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Kalin Kouzmanov, Robert Moritz, Quadt Von, Massimo Chiaradia, Irena Peytcheva, et al.. Late Cretaceous porphyry Cu and epithermal Cu–Au association in the Southern Panagyurishte District, Bulgaria: the paired Vlaykov Vruh and Elshitsa deposits. *Mineralium Deposita*, 2009, 44, pp.611-646. <10.1007/s00126-009-0239-1>. <insu-00411665>

HAL Id: insu-00411665

<https://hal-insu.archives-ouvertes.fr/insu-00411665>

Submitted on 28 Sep 2009

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Late Cretaceous porphyry Cu and epithermal Cu-Au association in the Southern Panagyurishte district, Bulgaria: the paired Vlaykov Vruh and Elshitsa deposits

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Keywords: epithermal, porphyry, radiogenic isotopes, U-Pb dating, Panagyurishte district, Srednogie, Bulgaria

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Abstract

Vlaykov Vruh-Elshitsa represents the best example of paired porphyry Cu and epithermal Cu-Au deposits within the Late Cretaceous Apuseni-Banat-Timok-Srednogorie magmatic and metallogenic belt of Eastern Europe. The two deposits are part of the NW-trending Panagyurishte magmato-tectonic corridor of central Bulgaria.

The deposits were formed along the SW flank of the Elshitsa volcano-intrusive complex and are spatially associated with N110-120-trending hypabyssal and subvolcanic bodies of granodioritic composition. At Elshitsa, more than ten lenticular to columnar massive ore bodies are discordant with respect to the host rock and are structurally controlled. A particular feature of the mineralisation is the overprinting of an early-stage high-sulphidation mineral assemblage (pyrite ± enargite ± covellite ± goldfieldite) by an intermediate-sulphidation paragenesis with a characteristic Cu-Bi-Te-Pb-Zn signature forming the main economic parts of the ore bodies. The two stages of mineralisation produced two compositionally different types of ores – massive pyrite and copper-pyrite bodies. Vlaykov Vruh shares features with typical porphyry Cu systems.

Their common geological and structural setting, ore-forming processes and paragenesis, as well as the observed alteration and geochemical lateral and vertical zonation, allow us to interpret the Elshitsa and Vlaykov Vruh deposits as the deep part of a high-sulphidation epithermal system and its spatially and genetically related porphyry Cu counterpart, respectively.

The magmatic-hydrothermal system at Vlaykov Vruh-Elshitsa produced much smaller deposits than similar complexes in the northern part of the Panagyurishte district (Chelopech, Elatsite, Assarel). Magma chemistry and isotopic signature are some of the main differences between the northern and southern parts of the district. Major and trace element geochemistry of the Elshitsa magmatic complex are indicative for the medium- to high-K calc-alkaline character of the magmas. $^{87}\text{Sr}/^{86}\text{Sr}_{(i)}$ ratios of igneous rocks in the range of 0.70464 to 0.70612 and $^{143}\text{Nd}/^{144}\text{Nd}_{(i)}$ ratios in the range of 0.51241 to 0.51255 indicate mixed crustal-mantle components of the magmas dominated by mantellic signatures. The epsilon Hf composition of magmatic zircons (+6.2 to +9.6) also suggests mixed mantellic-crustal sources of the magmas. However, Pb isotopic signatures of whole rocks ($^{206}\text{Pb}/^{204}\text{Pb} = 18.13 - 18.64$, $^{207}\text{Pb}/^{204}\text{Pb} = 15.58 - 15.64$, and $^{208}\text{Pb}/^{204}\text{Pb} = 37.69 - 38.56$) along with common inheritance component detected in magmatic zircons also imply assimilation processes of pre-Variscan and Variscan basement at various scales.

U-Pb zircon and rutile dating allowed determination of the timing of porphyry ore formation at Vlaykov Vruh (85.6 ± 0.9 Ma), that immediately followed the crystallisation of the subvolcanic dacitic bodies at Elshitsa (86.11 ± 0.23 Ma) and the Elshitsa granite (86.62 ± 0.02 Ma).

