected, Cd contents are much lower in chalcopyrite than in  
525 sphalerite from the same deposit (Tables 1 and 3). Similarly, Hg contents in chalcopyrite (up  
526 to a few ppm, Table 2) are much lower than in sphalerite (Table 1) from the respective  
527 deposits, and similar to or slightly higher than Hg content in chalcopyrite from the Vorta and  
528 Sulitjelma VMS deposits (Table 4).

529 Indium content in analyzed chalcopyrite grains, mostly at the level of tens of parts per  
530 million, is only slightly lower than In content in sphalerite grains from the same deposits  
531 (Tables 1 and 2), and steadily higher than in chalcopyrite grains from the Vorta and  
532 Sulitjelma VMS deposits (Table 4). Germanium content in the analyzed chalcopyrite  
533 samples is mostly at the level of some parts per million (Table 2) and usually higher than in  
534 sphalerite from the respective deposits (Table 1); Ge content is not provided in reference  
535 data for chalcopyrite from other VMS deposits (Table 4). The content of Ga (mostly at the  
536 level of a few ppm to some tens of ppm) is lower in chalcopyrite than in sphalerite from the  
537 same deposit (Tables 1 and 2); the content of Ga in chalcopyrite from the Vorta and  
538 Sulitjelma VMS deposits is systematically lower than in the samples analyzed here (Table  
539 4).

## 540 **5.2. Correlation trends and substitution mechanisms**

### 541 5.2.1. Sphalerite

542 The dataset shows a marked negative correlation trend between Fe and Zn with a slope of  
543  $\sim 1$  (Fig. 9A) that indicates a dominant  $\text{Fe}^{2+} \leftrightarrow \text{Zn}^{2+}$  simple isovalent substitution. Fair  
544 positive correlations between Ag and both Pb (Fig. 9B) and Sb (Fig. 9C) suggest that these  
545 three elements likely occur in solid solution in the studied sphalerite samples. Tin is also  
546 positively correlated with Ag and the reasonable positive correlation in the Sb+Sn vs. Ag+Pb  
547 plot (Fig. 9D) with a slope  $\sim 1$  suggests incorporation of these elements via coupled  
548 substitutions of the type  $\text{Ag}^+ + \text{Pb}^{2+} + (\text{Sn},\text{Sb})^{3+} \leftrightarrow 3\text{Zn}^{2+}$  and likely also via independent  
549 variations of this scheme (Cook et al., 2009). Nevertheless, the observation of slight ragged

550 downhole ablation spectra for Sn and Pb in a few analyses (e.g., Fig. S3D in Supplementary  
551 Material) might suggest the local presence of sub-microscopic inclusions of galena and/or  
552 Sn-Pb sulfosalts.

553 The dataset shows a marked correlation trend between Cu and In in samples from María  
554 Teresa, particularly in those with the highest In contents, with a slope  $\sim 1$  (Fig. 10A) thus  
555 pointing to a  $\text{Cu}^+ + \text{In}^{3+} \leftrightarrow 2\text{Zn}^{2+}$  coupled substitution (Cook et al., 2012; Trofimov et al.,  
556 2020; Benites et al., 2021a,b). Gallium yields a fair positive correlation with Cu with some  
557 analyses plotting along or close to the 1:1 correlation line (Fig. 10B) suggesting a  $\text{Cu}^+ +$   
558  $\text{Ga}^{3+} \leftrightarrow 2\text{Zn}^{2+}$  coupled substitution. Germanium contents do not show clear correlation  
559 trends with the other analyzed elements. The alignment of the analyses in the  
560 Sn+In+Ge+Ga vs. Cu + Ag plot ( $R = 0.55$ ; Fig. 10C) would suggest supplementary  
561 complex substitution mechanisms involving combinations of coupled substitutions of  
562 trivalent cations (i.e., In, Ga) with monovalent cations (i.e., Cu, Ag) and direct substitution  
563 of quadrivalent cations (Sn, Ge) with vacancies for Zn. However, the lacking consensus on  
564 the oxidation state of Ge (+2 or +4) and Sn (+3 or +4) in sphalerite leaves open the  
565 possibility of other substitution schemes (see Cook et al., 2009, 2015; Bellisont et al.,  
566 2014). The low content of both Sn and Ge in the studied sphalerite grains and the apparent  
567 lack of individual correlation trends with other analyzed elements do not allow to give a  
568 definitive conclusion on their incorporation in the crystal lattice.

### 569 5.2.2. Chalcopyrite

570 Unlike sphalerite, chalcopyrite is not a fully ionic mineral but bonding between atoms in its  
571 structure is “effectively” covalent with valencies of Fe fluctuating between divalent and  
572 trivalent and valencies of Cu, between monovalent and divalent (Donnay et al., 1958; Hall  
573 and Stewart, 1973; Li et al., 2013). Therefore, the incorporation mechanism of trace  
574 elements in chalcopyrite is far more complex to draw than for sphalerite (see George et al.,  
575 2018). Weak to fair positive correlation trends between Zn, Cd, In, Ge, and Ga are observed  
576 across our dataset (Fig. 11). George et al. (2018) also noticed the positive correlation  
577 between Zn and Cd in chalcopyrite samples from a variety of mineralization styles and  
578 argued that systematic variations of Cd:Zn ratio are strongly dependent on crystallization

579 temperature (this question is further addressed in the following section). Our data would  
580 suggest that the incorporation of Zn, Cd, In, Ge, and Ga to the structure of chalcopyrite is  
581 not completely independent (i.e., simple substitution) but that coupled substitutions  
582 involving cations of these elements are likely (see also [George et al., 2018](#) and their  
583 references).

### 584 **5.3. Trace element distribution profiles and conditions of crystallization (and re-** 585 **crystallization)**

586 Metal and mineral zonation ([Fig. 2](#)) and the overprinting and replacement of earlier sulfides  
587 (e.g., sphalerite) by minerals that are stable at higher temperature (e.g., chalcopyrite; [Figs.](#)  
588 [4-6](#)) as described for Sofía-D in the María Teresa deposit denote a thermally intensifying  
589 hydrothermal system during the growth of the massive sulfide body; in VMS systems, this  
590 process is commonly known as zone refining ([Eldridge et al., 1983](#); [Hannington et al.,](#)  
591 [1998](#); [Grant et al., 2015](#); [Almodóvar et al. 2019](#)). According to the GGIMFIS  
592 geothermometer of [Frenzel et al. \(2016\)](#) – which uses the contents of Fe, Mn, In, Ga, and  
593 Ge in sphalerite, the analyzed sphalerite from the María Teresa deposit crystallized at  
594 temperatures in the range between 320 and 257 °C (IQR). Such a temperature range is in  
595 good agreement with a VMS mineralization containing fair abundant chalcopyrite  
596 ([Hannington et al., 1995](#); [Ohmoto, 1996](#); [Franklin et al., 2005](#)). Assuming that  $fO_2$  and  $fS_2$   
597 were stable during the crystallization of sphalerite (see [Hannington and Scott, 1989](#) and  
598 [Keith et al. 2014](#)) in the Sofía-D sulfide body, sphalerite composition would record a  
599 relatively sustained decrease in the crystallization temperature bottom to top (335-327 °C  
600 for sample 2019-MT-015 at 128 m to 255-249 °C for sample 2019-MT-002 at 74.6 m; [Fig.](#)  
601 [7J](#)). This decrease in the calculated crystallization temperature is actually a numerical  
602 translation of the depletion in In and Mn (directly proportional to the GGIMFIS  
603 temperature) and the enrichment in Ge (inversely proportional to the GGIMFIS  
604 temperature) bottom to top in sphalerite of the Sofía-D massive sulfide body ([Fig. 7](#)).  
605 Noteworthy, an increase in the availability of Cu in the mineralizing fluid towards the Cu  
606 zone probably also enhanced the incorporation of In to the sphalerite crystal lattice through  
607 the coupled substitution mentioned above ( $Cu^+ + In^{3+} \leftrightarrow 2Zn^{2+}$ ) during zone refining  
608 processes.

609 The Cd:Zn ratio in chalcopyrite is commonly used to trace variations in physicochemical  
610 conditions during its crystallization. Higher Cd:Zn ratios are typically ascribed to higher  
611 temperatures of crystallization (George et al. 2018). In the Sofía-D sulfide body,  
612 chalcopyrite shows rather constant Cd:Zn ratios at different depths and sections (Fig. 8K)  
613 suggesting relatively homogeneous physicochemical conditions during the crystallization of  
614 this mineral throughout the Cu zone of the mineralized body (Fig. 2). In contrast, the  
615 increase of Se contents in chalcopyrite (and sphalerite) with depth would suggest an  
616 increase in the temperature and/or magmatic volatile influx (Auclair et al., 1987; Huston et  
617 al., 1995; Layton-Matthews et al., 2008; Martin et al. 2019). The concentration of Se, and  
618 most trace elements indeed, in the structure of chalcopyrite is distinguishingly strongly  
619 influenced by the co-crystallizing phases so that the continuous crystallization of galena  
620 during zone refining probably partitioned Se and depleted chalcopyrite in this element  
621 (Cook, 1996; Carvalho et al., 2018; George et al. 2018). The progressively increasing  
622 abundance of galena towards the upper sections of the Sofía-D body may therefore explain  
623 the progressive depletion of Se in chalcopyrite at shallower levels.

624 In short, our results suggest that in zone-refined VMS mineralization such as the Sofía-D  
625 massive sulfide body in the María Teresa deposit, chalcopyrite and particularly sphalerite  
626 register systematic changes in their trace element compositions across the mineralized  
627 body. Bottom to top of the sulfide body, observed trends include i) in sphalerite, depletion  
628 in In, Cu, Mn, Se, and Bi and enrichment in Ge; and ii) in chalcopyrite, depletion in Se and  
629 Bi (the latter probably in microinclusions). Furthermore, these patterns agree with expected  
630 physicochemical trends involving decreasing temperatures and influx of hydrothermal  
631 fluids with a direct magmatic contribution towards the margins of the sulfide body (Auclair  
632 et al., 1987; Huston et al., 1995; Layton-Matthews et al., 2008; Yeats et al., 2014; George et  
633 al. 2018; Schmidt et al., 2018; Martin et al. 2019).

634 Ore mineralization at the Sofía-D massive sulfide body in the María Teresa deposit  
635 underwent contact metamorphism connected to nearby granodiorite intrusions (Fig. 2). The  
636 identification of key trace-element distribution patterns in sphalerite and chalcopyrite  
637 across this body and their reconciliation with expected original trends suggest that contact  
638 metamorphism occurred in a closed system and more importantly, that body-scale trace

639 element distribution was largely unaffected. Similarly, Polliand et al. (1999) determined  
640 that in the Perubar VMS deposit, sulfur isotope re-equilibration between sulfate-sulfide  
641 pairs during contact metamorphism occurred in a closed system and only between adjacent  
642 mineral grains. Lockington et al. (2014) found out that the content of lattice-bound trace  
643 elements in sphalerite (e.g., Mn, Fe, Cd, Cu, In, Ge, Hg, and Se) from metamorphosed  
644 VMS deposits worldwide showed no correlation with the metamorphic grade but were  
645 defined primarily by physicochemical conditions during initial deposition; in contrast, the  
646 contents of elements that typically occur as micro- and nano-inclusions in sphalerite (e.g.,  
647 Pb and Bi) decreased with increasing metamorphic grade. Therefore, the use of trace  
648 element patterns defined by Mn, In, Ge, and Se in sphalerite, and of Se in chalcopyrite, are  
649 likely to reflect primary trends.

## 650 **6. Conclusions and implications**

651 In this study, systematic trace element determinations have been conducted in sphalerite  
652 and chalcopyrite from the Cretaceous arc-related Palma, Perubar, and María Teresa VMS  
653 deposits in Peru and the Cerro de Maimón VMS deposits in the Dominican Republic.  
654 Sphalerite from the four deposits incorporated Fe, Mn, Cd, Hg, Ag, Sb, Se, In, Ge, and Ga  
655 in solid solution. The incorporation of other elements such as Sn, Tl, Bi, and Pb possibly  
656 resulted in some cases from microinclusions as suggested by occasional ragged ablation  
657 spectra. The contents of minor and trace elements in analyzed sphalerite show significant  
658 variations even between samples from any individual deposit. Iron contents (44009-18168  
659 ppm) correlate negatively with Zn suggesting a dominant  $\text{Fe}^{2+} \leftrightarrow \text{Zn}^{2+}$  simple isovalent  
660 substitution. Cadmium (7082- 4972 ppm), Mn (4578-140 ppm), and Hg (453-107 ppm) are  
661 much concentrated than Pb (9.5-1.1 ppm), Sb (3.7-0.31 ppm), Tl (0.70-0.045 ppm), and Bi  
662 (0.60-0.0055 ppm).