Strontium isotope analyses of hydrothermal sulphates and carbonates ($^{87}\text{Sr}/^{86}\text{Sr} = 0.70581-0.70729$) suggest large-scale interaction between mineralising fluids and basement lithologies at

Elshitsa-Vlaykov Vruh. Lead isotope compositions of hydrothermal sulphides ($^{206}\text{Pb}/^{204}\text{Pb} = 18.432 - 18.534$, $^{207}\text{Pb}/^{204}\text{Pb} = 15.608 - 15.647$, and $^{208}\text{Pb}/^{204}\text{Pb} = 37.497 - 38.630$) allow attribution of ore-formation in the porphyry and epithermal deposits in the Southern Panagyurishte district to a single metallogenic event with a common source of metals.

Introduction

Spatial and genetic association between high-sulphidation epithermal precious metal and porphyry copper (\pm gold) deposits is a common feature of many mineral districts along active continental margins worldwide (e.g. Corbett and Leach 1998; Hedenquist et al. 1998; Muntean and Einaudi 2001; Sillitoe and Hedenquist 2003). Modern interpretations suggest various sources for the mineralising fluids in the shallow epithermal environment of these systems – from dominantly magmatic to mixed magmatic-meteoritic; a consensus exists concerning the prevailing magmatic origin of the fluids in the deeper porphyry environment (Arribas 1995; Sillitoe and Hedenquist 2003; Heinrich 2005; Seedorff et al. 2005). The transition between porphyry and epithermal environments, including evolution of the fluid composition and of the sulphidation state of the mineralising fluids, lateral and vertical zonation in the distribution of wallrock alteration zones and ore mineral assemblages has been noted in many Tertiary mineral districts, where tectonic and erosional processes did not significantly affect the original architecture of the magmatic-hydrothermal system (e.g. Cooke and Bloom 1990; Muntean and Einaudi 2000, 2001, Brathwaite et al. 2001; Müller et al. 2002), while Mesozoic and older systems are rarely well preserved (e.g. Kesler et al. 2005; Chambefort and Moritz 2006; Chambefort et al. 2007).

The Upper Cretaceous Apuseni-Banat-Timok-Srednogorie (ABTS) belt in Eastern Europe (Berza et al. 1998; Ciobanu et al. 2002; Heinrich and Neubauer 2002; Popov et al. 2002) is a metallogenic province enclosing important parts of the copper and gold reserves of the European continent and is the westernmost branch of the planetary-scale Tethyan Eurasian copper belt (Jankovic 1977). In Bulgaria, this belt coincides with a Late Cretaceous magmatic arc, hosting important porphyry Cu and epithermal Cu-Au deposits. Until the mid-1990's the epithermal Cu-Au deposits in the intensively mineralised central part of this zone, known as the Panagyurishte mineral district (Bogdanov 1980), were described by most of the authors as “volcanic-hosted massive sulphides”. Nowadays they are recognised as typical examples of high-sulphidation epithermal systems (Mutafchiev and Petrunov 1996; Moritz et al. 2004).

In this contribution, we studied the magmatism and ore-formation in the Elshitsa magmatic complex in the southern part of the Panagyurishte district, hosting the spatially associated Vlaykov Vruh porphyry Cu and the Elshitsa epithermal Cu-Au deposits, as well as several other small

deposits and ore occurrences. Our investigation was focused on: (1) the petrological and geochemical characterisation of the Elshitsa volcano-intrusive complex; (2) the mineral paragenesis at Vlaykov Vruh and Elshitsa; (3) radiogenic isotope tracing, and (4) high-precision U-Pb dating of magmatic and hydrothermal events. This approach allows us to provide new insights into the processes of magma generation that gave rise to magmatic-hydrothermal ore formation at the porphyry-epithermal interface, in one of the best examples of a tight porphyry-epithermal association within the Cretaceous ABTS belt in Eastern Europe.

Geological setting of the Panagyurishte district

The Panagyurishte ore district (Popov and Popov 1997, 2000; Popov et al. 2003) consists of a NNW-SSE trending alignment of porphyry and epithermal Cu-Au deposits situated 60 to 90 km east of Sofia (Fig. 1) in the central part of the Bulgarian segment of the ABTS belt known as the Srednogorie zone. The Srednogorie zone is an 80-100 km wide, east-west oriented tectono-magmatic unit situated between the Balkan zone to the north and the Rhodopean Massif to the south (Figs. 1a-b; Boccaletti et al. 1974; Aiello et al. 1977; Dabovski et al. 1991). The Central Srednogorie belongs to the undivided and rear part of the Late Cretaceous island arc system in Bulgaria (Fig. 1b; Dabovski et al. 1991; Kamenov et al. 2000).