663 Regarding potential *high-tech* by-products in the studied sphalerite grains, In (42-1.6 ppm,  
664 up to 415 ppm) and Ga (31-5.9 ppm, up to 96 ppm) contents are mostly higher than Ge  
665 contents (0.85-0.16 ppm, up to 9.3 ppm). In samples from the María Teresa deposit, the  
666 concentrations of In and Cu yield a positive correlation trend with a slope  $\sim 1$  corroborating  
667 a  $\text{Cu}^+ + \text{In}^{3+} \leftrightarrow 2\text{Zn}^{2+}$  coupled substitution. The dataset also shows a positive correlation



668 trend between Ga and Cu; analyses of sphalerite from the María Teresa, Perubar and Cerro  
669 de Maimón deposits lie close to the Ga:Cu 1:1 ratio line thus pointing to a  $\text{Cu}^+ + \text{Ga}^{3+} \leftrightarrow$   
670  $2\text{Zn}^{2+}$  coupled substitution. Furthermore, the narrow arrangement of most analyses along  
671 the 1:1 ratio line in a In + Ga + Ge + Sn vs. Cu + Ag plot suggests the occurrence of  
672 additional, more complex substitution mechanisms involving different combinations of  
673 monovalent ( $\text{Cu}^+$ ,  $\text{Ag}^+$ ), divalent ( $\text{Ge}^{2+?}$ ), trivalent ( $\text{In}^{3+}$ ,  $\text{Ga}^{3+}$ ,  $[\text{Sn}^{3+?}]$ ), and quadrivalent  
674 ( $\text{Ge}^{4+}$ ,  $\text{Sn}^{4+}$ ) cations.

675 The trace element composition of chalcopyrite was investigated for the María Teresa and  
676 Cerro de Maimón deposits as chalcopyrite is a trace phase in the Palma and Perubar  
677 deposits. Trace elements such as Zn, Ag, Sn, Cd, Se, In, Ga, and Ge are present in solid  
678 solution as opposed to the likely occurrence of Mo, Au, Tl, Sb, Pb, and Bi in  
679 microinclusions. Deleterious elements show low contents (Zn = 426-190 ppm, Se = 64-22  
680 ppm, Sn = 53-1.3 ppm, Cd = 12-6.0 ppm, Tl < 11 ppm, Bi < 10 ppm, Ni < 2.5 ppm, As <  
681 2.0 ppm, Hg < 3.8 ppm) relative to reference maximum contents before charge is incurred  
682 *sensu* Fontain (2013). The contents of Ag (136-23 ppm, up to 296 ppm) and Au (< 0.40  
683 ppm) are also relatively low. In analyzed chalcopyrite grains, the contents of the potential  
684 *high-tech* by-products In (28-8.4 ppm, up to 49 ppm) and Ga (8.3-2.9 ppm, up to 24 ppm)  
685 are lower, and the contents of Ge (5.9-0.70 ppm, up to 80 ppm), higher, than in sphalerite  
686 from the respective deposits. General positive correlation trends between Zn, Cd, In, Ge,  
687 and Ga in chalcopyrite suggest varied coupled substitution mechanisms of Fe (+2 and +4)  
688 and Cu (+1 and +3) with fluctuating valences due to covalent bonding in this mineral.

689 A comprehensive sampling across the Sofia-D massive sulfide body in the María Teresa  
690 deposit revealed significant mineral and trace-element zonation. Mineral zones in Sofia-D  
691 include a core of dominant “barren” pyrite sheathed with facies of chalcopyrite (Cu zone)  
692 that grades outwards to sphalerite ± fahlore ± galena ± chalcopyrite (Zn zone) and galena +  
693 sphalerite ± fahlore ± barite (Pb – Zn – Ag zone) carapaces. Strong replacement of early  
694 sphalerite by chalcopyrite and metalliferous zonation strongly suggest zone refining. The  
695 contents of some trace elements in sphalerite and to a lesser extent in chalcopyrite display  
696 systematic vertical distribution patterns. In sphalerite, these include, bottom to top of the  
697 sulfide body, depletion in In, Cu, Mn, and Se and enrichment in Ge. This distribution

698 agrees with the expected decrease of the crystallization temperature and/or volatile  
699 magmatic influx from the feeders of the massive mineralization. In chalcopyrite, the  
700 distributions of trace elements and Sc/Zn ratios are rather uneven except for constant  
701 enrichment in Bi (probably present as microinclusions) and Se towards the basal portion of  
702 the sulfide body. Unlike sphalerite, the trace element composition of chalcopyrite is  
703 strongly dependent on the co-crystallizing phases over physicochemical conditions. Co-  
704 crystallization and/or reequilibration of chalcopyrite with galena at shallower levels may  
705 therefore account for progressive depletion of Se in chalcopyrite.

706 The Sofia-D massive sulfide mineralization shows strong textural and mineralogical  
707 evidence of recrystallization during thermal metamorphism caused by nearby granodiorite  
708 intrusions. The impact of the thermal metamorphism in the absolute trace elements  
709 contents in sphalerite and chalcopyrite is unknown. However, the preservation of expected  
710 original trace element distribution patterns would suggest a closed metamorphic system  
711 and that element interdiffusion was local, probably attaining only contiguous sulfide  
712 grains. Accordingly, we propose that the distribution patterns described in this article for  
713 lattice-bound trace elements in sphalerite and chalcopyrite may be useful during  
714 exploration in order to trace the polarity of massive sulfide bodies in metamorphosed and  
715 tectonized terranes.

## 716 **Acknowledgments**

717 This study was economically supported by the Peruvian CONCYTEC-FONDECYT-World  
718 Bank project 107-2018-FONDECYT-BM-IADT-AV, the Spanish projects CGL2015–  
719 65824 and PID2019-105625RB-C21, and a Huiracocha Ph. D granted by the PUCP to  
720 D.B. Minera Colquisiri provided field and logistical support, access to the mine area, and  
721 the documents to prepare Fig. 2; its staff is most gratefully acknowledged, particularly  
722 Juan José Herrera, Edgar Pichardo, and Óscar Chirinos. We also thank Perilya-  
723 CORMIDOM for providing specimens from the Cerro de Maimón deposits. The analyzed  
724 material from Perubar and Palma was sampled (1998-2000) by Marc Polliand in the frame  
725 of his PhD thesis at the University of Geneva. The senior author acknowledges C.E.  
726 Nelson for insightful debate on the origin and classification of VMS deposits hosted in  
727 Cretaceous arc-related volcanic units in Cuba and the Dominican Republic. We appreciate

728 the technical support by Xavier Llovet (CCiT-UB) during the acquisition of EPMA data,  
729 and by Peter Tollan (ETHZ) during the acquisition of LA-ICP-MS data. We are grateful to  
730 two anonymous reviewers for their constructive comments.

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1063 **Figure captions**

1064 **Fig. 1.** Geologic map of the Peruvian Coastal Batholith and Mesozoic Western Peruvian  
1065 Trough (WPT) with related volcano-sedimentary basins hosting massive sulfide deposits.  
1066 Modified from [Polliand et al. \(2005\)](#). Distribution of the Cretaceous Peruvian Coastal  
1067 Batholith and strata after the digital version of the geological map of Peru ([INGEMMET,  
1068 2000](#)).

1069 **Fig. 2.** Representative schematic cross-section across the Sofía D massive sulfide body in  
1070 the María Teresa VMS deposit, Peru (redrawn from geological documents of the María  
1071 Teresa Mine geological department as of December 2020).

1072 **Fig. 3.** (A) Geographic location of the Early Cretaceous boninite and island arc tholeiite  
1073 volcanic series and ophiolitic peridotites along the Greater Antilles. (B) Geological map of  
1074 the Maimón Formation and surrounding Early Cretaceous arc-related volcanic units  
1075 (modified from [Martín and Draper, 1999](#)).

1076 **Fig. 4.** Drill core photographs of sulfide mineralization from the Sofía-D massive sulfide  
1077 body in the María Teresa VMS deposit. (A-C) Semi-massive sulfide ore dominated by  
1078 medium-grained pale-brown sphalerite grains partially replaced by an assemblage of pyrite  
1079 and chalcopyrite. (D) Massive sulfide ore composed of coarse-grained red-brownish  
1080 sphalerite grains cut by irregular veinlets of chalcopyrite and pyrite. (E) Dark red, medium-  
1081 sized sphalerite partially replaced by an assemblage of pyrite and chalcopyrite and the  
1082 whole assemblage cut by thin, irregular chalcopyrite veinlets. (F) Coarse-grained pyrrhotite  
1083 and chalcopyrite in a sample picked up close to a contact with a porphyritic andesitic dyke.  
1084 (G) Silicified volcanoclastic rock with chalcopyrite mineralization along clast contacts. (F)  
1085 Irregular veinlets of chalcopyrite in a silicified intrusive rock. Key: cpy = chalcopyrite; po  
1086 = pyrrhotite; py = pyrite; qz = quartz; sl = sphalerite.

1087 **Fig. 5.** Photomicrographs (reflected light) of textural features in the samples from the  
1088 Sofía-D massive sulfide bodies in the María Teresa deposit used for this study. (A) Foam  
1089 textured pyrite with triple junctions at 120° along with interstitial chalcopyrite and  
1090 sphalerite. (B) Anhedral roundish and sub-roundish pyrite grains with irregular fractures  
1091 are surrounded by an assemblage of sphalerite, galena and tennantite. (C-D) Anhedral, sub-  
1092 roundish pyrite and subhedral arsenopyrite grains both with internal fractures with  
1093 interstitial chalcopyrite, sphalerite, and tennantite. (E) Detail of a sphalerite-galena-  
1094 tennantite intergrowth between roundish pyrite grains; note the general foam texture  
1095 described by the contacts between all phases. (F) Pervasive replacement of sphalerite by  
1096 chalcopyrite along borders and as fines disseminations (chalcopyrite disease texture). (G)  
1097 Detail of chalcopyrite with exsolutions of sharply bounded cubanite laths. (H) Detail of  
1098 minute inclusions of auriferous silver in chalcopyrite veinlets crosscutting arsenopyrite.



1099 Key: apy = arsenopyrite; Ag\_Au = auriferous silver; cbn = cubanite; cpy = chalcopyrite;  
1100 mag = magnetite; po = pyrrhotite; py = pyrite; qz = quartz; sl = sphalerite; tn = tennantite.

1101 **Fig. 6.** Paragenetic sequence for the Sofia-D sulfide body in the Calera area, María Teresa  
1102 VMS deposit. Thick bars indicate higher abundances and dashed lines, lesser abundances.

1103 **Fig. 7.** Boxplots of log-transformed element contents (A-I) and crystallization temperature  
1104 (J) according to the GGIMFis geothermometer of [Frenzel et al. \(2016\)](#) for sphalerite at  
1105 different depths along the drill core 34-18-S in the mine section 300 across the Sofia-D  
1106 massive sulfide body, María Teresa deposit (LA-ICP-MS data). Box-and-whisker plots  
1107 provide IQR (box), 1.5 IQR (whiskers), median (line), and outliers (circles).

1108 **Fig. 8.** Boxplots of log-transformed element contents in chalcopyrite at different depths  
1109 along the drill core 34-18-S in the mine section 300 across the Sofia-D massive sulfide  
1110 body, Calera sector, María Teresa deposit (LA-ICP-MS data). Box-and-whisker plots  
1111 provide IQR (box), 1.5 IQR (whiskers), median (line), and outliers (circles).

1112 **Fig. 9.** Binary plots showing inter-element correlations in sphalerite: (A) Fe vs. Zn, (B) Ag  
1113 vs. Pb, (C) Ag vs. Sb, and (D) Sb+Sn vs. Ag+Pb. Dashed lines represent 1:1 molar  
1114 correlation. Additionally, contents are presented in box-and-whisker plots, which provide  
1115 IQR (box), 1.5 IQR (whiskers), median (line) and outliers (circles). The composition of  
1116 sphalerite from other VMS deposits worldwide are shown for comparison (data from  
1117 [Carvalho et al., 2018](#), [Cook et al., 2019](#), [Revan et al., 2014](#), and [Ye et al., 2011](#)).

1118 **Fig. 10.** Binary plots showing inter-element correlations in sphalerite: (A) In vs. Cu, (B)  
1119 Ga vs. Cu, and (C) Sn+In+Ge+Ga vs. Cu+Ag. Dashed lines represent 1:1 molar  
1120 correlation. Additionally, contents are presented in box-and-whisker plots, which provide  
1121 IQR (box), 1.5 IQR (whiskers), median (line) and outliers (circles). The composition of  
1122 sphalerite from other VMS deposits worldwide are shown for comparison (data from  
1123 [Carvalho et al., 2018](#), [Cook et al., 2019](#), [Revan et al., 2014](#), and [Ye et al., 2011](#)).

1124 **Fig. 11.** Binary plots showing inter-element correlations in chalcopyrite: (A) Bi vs. Cu, (B)  
1125 In vs. Cu, (C) Ga vs. Cu, and (D) Sn+In+Ge+Ga vs. Cu+Ag. Dashed lines represent  
1126 1:1molar correlation. Additionally, contents are presented in box-and-whisker plots, which

1127 provide IQR (box), 1.5 IQR (whiskers), median (line) and outliers (circles). The  
1128 composition of sphalerite from other VMS deposits worldwide are shown for comparison  
1129 (data from [Carvalho et al., 2018](#), [George et al., 2018](#), and [Revan et al., 2014](#)).

### 1130 **Table captions**

1131 **Table 1.** Summary of element contents in sphalerite from the María Teresa, Perubar,  
1132 Palma, and Cerro de Maimón VMS deposits (LA-ICP-MS data).

1133 **Table 2.** Summary of element contents in chalcopyrite from the María Teresa and Cerro de  
1134 Maimón VMS deposits (LA-ICP-MS data).

1135 **Table 3.** Summary of LA-ICP-MS data for sphalerite from subduction-related VMS  
1136 deposits reported in the literature.

1137 **Table 4.** Summary of LA-ICP-MS data for chalcopyrite from subduction-related VMS  
1138 deposits reported in the literature.

### 1139 **Supplementary material**

1140 **Table S1.** List of samples and their location.

1141 **Table S2.** Representative EPMA analyses of sphalerite and chalcopyrite from the María  
1142 Teresa, Palma and Perubar VMS deposits in Peru. Structural formula of sphalerite was  
1143 normalized to S = 1. Structural formula of chalcopyrite was normalized to S = 2.

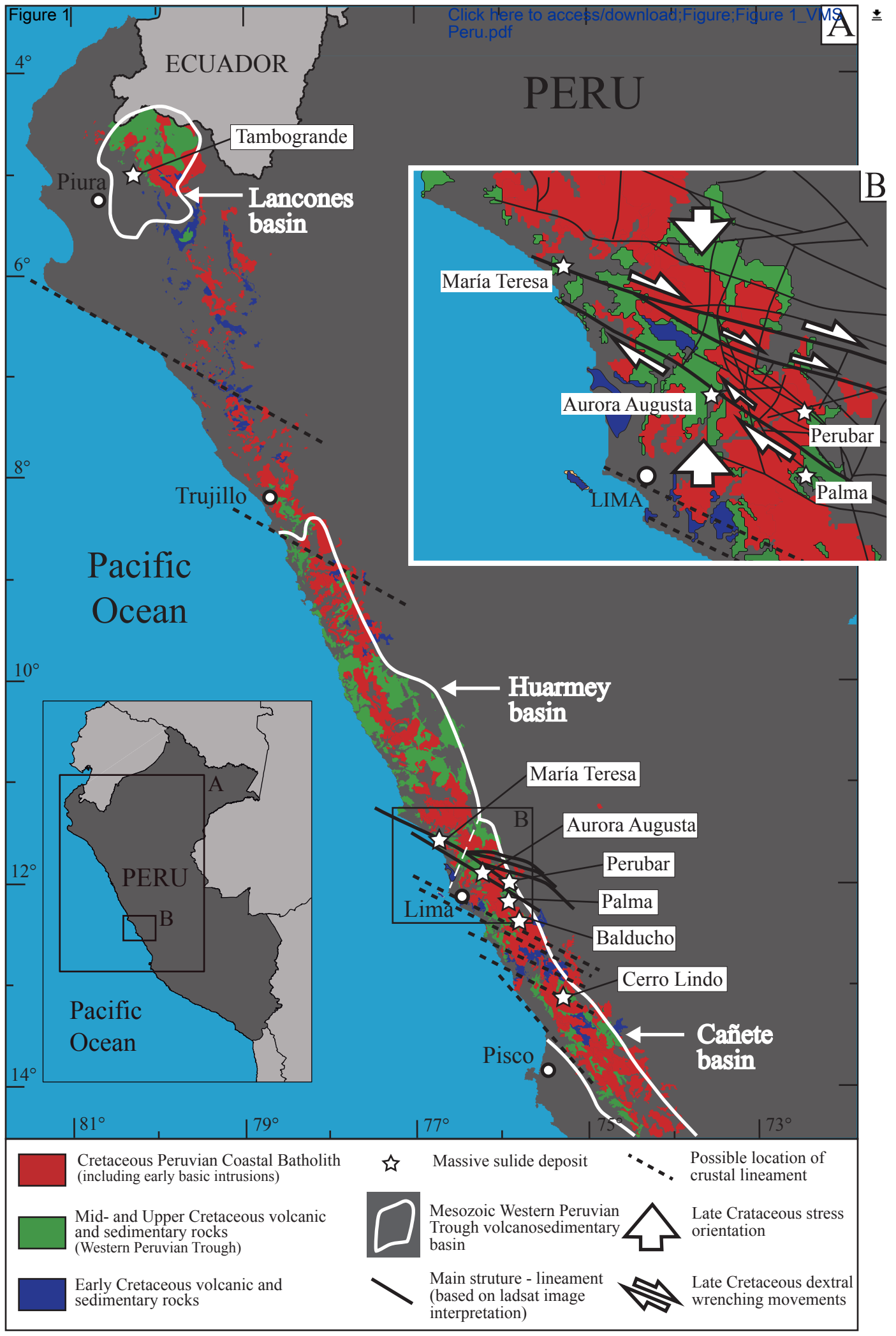
1144 **Table S3.** LA-ICP-MS trace element analyses on sulfides and reference materials.

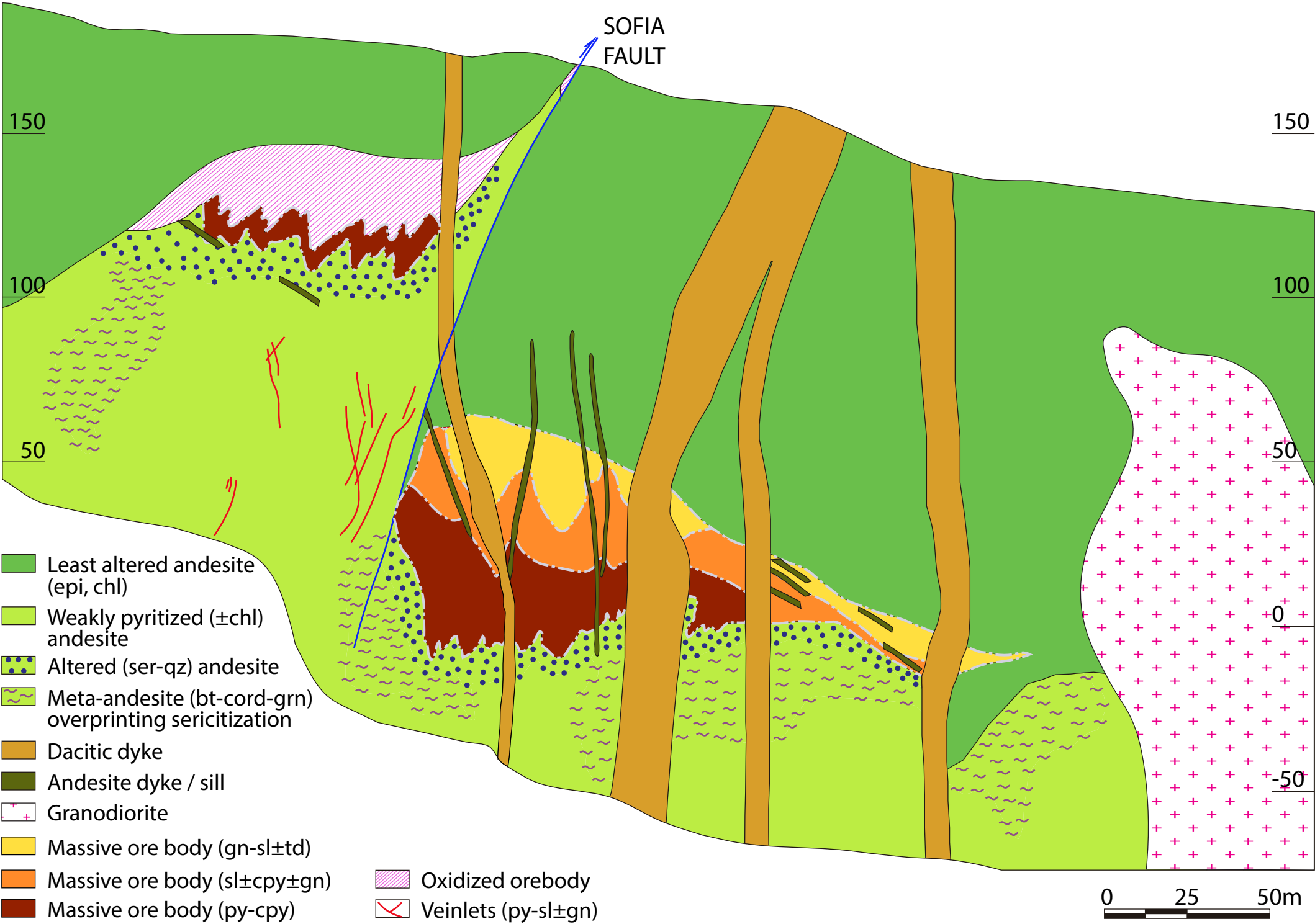
1145 **Fig. S1.** Photomicrographs (reflected light) of textural features in samples used for this study  
1146 from the Perubar deposit. **A)** Pyrrhotite intergrown with massive sphalerite. A minute (4  $\mu\text{m}$ )  
1147 grain of electrum occurs in the contact between both minerals. **B)** Sphalerite-galena  
1148 assemblage with local chalcopyrite inclusions in a matrix of baryte. **C)** Pyrrhotite, pyrite,  
1149 sphalerite, and chalcopyrite. A pyrite crystal shows microporosity in its core and massive  
1150 overgrowth bands. **D)** Subhedral magnetite crystal surrounded by massive sphalerite, the  
1151 latter also including pyrite and pyrrhotite inclusions. **E)** Elongated molybdenite crystal in

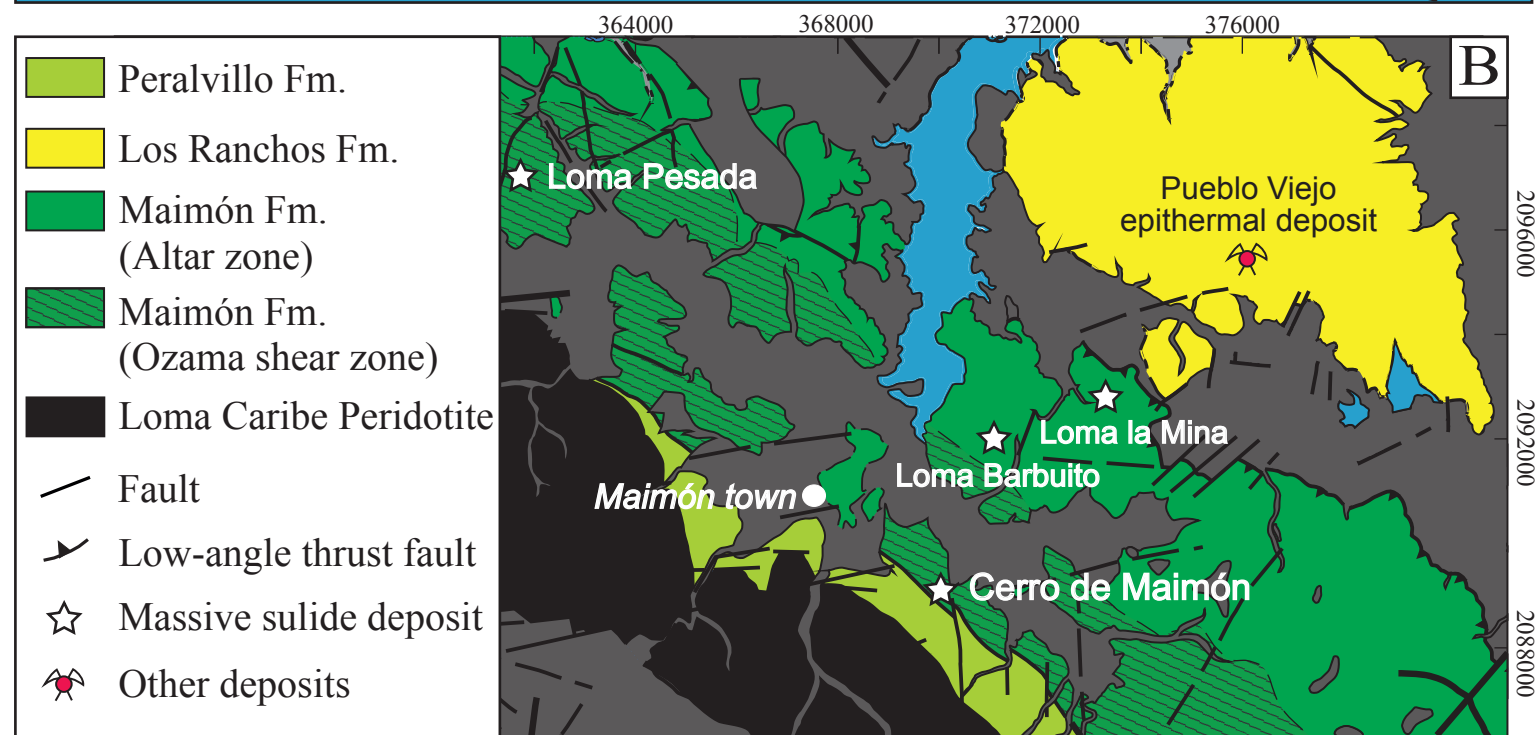
1152 contact with sphalerite. **F)** Electrum grains (4-8  $\mu\text{m}$ ) along pyrite grain boundaries.  
1153 Pyrrhotite partly replaced by pyrite and marcasite. Abbreviations: **bar**: baryte; **cpy**:  
1154 chalcopyrite; **el**: electrum; **gn**: galena; **mc**: marcasite; **mt**: magnetite; **mb**: molybdenite; **py**:  
1155 pyrite; **po**: pyrrhotite; **sl**: sphalerite.

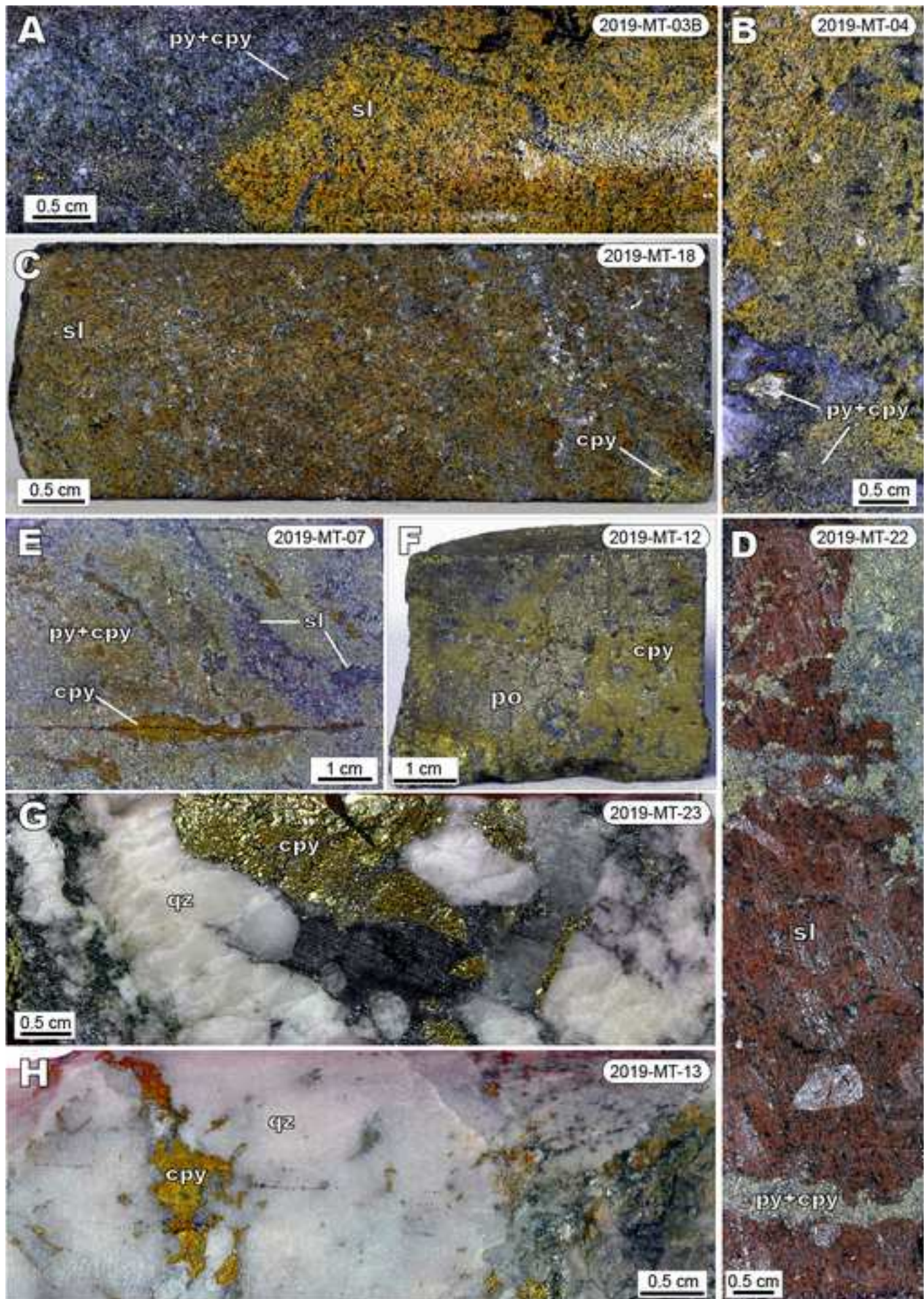
1156 **Fig. S2.** Photomicrographs (reflected light) of textural features in samples used for this study  
1157 from the Palma deposit. **A)** Aggregates of spheroidal and subhedral pyrite grains cemented  
1158 by sphalerite and galena. **B)** Mass of sphalerite and minor galena along with relictic  
1159 pyrrhotite. **C)** Sphalerite replacement of pyrrhotite; sphalerite also lines voids in cariated  
1160 pyrite. **D)** Pyrite partly showing "atoll-like texture" in massive sphalerite. **E)** Pyrrhotite  
1161 aggregates replaced by massive sphalerite and galena. **F)** Subhedral pyrite in massive  
1162 sphalerite with pyrrhotite relicts. Fine disseminations of chalcopyrite in sphalerite.  
1163 Abbreviations: **cpy**: chalcopyrite; **gn**: galena; **py**: pyrite; **po**: pyrrhotite; **sl**: sphalerite.

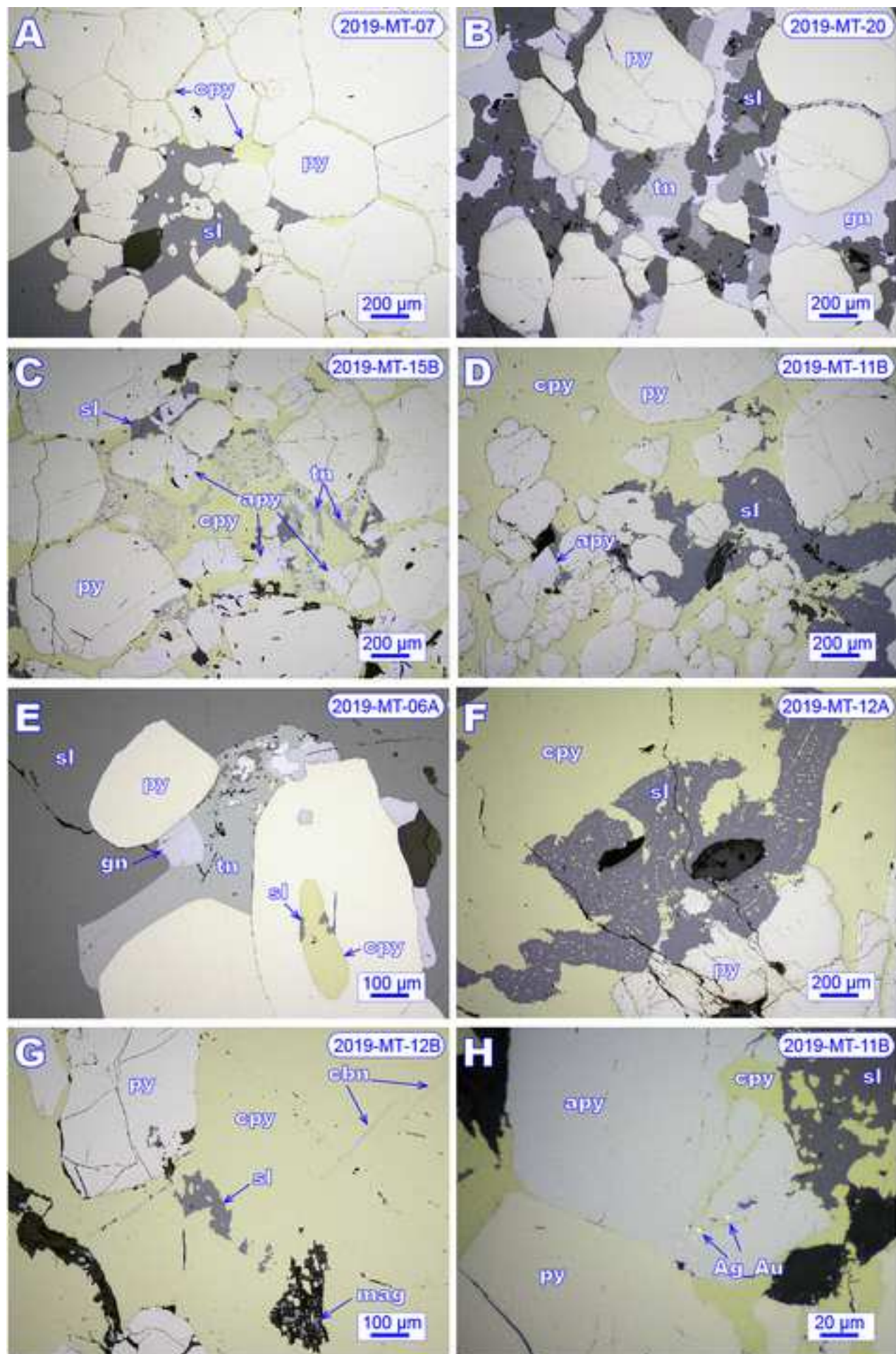
1164 **Fig. S3.** Representative downhole LA-ICP-MS spectra for selected elements in sphalerite  
1165 (A-D) and chalcopyrite (E-F).













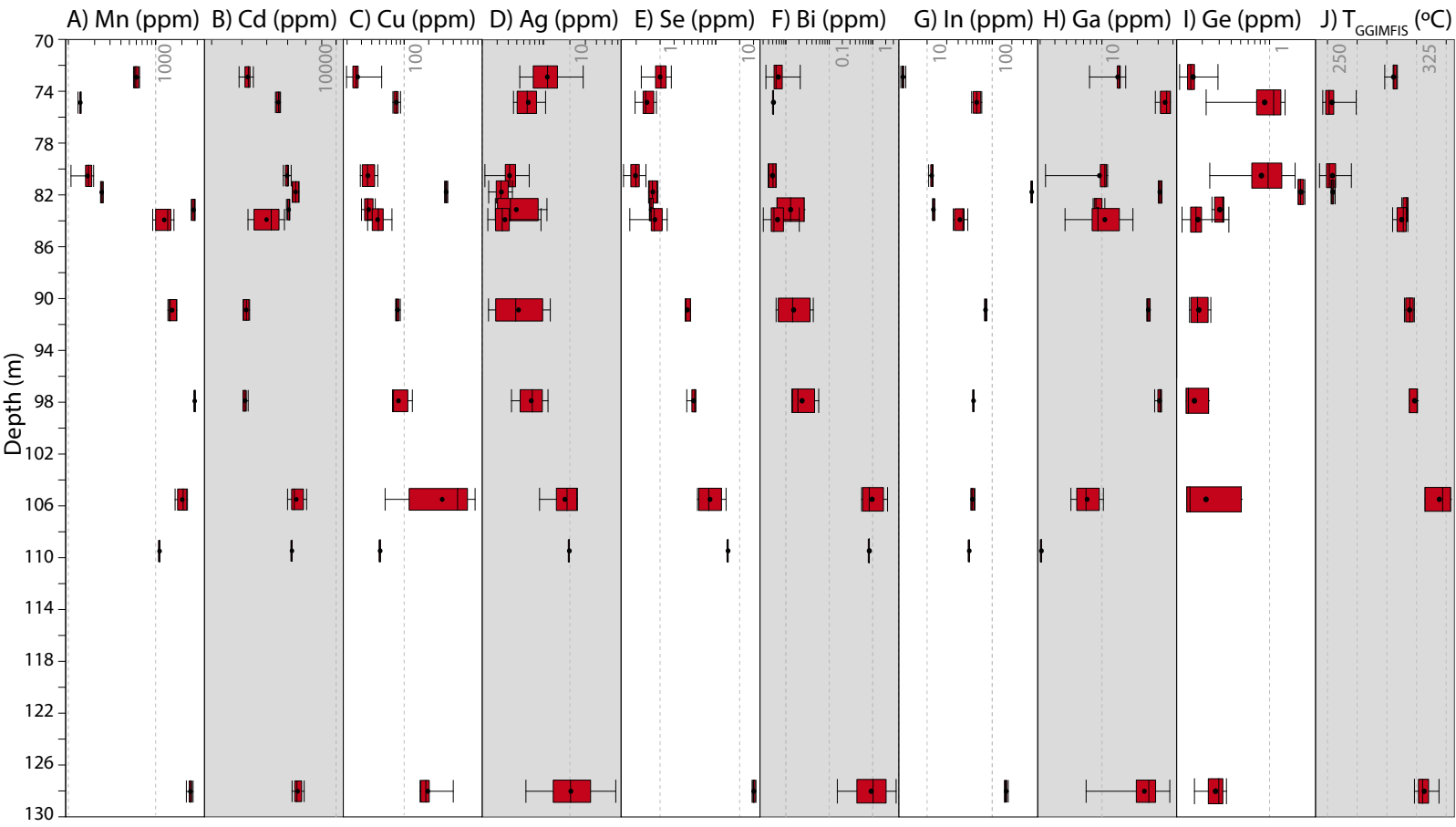
# Sofía-D, Calera sector, María Teresa VMS deposit

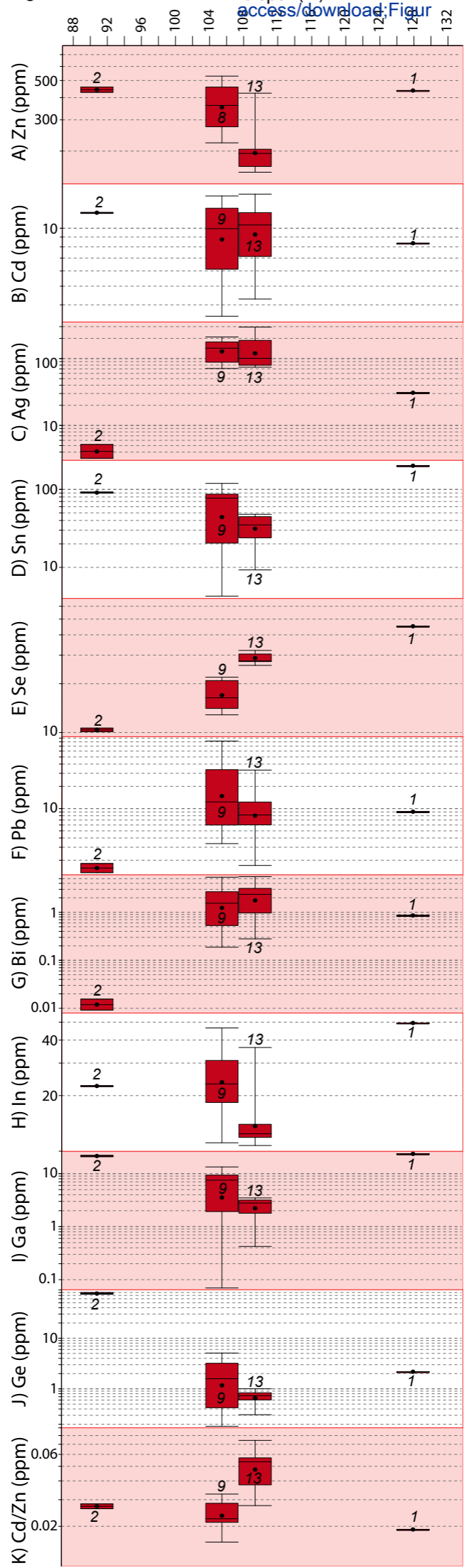
[Click here to access/download/Figure/Figure 6\\_Paragenetic seq MT.pdf](#)

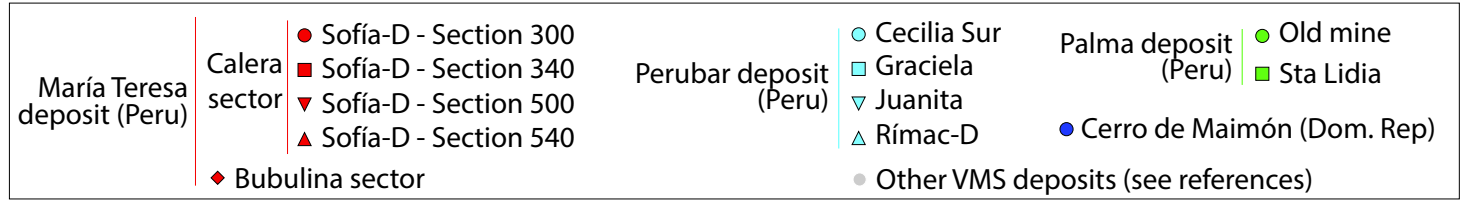
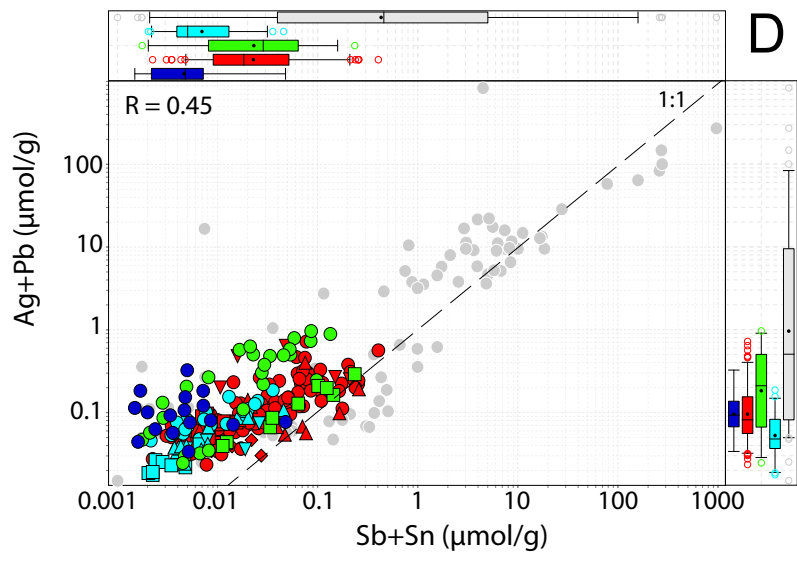
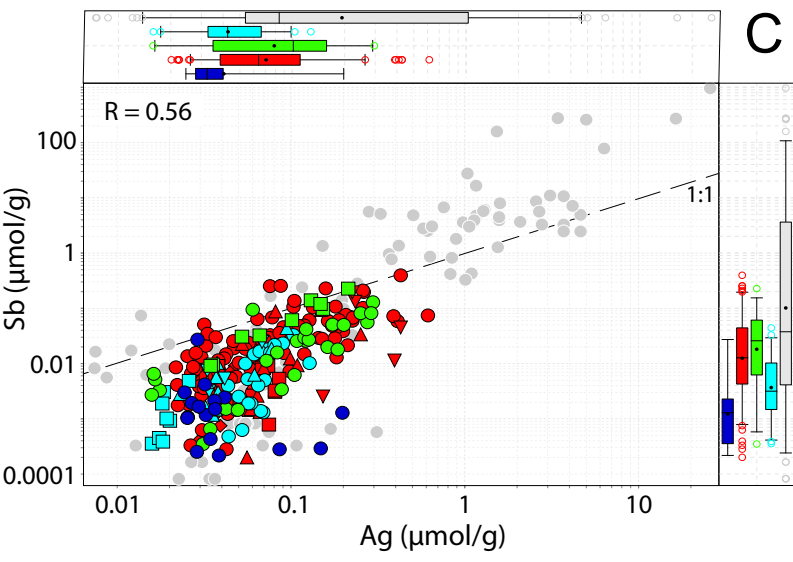
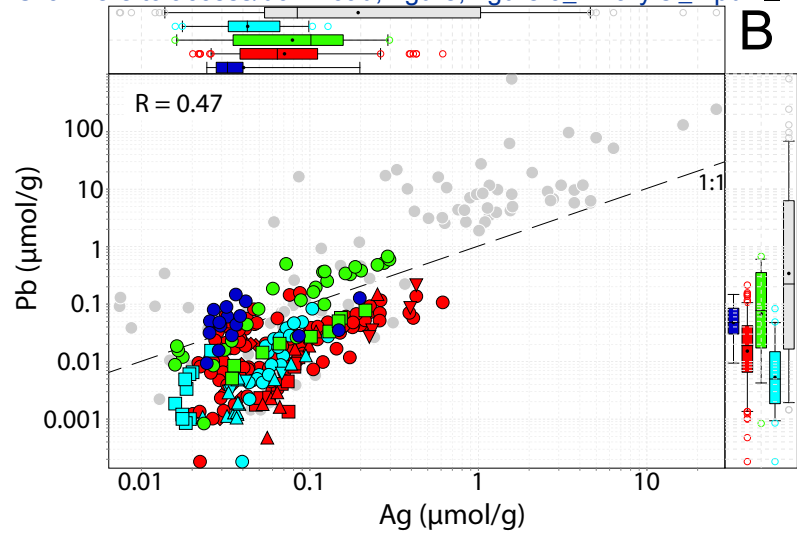
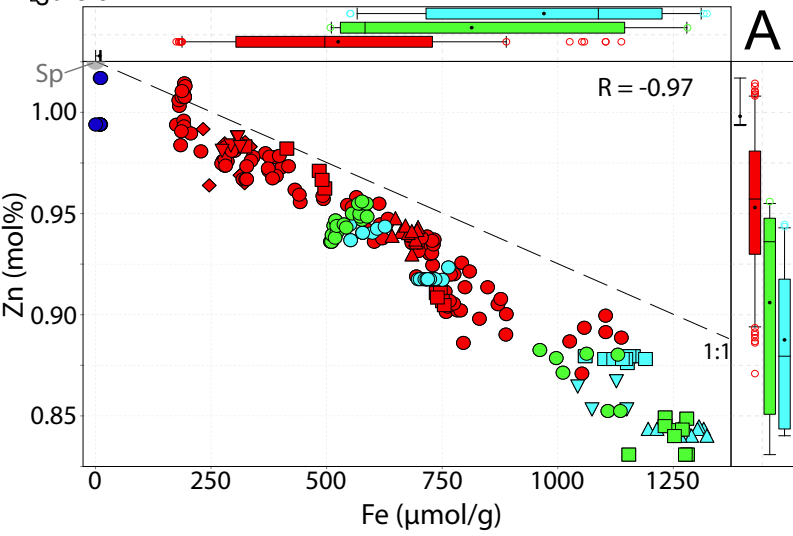
Mineral Formula	VMS mineralization	Contact metamorphism
Arsenopyrite $FeAsS$	—	
Pyrite $FeS_2$	█	
Galena $PbS$		
Tennantite-(Fe) $Cu_6(Cu_4Fe_2)As_4S_{13}$	█	
Bournotite $PbCuSbS_3$	█	
Boulangerite $Pb_5Sb_4S_{11}$	---	
Polybasite $[(Ag,Cu)_6(Sb,As)_2S_7][Ag_9CuS_4]$	---	
Sphalerite $ZnS$	█	
Chalcopyrite $CuFeS_2$	█	
Cubanite $CuFe_2S_3$	█	? - - ?
Auriferous Ag $(Ag,Au)$	---	
Pyrrhotite $Fe_{1-x}S$		█
Magnetite $Fe^{2+}Fe^{3+}_2O_4$		---

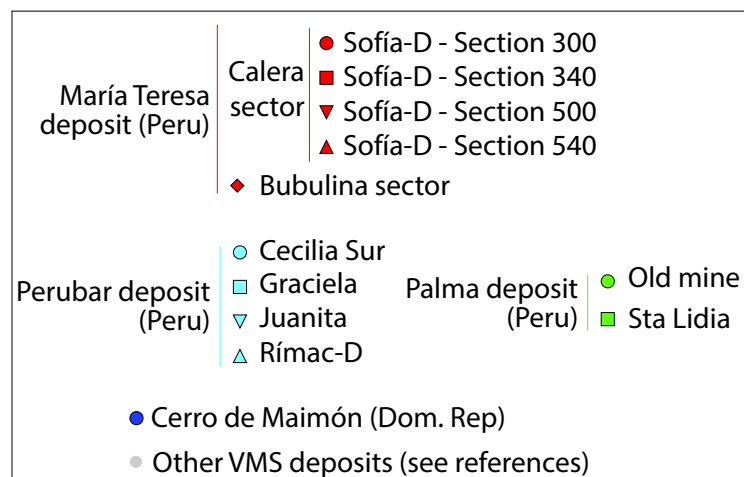
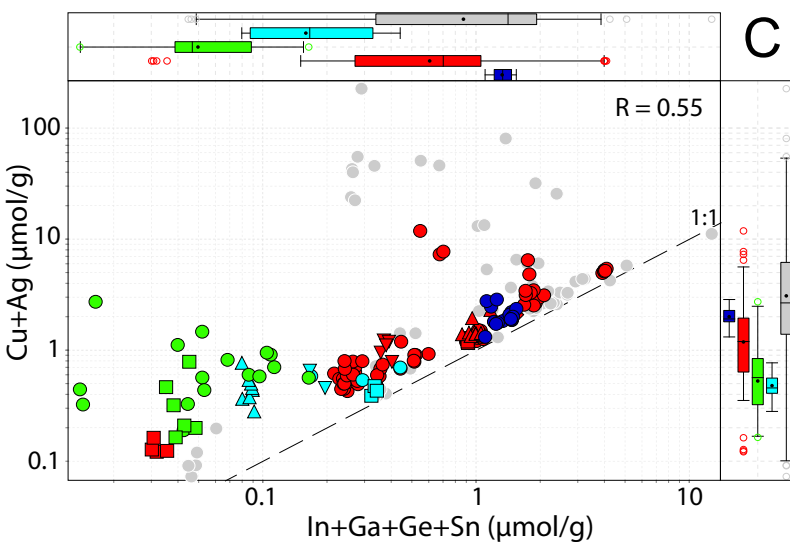
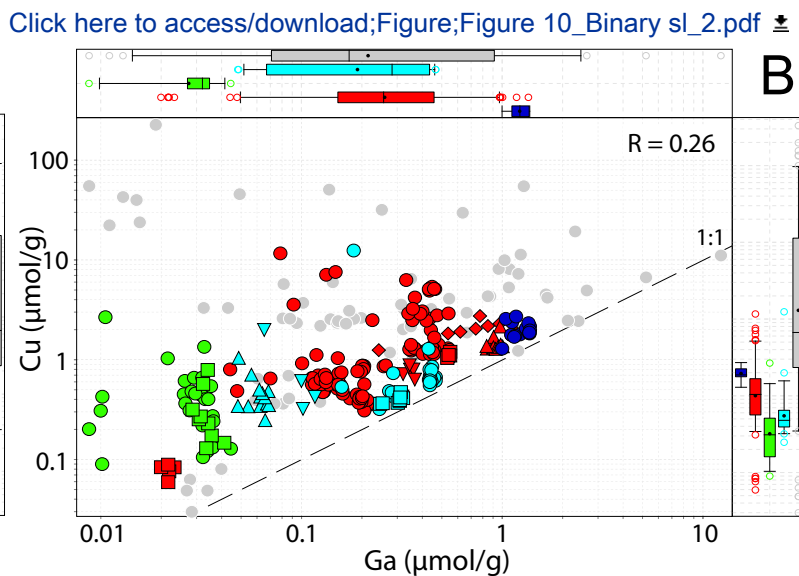
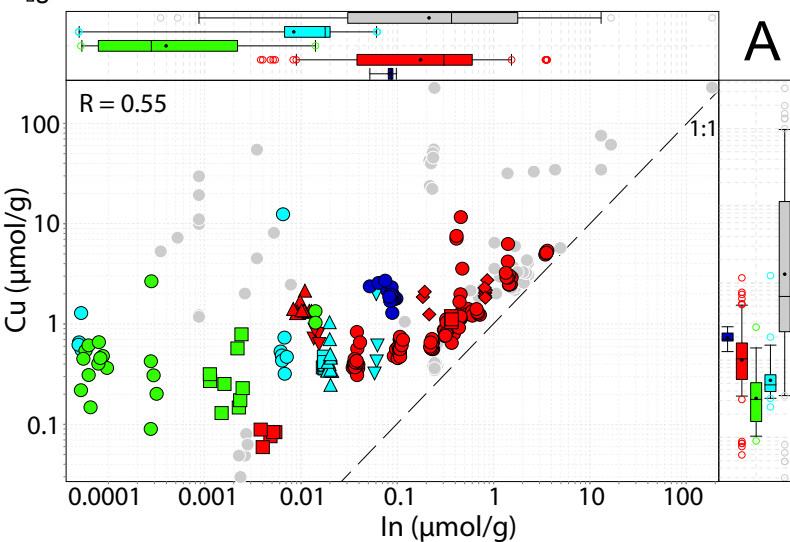
Zone refining

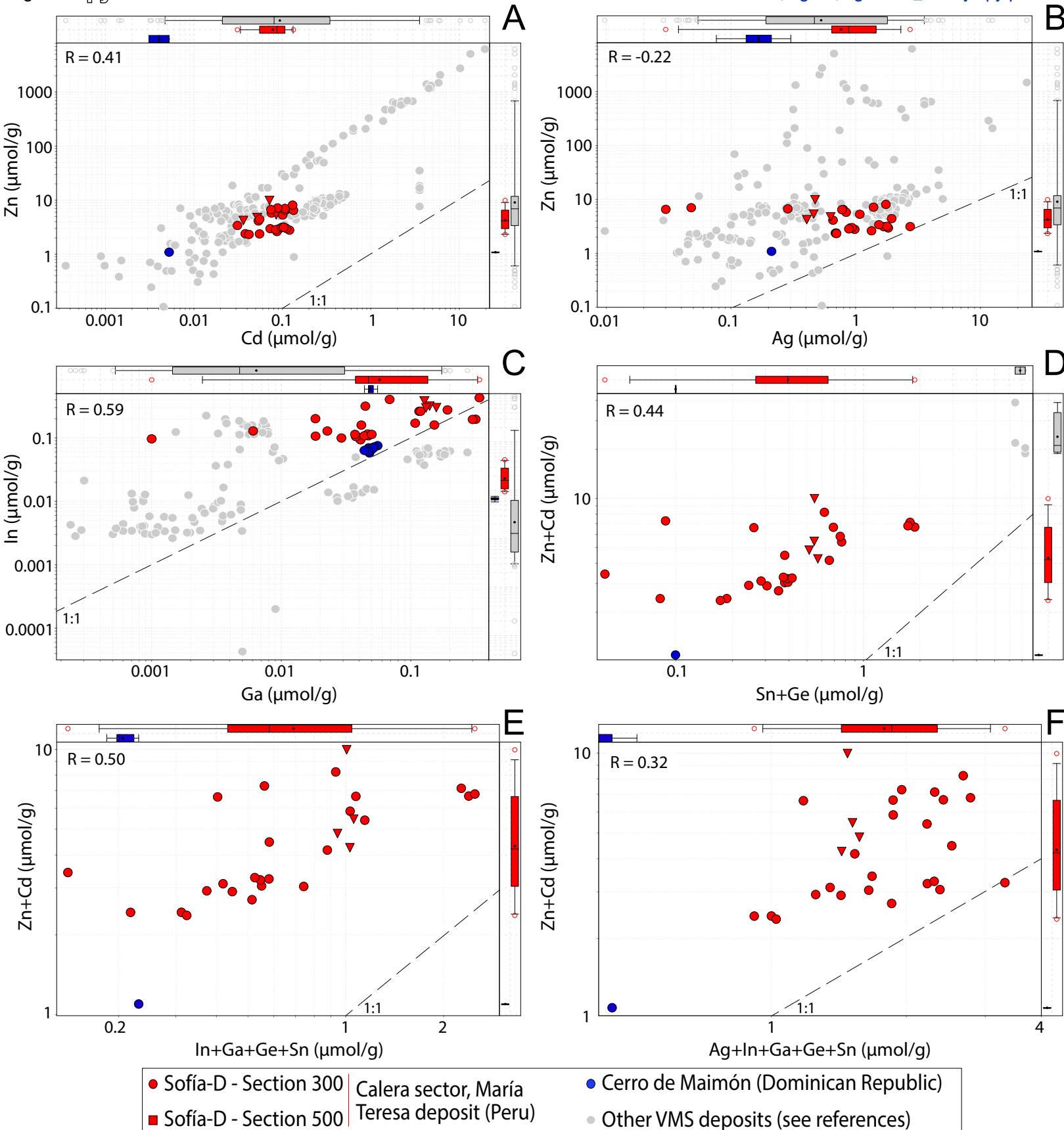












Deposit	Sector / Body / Section		Mn (ppm)	Fe (ppm)	Cu (ppm)
María Teresa (Peru)	All samples (n = 141)	Min.	107	9772	3.8
		Max.	4998	63534	740
		GM	1353	29317	96
		SD	1236	13930	106
		IQR	2343-326	40675-17122	98-36
	Bubulina / Bubulina 1180 (n = 9)	Min.	3139	13000	79
		Max.	4998	18702	174
		GM	4474	16628	126
		SD	550	2135	26
	Calera / Sofía-D, all samples (n = 132)	Min.	107	9772	3.8
		Max.	2857	63534	740
		GM	716	26558	61
		SD	939	13984	109
		IQR	1771-303	40842-17173	87-36
	Calera / Sofía-D / 300 (n = 105)	Min.	107	9772	20
		Max.	2857	63534	740
		GM	828	25631	66
SD		967	14897	120	
Calera / Sofía D / 340 (n = 10)	Min.	113	18062	3.8	
	Max.	1662	42132	76	
	GM	468	31839	19	
	SD	799	9365	35	
	IQR	1646-139	41573-27114	70-5.3	
Calera / Sofía D / 500 (n = 5)	Min.	320	15296	40	
	Max.	584	38233	82	
	GM	502	19140	53	
	SD	97	8832	15	
Calera / Sofía D / 540 (n = 12)	Min.	214	35835	82	
	Max.	357	38973	137	
	GM	322	37824	91	
	SD	36	1036	16	
	IQR	339-326	38681-37310	89-84	
All samples (n = 50)	min	667	30747	16	
	max	7639	73876	789	
	GM	2421	51935	35	
	SD	2738	15233	108	
	IQR	5964-907	67570-39965	41-25	
Cecilia Sur	Min.	667	30747	20	

Perubar (Peru)	(n = 22)	Max.	1047	42636	789
		GM	906	37667	45
		SD	75	3821	160
		IQR	939-879	40139-34273	46-36
	Graciela (n = 10)	Min.	6704	59164	23
		Max.	7639	66451	30
		GM	7267	63460	26
		SD	240	2010	2
		IQR	7367-7197	64413-62864	27-24
	Juanita (n = 4)	Min.	1669	58263	20
		Max.	1973	64193	126
		GM	1801	61300	41
		SD	128	2706	49
		IQR	1856-1736	63248-59558	61-25
	Rímac-D (n = 14)	Min.	5121	66771	16
Max.		6090	73876	66	
GM		5631	71109	27	
SD		325	1984	13	
IQR		5915-5408	72557-70501	30-22	
Palma (Peru)	All samples (n = 37)	Min.	2303	28384	5.7
		Max.	10848	71526	170
		GM	6676	42211	20
		SD	2413	17860	29
		IQR	7751-6734	63423-30015	31-11
	Old mine (n = 28)	Min.	2303	28384	5.7
		Max.	7751	63423	170
		GM	5794	35929	21
		SD	1822	12819	33
	Sta Lidia (n = 9)	Min.	9837	64436	8.2
		Max.	10848	71526	50
		GM	10375	69687	17
		SD	332	2237	14
		IQR	10604-10119	71269-68821	44520.0
	Cerro de Maimón (Dominican Republic)	Cerro de Maimón (n = 17)	Min.	406	B.D.L.
Max.			539	643	172
GM			467	371	124
SD			40	193	22
IQR			482-446	585-535	138-113

B.D.L.: Below detection limit; IQR: interquartile range; GM: geometric mean; SD: standar

\*Ragged ablation spectra in some analyses indicate that these elements may occur as su

\*\* According to Zn concentrate analyses, Cd values (LA-ICP-MS data) may be overevalua



Zn (ppm)	Ga (ppm)	Ge (ppm)	As (ppm)	Se (ppm)	Mo (ppm)	Ag (ppm)
569500	1.4	B.D.L.	4.6	B.D.L.	B.D.L.	2.2
663300	94	2.3	9.3	32	1.6	66
623168	25	0.51	6.0	6.4	0.16	11
22726	19	0.56	0.8	9.3	0.19	10
641300-608100	32-11	0.65-0.17	6.5-5.5	5.3-1.6	0.20-0.047	12-4.2
630300	17	B.D.L.	4.7	4.3	0.25	3.2
648500	94	0.16	5.8	5.3	0.73	5.2
640091	51	0.16	5.3	4.8	0.37	4.2
6640	23	-	0.3	0.4	0.15	0.7
643900-633600	68-43	0.16-0.16	5.4-5.1	5.0-4.6	0.36-0.31	4.6-3.8
569500	1.4	B.D.L.	4.6	B.D.L.	B.D.L.	2.2
663300	70	2.3	9.3	32	1.6	66
621589	17	0.33	6.0	3.1	0.09	7.9
22982	17	0.56	0.8	9.6	0.18	10.6
639625-605100	31-11	0.66-0.17	6.5-5.5	5.5-1.5	0.17-0.045	13-4.2
569500	3.1	B.D.L.	5.2	B.D.L.	B.D.L.	2.2
663300	38	2.3	9.3	32	1.6	66
621999	15	0.37	6.1	3.3	0.11	7.5
24203	10	0.60	0.7	10.6	0.19	10.5
639600-602500	27-10	0.79-0.17	6.4-5.7	5.6-1.5	0.20-0.050	13-3.9
591600	1.4	B.D.L.	6.7	B.D.L.	B.D.L.	4.1
642900	38	0.21	7.1	7.1	0.13	9.2
614743	8	0.16	6.9	3.1	0.06	6.9
22717	19	0.04	0.1	2.8	0.05	1.8
634275-594450	38-1.5	0.19-0.13	7.0-6.9	6.6-1.3	0.11-0.027	8.6-5.7
608000	22	0.12	4.9	1.3	0.033	11
645800	59	0.95	5.6	3.7	0.61	46
637688	28	0.37	5.3	2.1	0.07	24
14717	14	0.31	0.2	0.9	0.23	14
645150-641800	25-24	0.64-0.29	5.4-5.3	2.4-1.7	0.065-0.038	38-18
608000	58	B.D.L.	4.6	2.1	0.037	5.7
619700	70	1.2	5.4	3.7	0.67	27
614685	64	0.22	5.0	2.8	0.06	8.9
2950	4	0.33	0.3	0.4	0.18	6.2
616125-613700	68-63	0.23-0.16	5.2-4.9	3.0-2.6	0.055-0.042	11-6.5
549400	3.4	B.D.L.	4.0	B.D.L.	0.046	1.7
617700	33	0.49	6.1	10	0.55	14
579909.2969	13	0.17	4.6	3.5	0.19	4.6
24068.28288	11	0.08	0.4	3.0	0.17	2.7
600000-551800	30-4.7	0.19-0.15	4.8-4.4	8.2-2.5	0.42-0.089	7.1-3.6
599800	11	B.D.L.	4.3	B.D.L.	0.046	4.3

617700	33	0.19	5.4	4.4	0.20	14
605057	25	0.16	4.7	1.7	0.085	6.8
7381	7	0.03	0.3	1.3	0.031	2.2
614400-600000	31-20	0.17-0.14	4.8-4.6	3.2-1.1	0.098-0.071	7.8-5.7
572900	17	B.D.L.	4.0	3.2	0.39	1.7
575100	22	0.20	4.7	4.2	0.55	2.8
574430	21	0.18	4.3	3.7	0.48	2.0
699	2	0.01	0.2	0.3	0.04	0.3
575100-574200	22-21	0.19-0.18	4.3-4.2	4.0-3.5	0.49-0.46	2.1-1.9
557900	4.5	B.D.L.	4.7	2.5	0.12	3.6
567000	8.1	0.49	6.1	3.9	0.17	7.4
562010	6.5	0.25	5.2	3.1	0.14	4.6
4813	1.5	0.26	0.6	0.6	0.03	1.8
565725-557900	7.4-6.4	0.40-0.22	5.4-5.0	3.3-2.9	0.16-0.12	5.2-3.6
549400	3.4	B.D.L.	4.4	6.5	0.33	2.5
552400	4.8	0.22	4.9	10	0.45	10
551092	4.2	0.17	4.6	8.6	0.38	4.4
1010	0.5	0.03	0.2	1.1	0.04	2.1
551600-550625	4.6-4.0	0.20-0.15	4.8-4.6	9.3-8.3	0.42-0.35	4.1-3.9
543300	0.61	B.D.L.	3.6	B.D.L.	0.13	1.7
625100	3.1	9.3	6.3	1.6	0.69	32
591619	1.9	0.6	4.9	0.94	0.44	8.5
31430	0.6	2.6	0.7	0.25	0.15	8.8
619200-557400	2.4-1.9	3.5-0.16	5.3-4.3	1.1-0.8	0.58-0.41	17-3.8
557400	0.61	B.D.L.	3.6	B.D.L.	0.13	1.7
625100	3.1	9.3	6.3	1.6	0.58	32
605878	1.8	0.9	4.8	1.0	0.39	8
22214	0.7	2.7	0.84	0.29	0.12	10
620375-603350	2.4-1.8	4.6-0.18	5.6-4.0	1.2-0.8	0.47-0.39	18-3.6
543300	2.0	B.D.L.	4.7	B.D.L.	0.64	3.7
555300	2.9	0.23	5.3	1.1	0.69	23
549369	2.3	0.16	5.1	0.86	0.66	9.3
4919	0.2	0.04	0.2	0.12	0.02	6.7
552500-543300	2.4-2.2	0.19-0.14	5.3-4.9	0.9-0.8	0.67-0.64	16-5.6
650000	70	0.23	3.6	17	B.D.L.	2.6
665000	96	6.2	10	67	1.7	21
652622	85.2	0.9	4.4	49	0.20	4.4
5894	9.0	1.4	1.6	17	0.52	5.3
650000-650000	95-78	1.2-0.68	4.4-3.9	61-57	0.30-0.090	4.2-3.1

rd deviation

ib-microinclusions

ted by 48-57%

Cd (ppm)	In (ppm)	Sn (ppm)*	Sb (ppm)	Hg (ppm)	Tl (ppm)*	Pb (ppm)*
4720	0.44	0.14	B.D.L.	98	B.D.L.	0.038
9409	415	20	48	788	19	44
6605**	60	1.2	4.3	392	1.0	6.2
1279	89	2.8	7.2	135	2.1	7.2
7373-5290	68-4.4	0.61-0.25	5.2-0.52	448-326	1.2-0.060	8.7-1.4
4720	21	0.30	0.048	664	B.D.L.	0.21
5044	99	3.2	1.1	788	0.0081	4.3
4867	59	1.0	0.28	724	0.0067	1.2
118	36	1.0	0.37	35	0.0010	1.7
4964-4804	94-25	1.7-0.35	0.54-0.15	737-704	0.0076-0.0061	4.1-0.62
4896	0.44	0.14	B.D.L.	98	B.D.L.	0.038
9409	415	20	48	677	19	44
6613	19	0.48	1.7	349	0.28	3.4
1235	91	2.86	7.3	107	2.18	7.3
7411-5446	54-4.3	0.58-0.24	5.5-0.65	438-318	1.3-0.084	8.8-1.6
4896	3.9	0.14	B.D.L.	98	0.0072	0.038
8051	415	16	48	485	19	33
6334	33	0.37	2.0	345	0.35	3.9
967	98	2.19	7.7	97	2.39	6.5
7239-5290	78-12	0.48-0.22	5.9-0.79	438-315	1.7-0.13	9.0-2.0
5592	0.44	0.18	0.10	213	B.D.L.	0.28
7044	42	1.8	1.1	443	0.32	2.5
6198	5	0.46	0.40	300	0.05	0.83
656	22	0.45	0.39	105	0.12	0.74
6899-5632	42-0.56	0.58-0.29	0.87-0.24	409-219	0.17-0.019	1.5-0.48
8905	1.1	0.34	0.31	329	0.085	3.9
9409	1.8	2.2	24	677	0.71	44
9121	1.6	0.95	3.3	574	0.23	10
166	0.2	0.71	9.6	130	0.24	15
9152-9054	1.7-1.6	1.6-0.64	13-1.4	654-620	0.44-0.12	16-6.0
8644	1.0	0.61	0.02	325	0.014	0.10
9168	1.5	20	11	356	1.2	31
8877	1.2	3.1	0.9	338	0.14	1.7
158	0.2	5.6	4.0	11	0.45	8.6
8951-8784	1.2-1.1	8.4-1.1	3.1-0.28	344-331	0.38-0.052	4.7-0.45
3977	0.0056	0.17	B.D.L.	15	B.D.L.	0.038
7076	7.1	1.4	5.4	117	1.0	17
5252	1.0	0.27	0.45	44	0.05	1.1
1070	1.8	0.17	1.15	39	0.18	3.3
6664.9-4296.6	2.3-0.78	0.31-0.22	1.2-0.18	107-18	0.13-0.025	2.9-0.39
5485	0.0056	0.20	B.D.L.	71	0.013	0.038

7076	0.82	0.43	4.0	117	1.0	17
6422	0.10	0.31	0.6	97	0.10	2.5
518	0.39	0.07	1.1	16	0.23	4.1
6797-5833	0.75-0.0060	0.35-0.26	1.9-0.23	108-79	0.20-0.047	5.8-1.2
4799	1.9	0.17	B.D.L.	37	0.045	0.18
5135	2.1	0.23	0.59	43	0.17	3.2
4983	2.0	0.20	0.13	40	0.076	0.56
89	0.0	0.02	0.20	2	0.043	0.93
5039-4952	2.0-2.0	0.21-0.19	0.27-0.053	41-39	0.10-0.055	1.2-0.22
3977	6.7	0.19	0.38	15	0.0060	0.63
4388	7.1	1.4	3.0	17	0.49	1.3
4208	6.9	0.35	0.91	16	0.05	0.83
185	0.2	0.57	1.19	1	0.23	0.31
4341-4110	7.0-6.9	0.56-0.20	1.4-0.59	17-16	0.18-0.031	1.0-0.64
3996	2.1	0.19	0.14	17	B.D.L.	0.22
4436	2.4	0.36	5.4	20	0.12	2.7
4235	2.3	0.25	0.47	18	0.011	0.52
104	0.1	0.04	1.40	1	0.034	0.73
4291-4185	2.3-2.2	0.27-0.22	0.75-0.20	19-18	0.028-0.0044	1.0-0.26
4335	0.0059	0.15	B.D.L.	213	B.D.L.	0.17
7951	1.6	0.90	28	1215	6.2	140
5703	0.046	0.30	2.2	603	0.2	14
1212	0.421	0.16	6.2	316	1.4	42
7109-4782	0.22-0.0091	0.41-0.20	7.1-0.78	667-593	1.4-0.020	71-3.9
4335	0.0059	0.15	B.D.L.	213	B.D.L.	0.17
7293	1.6	0.63	16	667	6.2	140
5227	0.0	0.27	2	487	0.3	19
848	0.52	0.14	4.1	166	1.5	44
5382-4742	0.032-0.0072	0.36-0.18	5.8-0.47	621-504	1.7-0.034	77-4.5
7109	0.13	0.26	1.1	1147	0.0056	1.1
7951	0.28	0.90	28	1215	0.21	16
7481	0.21	0.40	6.1	1169	0.034	5.0
251	0.06	0.19	8.8	21	0.070	5.1
7647-7315	0.27-0.17	0.42-0.33	15-3.7	1177-1156	0.11-0.018	10-3.0
3413	5.9	0.14	0.026	21	0.036	1.9
3905	11	2.5	3.3	30	1.51	30
3684	9.5	0.37	0.15	23.8	0.15	9.9
146	1.3	0.60	0.77	2.9	0.38	8.2
3829-3574	10-9.5	0.50-0.23	0.26-0.052	26-22	0.23-0.061	17-6.6

Bi (ppm)\*  
B.D.L.  
3.5  
0.44  
0.77  
0.60-0.0055  
B.D.L.  
B.D.L.  
B.D.L.  
B.D.L.  
B.D.L.  
B.D.L.  
3.46  
0.04  
0.77  
0.60-0.0055  
B.D.L.  
3.5  
0.053  
0.801  
0.72-0.0057  
B.D.L.  
0.0051  
0.0046  
0.0006  
0.0050-0.0044  
B.D.L.  
0.0061  
0.0061  
-  
0.0061-0.0061  
B.D.L.  
0.011  
0.0069  
0.0035  
0.0084-0.0053  
B.D.L.  
0.0037  
0.0033  
0.0005  
0.0035-0.0031  
B.D.L.

B.D.L.  
B.D.L.  
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B.D.L.  
B.D.L.  
B.D.L.  
B.D.L.  
B.D.L.  
0.0037  
0.0033  
0.0005  
0.0035-0.0031  
B.D.L.  
0.015  
0.004  
0.006  
0.0066-0.0025  
B.D.L.  
0.015  
0.005  
0.007  
0.0089-0.0025  
B.D.L.  
0.0037  
0.0037  
-  
0.0037-0.0037  
0.054  
3.5  
0.5  
1.0  
1.1-0.33

Deposit	Sector/Body/Section		Fe (ppm)	Co (ppm)	Ni (ppm)
María Teresa (Peru)	All samples (n = 40)	Min.	264304	B.D.L.	B.D.L.
		Max.	373972	0.087	2.2
		GM	331188	0.075	1.5
		SD	24272	0.010	0.4
		IQR	348538-319792	0.079-0.070	1.9-1.2
	Calera / Sofía D / 300 (n = 25)	Min.	264304	B.D.L.	B.D.L.
		Max.	373972	0.087	2.2
		GM	331249	0.075	1.4
		SD	25625	0.010	0.4
		IQR	348538-319792	0.079-0.070	1.8-1.2
	Calera / Sofía D / 500 (n = 4)	Min.	312109	B.D.L.	B.D.L.
		Max.	350244	B.D.L.	1.9
		GM	330803	-	1.9
		SD	15609	-	0.1
		IQR	336816-325247	-	1.9-1.8
Cerro de Maimón (Dominican Republic)	Cerro de Maimón (n = 11)	Min.	270450	1.2	B.D.L.
		Max.	316435	3.1	2.5
		GM	296657	1.8	1.7
		SD	14668	0.6	0.4
		IQR	307234-285613	2.0-1.5	1.9-1.5

B.D.L.: Below detection limit; IQR: interquartile range; GM: geometric mean; SD: standard deviation

\*Ragged ablation spectra indicate that these elements may occur as sub-microinclusions

Cu (ppm)	Zn (ppm)	Ga (ppm)	Ge (ppm)	As (ppm)	Se (ppm)	Mo (ppm)*	Ag (ppm)
340000	152	0.070	0.18	0.18	10	B.D.L.	3.2
352700	647	24	80	1.9	59	0.044	295
346217	277	4.0	1.6	0.43	25	0.027	82
3871	133	6.4	19.4	0.41	13	0.008	67
348300-343900	425-194	9.2-2.6	4.3-0.66	0.74-0.29	31-19	0.034-0.021	150-71
342000	152	0.070	0.18	0.18	10	B.D.L.	3.2
352700	529	24	80	1.9	45	0.044	295
347222	263	3.5	1.2	0.45	22	0.027	89
3155	124	6.8	20.9	0.43	8	0.008	68
349700-345000	425-192	8.1-2.6	2.2-0.64	0.80-0.29	30-16	0.035-0.021	166-76
340000	276	8.9	7.3	0.25	49	B.D.L.	44
340000	647	11	12	0.36	59	0.033	68
340000	374	10	9.4	0.30	53	0.025	52
-	170	1	2.4	0.05	4	0.007	10
340000-340000	425-303	10-9.1	11-7.8	0.33-0.28	54-50	0.028-0.022	55-48
346300	B.D.L.	3.1	4.8	0.36	60	5.4	8.3
346300	71	3.9	6.6	2.0	94	250	33
346300	71	3.4	5.6	0.86	81	40	18
-	-	0.2	0.6	0.54	11	69	7
346300-346300	-	3.6-3.3	6.1-5.3	1.3-0.56	89-79	83-23	23-15

ard deviation

15



Cd (ppm)	In (ppm)	Sn (ppm)	Sb (ppm)*	Au (ppm)*	Hg (ppm)	Tl (ppm)*	Pb (ppm)*
3.5	11	4.3	1.4	B.D.L.	B.D.L.	0.019	1.4
15	49	201	31	0.40	3.8	11	82
8.8	20	43	6.8	0.069	1.1	0.8	7.8
3.3	12	40	7.3	0.091	1.1	3.0	16.4
12-6.2	32-12	77-33	12-3.2	0.12-0.039	1.9-0.69	2.4-0.28	12-4.3
3.5	11	4.3	1.4	B.D.L.	B.D.L.	0.019	1.4
15	49	201	31	0.40	3.8	11	82
9.2	18	42	6.6	0.077	1.0	0.8	8.8
3.2	11	43	7.4	0.095	0.9	3.2	17.2
12-8.2	23-12	78-30	10-3.2	0.13-0.041	1.5-0.69	2.4-0.28	14-6.0
4.1	34	43	3.6	0.016	0.57	0.62	1.5
9.6	44	53	17	0.075	3.7	3.4	7.9
6.6	37	49	7.8	0.036	1.9	1.7	3.6
2.4	5	4	7.2	0.029	1.5	1.1	3.2
8.4-5.4	38-34	52-47	16-3.8	0.065-0.020	3.5-1.5	2.5-1.5	7.0-1.9
B.D.L.	6.7	1.2	0.05	0.015	B.D.L.	0.026	5.1
0.59	8.7	1.8	0.34	0.13	0.27	0.42	39
0.45	7.7	1.5	0.16	0.043	0.17	0.078	12
0.17	0.6	0.2	0.10	0.034	0.06	0.123	12
0.53-0.41	8.1-7.3	1.6-1.4	0.25-0.10	0.071-0.027	0.21-0.14	0.15-0.042	21-7.1

Bi (ppm)\*  
0.0092  
5.5  
1.0  
1.6  
2.5-0.54  
0.0092  
5.5  
1.0  
1.6  
2.6-0.86  
0.45  
2.5  
1.0  
1.0  
1.8-0.52  
1.5  
10  
3.8  
3.4  
8.0-2.2

Deposit	Location	ppm	Mn	Fe	Cu
Vorta (n = 8)	Romania	Max	761	2617	719
		Min	165	917	131
		GM	485	1844	334
		SD	201	584	193
		IQR	650-424	2404-1524	475-274
Eskay Creek (n = 12)	Canada	Max	4744	7112	3483
		Min	2296	3880	67
		GM	3206	4814	356
		SD	648	900	1017
		IQR	3437-3113	5284-4316	835-116
Zinkgruvan (n = 5)	Sweden	Max	532	33261	5.1
		Min	499	31735	1.9
		GM	513	32330	3.3
		SD	14	737	1.2
		IQR	524-502	33006-31751	4.0-3.1
Marketorp (n = 8)	Sweden	Max	18729	119418	14368
		Min	1684	86959	1414
		GM	2606	96260	3020
		SD	5930	12125	4246
		IQR	2373-1714	100257-88877	3290-2281
Sauda (n = 10)	Norway	Max	1449	120205	44
		Min	874	69576	22
		GM	959	77585	28
		SD	170	14797	8.6
		IQR	939-903	75796-72800	37-23
Çayeli (n = 40)	NE Turkey	Max	486	182759	210618
		Min	0.35	8.0	65
		GM	22	2857	9638
		SD	102	32933	41584
		IQR	83-5.2	16904-822	35768-6982
Laochang (n = 30)	S China	Max	4111	154203	2020
		Min	2626	121694	119
		GM	3040	130599	211
		SD	364	7814	339
		IQR	3255-2724	137135-125803	254-151

B.D.L.: Below detection limit; n.a.: not analyzed; GM: geometric mean; SD: stan

Ga	Ge	As	Se	Mo	Ag	Cd	In	Sn
363	23	n.a.	1.4	149	21	5088	0.060	0.24
75	2.7	n.a.	0.82	8.0	1.4	4169	0.040	0.050
118	8.0	n.a.	1.1	40	3.7	4734	0.049	0.080
100	7.2	n.a.	0.18	50	6.7	337	0.014	0.072
171-79	12-5.8	n.a.	1.2-1.0	71-20	5.3-2.0	5044-4562	0.055-0.045	0.080-0.055
857	14	1550	4.1	31	2785	4955	14	31
19	0.47	1	1.3	0.15	6.8	2720	0.10	2.6
86	1.2	27	2.1	0.63	102	3796	0.29	9.1
228	3.7	435	1.3	8.7	872	679	3.9	11
123-61	1.4-0.89	196-3.9	3.3-1.5	0.77-0.29	575-15	4394-3294	0.45-0.10	19.8-5.2
2.8	1.2	n.a.	1.9	n.a.	13	3943	0.32	0.25
1.9	0.82	n.a.	1.9	n.a.	4.6	3661	0.26	0.080
2.2	1.1	n.a.	1.9	n.a.	6.0	3811	0.29	0.13
0.38	0.15	n.a.	-	n.a.	3.5	126	0.026	0.077
2.4-1.9	1.2-1.1	n.a.	1.9-1.9	n.a.	6.1-4.6	3933-3718	0.31-0.27	0.21-0.090
9.6	13	n.a.	36	0.68	34	3619	28	0.81
0.61	1.8	n.a.	29	0.11	7.4	2968	25	0.090
1.5	2.7	n.a.	33	0.27	12	3246	26	0.20
3.1	3.7	n.a.	1.8	0.40	8.8	191	1.1	0.26
1.8-0.87	2.6-1.9	n.a.	34-33	0.54-0.25	14-8.7	3304-3135	27-26	0.29-0.13
8.6	1.4	n.a.	109	0.17	5.5	3488	42	1.3
4.1	0.86	n.a.	89	0.17	2.4	1973	27	0.27
5.5	1.0	n.a.	94	0.17	3.7	2170	29	0.56
1.4	0.16	n.a.	6.0	-	1.0	455	4.4	0.38
6.0-4.7	1.1-0.92	n.a.	96-90	0.17-0.17	4.5-3.1	2117-2031	30-28	0.97-0.38
n.a.	n.a.	7261	62	222	450	8957	n.a.	512
n.a.	n.a.	0.19	0.13	0.030	0.80	1051	n.a.	0.43
n.a.	n.a.	345	7.1	5.0	51	3234	n.a.	14
n.a.	n.a.	1874	12	45	98	1936	n.a.	81
n.a.	n.a.	1693-275	13-5.0	18-0.99	138-40	3430-2636	n.a.	44-4.4
117	15	36	4.0	n.a.	10	9600	566	38
2.3	2.1	0.33	0.62	n.a.	4.8	8306	66	2.2
15	3.8	2.2	1.6	n.a.	6.7	8736	184	5.6
24	2.5	9.7	1.1	n.a.	1.3	240	89	6.9
24-8.9	3.9-2.9	4.1-0.93	2.5-1.0	n.a.	7.5-5.9	8853-8562	235-164	7.7-3.9

ard deviation; IQR: interquartile range

Sb	Hg	Tl	Pb	Bi	Reference
17	n.a.	1.1	3426	0.71	Cook et al. (2009)
0.040	n.a.	0.010	0.46	0.020	
1.1	n.a.	0.034	12	0.16	
5.7	n.a.	0.50	1201	0.27	
2.3-0.75	n.a.	0.040-0.010	68-2.1	0.49-0.085	
117467	n.a.	21	50955	0.19	Cook et al. (2009)
5.1	n.a.	0.081	1.2	0.020	
1034	n.a.	0.83	467	0.041	
33692	n.a.	7.1	15639	0.056	
31973-33	n.a.	2.3-0.17	17228-13	0.053-0.028	
4.1	n.a.	n.a.	193	0.15	Cook et al. (2009)
0.12	n.a.	n.a.	1.4	0.060	
0.54	n.a.	n.a.	8.0	0.094	
1.7	n.a.	n.a.	84	0.033	
1.4-0.24	n.a.	n.a.	15-2.8	0.10-0.080	
0.080	n.a.	0.30	29	74	Cook et al. (2009)
0.040	n.a.	0.024	0.81	2.2	
0.057	n.a.	0.072	7.1	13	
0.019	n.a.	0.10	9.6	24	
0.070-0.040	n.a.	0.11-0.045	17-6.5	30-4.3	
0.10	n.a.	n.a.	n.a.	0.99	Cook et al. (2009)
0.010	n.a.	n.a.	n.a.	0.12	
0.018	n.a.	n.a.	n.a.	0.31	
0.035	n.a.	n.a.	n.a.	0.27	
0.020-0.010	n.a.	n.a.	n.a.	0.47-0.18	
3354	n.a.	97	171820	254	Revan et al. (2014)
0.68	n.a.	0.030	8.0	B.D.L.	
141	n.a.	5.2	727	-	
632	n.a.	20	26976	-	
683-52	n.a.	13-4.9	2123-549	113-6.8	
4.1	n.a.	2.6	556	0.050	Ye et al. (2011)
B.D.L.	n.a.	0.0060	0.30	B.D.L.	
-	n.a.	0.19	2.9	-	
-	n.a.	0.61	101	-	
1.6-0.10	n.a.	0.72-0.053	9.2-0.58	0.010-0.010	

Deposit	Location	ppm	Co	Ni	Zn	Ga	Ge
Vorta (n = 8)	Romania	Min	12	n.a.	199	0.11	n.a.
		Max	12	n.a.	538	2.2	n.a.
		GM	12	n.a.	391	0.71	n.a.
		SD	-	n.a.	130	0.70	n.a.
		IQR	12-12	n.a.	479-395	1.4-0.49	n.a.
Sulitjelma (n = 120)	Norway	Min	0.016	0.049	97	0.017	n.a.
		Max	28	5.1	861	19	n.a.
		GM	0.8	0.42	381	0.43	n.a.
		SD	3.7	1.3	195	4.1	n.a.
		IQR	2.6-0.27	1.0-0.20	569-329	2.3-0.098	n.a.
Çayeli (n = 69)	NE Turkey	Min	0.010	B.D.L.	7.0	n.a.	n.a.
		Max	166	200	407977	n.a.	n.a.
		GM	0.7	-	1305	n.a.	n.a.
		SD	27	-	69136	n.a.	n.a.
		IQR	1.8-0.17	7.0-0.38	30530-56	n.a.	n.a.
Kutlular (n = 9)	NE Turkey	Min	B.D.L.	0.010	70	n.a.	n.a.
		Max	0.90	0.34	5891	n.a.	n.a.
		GM	-	0.037	304	n.a.	n.a.
		SD	-	0.14	1885	n.a.	n.a.
		IQR	0.26-0.010	0.21-0.010	751-98	n.a.	n.a.

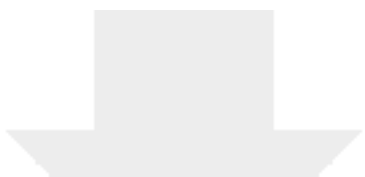
B.D.L.: Below detection limit; n.a.: not analyzed; GM: geometric mean; SD: standard deviat

As	Se	Mo	Ag	Cd	In	Sn	Sb	Au
2.8	n.a.	n.a.	18	0.48	0.0049	0.19	0.11	0.0080
5.8	n.a.	n.a.	92	3.2	0.76	0.79	5.6	0.034
4.1	n.a.	n.a.	45	1.6	0.044	0.39	2.0	0.021
1.3	n.a.	n.a.	23	1.0	0.43	0.28	1.8	0.012
5.2-3.2	n.a.	n.a.	55-41	2.2-1.3	0.39-0.014	0.73-0.23	3.8-2.8	0.033-0.019
1.2	4.4	n.a.	5.1	0.79	0.301	0.091	0.025	0.0031
5.3	926	n.a.	331	58	26	75	15	0.76
2.4	101	n.a.	71	10	2.1	4.6	0.7	0.032
1.1	267	n.a.	95	16	6.1	26	3.7	0.14
3.4-1.7	202-51	n.a.	201-32	29-3.1	6.7-0.61	51-0.43	2.0-0.18	0.088-0.013
B.D.L.	3.6	0.11	4.0	0.040	n.a.	1.1	B.D.L.	B.D.L.
5008	408	290	2549	2194	n.a.	96	3524	25
-	62	15	46	12	n.a.	15	-	-
-	124	65	375	366	n.a.	24	-	-
656-5.6	225-17	54-4.6	132-12	163-1.3	n.a.	39-8.4	117-2.2	0.64-0.063
1.7	220	0.010	1.1	0.67	n.a.	9.1	0.50	0.030
20	340	2223	45	33	n.a.	31	7.8	1.3
5.4	281	7.7	6.8	2.7	n.a.	18	1.4	0.17
5.4	46	795	18	10	n.a.	7.1	2.5	0.48
7.3-2.8	334-258	265-0.20	7.0-4.14	5.6-0.99	n.a.	22-15	1.4-0.82	0.21-0.090

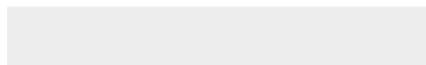
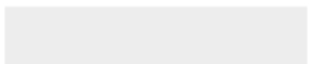
ion; IQR: interquartile range

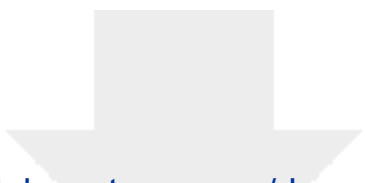
Hg	Tl	Pb	Bi	Reference
0.34	0.014	6.1	0.010	George et al. (2018)
2.6	0.17	53	2.2	
0.62	0.068	14	0.075	
0.82	0.061	26	1.3	
0.82-0.38	0.14-0.058	30-7.1	1.1-0.015	
0.051	0.0016	0.071	0.019	George et al. (2018)
0.74	3.0	28	11	
0.20	0.015	0.93	0.41	
0.17	0.50	5.4	2.2	
0.32-0.12	0.089-0.0036	1.7-0.37	0.89-0.15	
n.a.	0.010	4.0	0.030	Revan et al. (2014)
n.a.	243	244075	1000	
n.a.	1.2	307	35	
n.a.	38	30588	217	
n.a.	14-0.14	1813-43	170-7.0	
n.a.	0.020	5.0	16	Revan et al. (2014)
n.a.	6.6	494	1764	
n.a.	0.21	37	93	
n.a.	2.5	186	587	
n.a.	0.59-0.030	55-16	94-50	






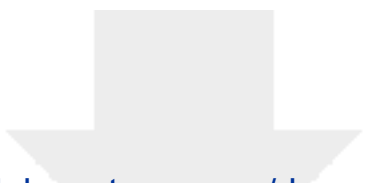
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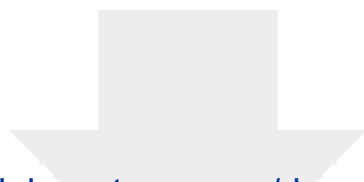


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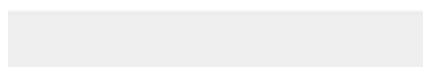
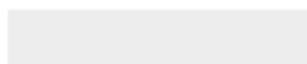





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