The main characteristics of basement lithologies, Mesozoic sedimentary cover and Late-Cretaceous magmatism of the Panagyurishte district are summarized in Table 1. The Panagyurishte district presently constitutes a mosaic of horst and graben structures delimited by regional ~N110-120 trending transform faults, which resulted from transtensional, dextral strike-slip tectonics (Ivanov et al. 2001; Chambefort and Moritz 2006).

Porphyry copper mineralisation in the district is commonly centred on relatively small, approximately 1 km², subvolcanic bodies, which usually crosscut the volcanic edifices or are intruded into pre-Mesozoic basement (Strashimirov et al. 2002, 2003; Fig. 1c). Elatsite, Medet, and Assarel are the largest porphyry Cu-Au deposits, and are located in the northern part of the district, whereas Tsar Assen and Vlaykov Vruh in the southern part are much smaller (Table 1; Fig. 1c). Drew (2005) recently interpreted the three largest porphyry copper deposits in Northern Panagyurishte as being formed in extensional fracture zones, along the edges of strike-slip fault duplex.

Based on current concepts on the genesis of magmatic-hydrothermal systems (e.g. Hedenquist and Lowenstern 1994; Hedenquist et al. 2000, 2001; Einaudi et al. 2003) the epithermal high-sulphidation Cu-Au deposits in the Panagyurishte district were interpreted as genetically linked to the porphyry Cu (-Au) systems, thus confirming the early interpretation of Dimitrov

(1960) who first described these deposits as “epithermal to mesothermal epigenetic, usually structurally controlled, and formed by replacement and open space deposition processes, with preferential development of alteration zones and ore bodies within the volcano-sedimentary units”. The largest deposit of this type – Chelopech, is located in the Northern Panagyurishte, while the Southern Panagyurishte hosts only small or sub-economic ore deposits (Table 1).

The Elshitsa volcano-intrusive complex

The Late Cretaceous Elshitsa complex in the Southern Panagyurishte district is a volcano-plutonic structure composed of the volcanic rocks of the Elshitsa effusive formation and the Elshitsa pluton, as well as numerous subvolcanic and subvolcanic-hypabyssal minor intrusions and dykes (Fig. 2a; Bogdanov et al., 1970b; Ignatovski et al. 1990; Popov, 2001). *The Elshitsa effusive formation* outcrops in two narrow NW-SE elongated strips on both sides of the Elshitsa intrusion. It comprises a lower andesitic and an upper dacitic units. Lava and coarse-grained pyroclastic rocks of the two units are gradually replaced by lapilli and ash tuffs to the northeast and southwest (Popov 2001).

The Elshitsa pluton was described first as part of a larger composite intrusive body called Elshitsa-Boshulia intrusive by Boyadjiev and Chipchakova (1962), but Ivanov et al. (2002) considered it as a separate, single magmatic body within a subequatorially elongated chain of intrusions of granitic composition parallel to the Iskar-Yavoritsa shear zone (IYSZ). The intrusion is NW-SE elongated (~N120). According to Boyadjiev and Chipchakova (1962), the pluton was formed during multiple magma pulses with varying compositions – an early gabbroic phase, followed by the emplacement of granitic magma, thus producing some hybrid rock varieties. Based on major and trace element geochemistry, Daieva and Chipchakova (1997) and Kamenov et al. (2007) inferred mixing and fractional crystallisation processes to explain the compositional varieties in the Elshitsa pluton. Based on their structural, petrological, and magnetic (AMS) features (Ivanov et al. 2002) and on radiogenic isotope geochemistry and U-Pb dating (Peytcheva et al. 2008) the mafic and felsic magmas in the Boshulia intrusive (south of Elshitsa) were interpreted as products of mingling and mixing processes in a differentiated magma chamber. Georgiev and Lazarova (2003) proposed a relatively shallow level of crystallisation for the Elshitsa pluton at 750°C and 3.2 kbar, corresponding to a depth of 9 km, compared to the deeper Boshulia pluton (depth of crystallisation up to 18-19 km, based on hornblende barometry). Handler et al. (2004) reported an 84.07 ± 0.54 Ma $^{40}\text{Ar}/^{39}\text{Ar}$ age for magmatic amphibole from the Elshitsa granodiorite.

The subvolcanic and hypabyssal intrusions and dykes were described by Boyadjiev and Chipchakova (1965) as an independent dyke system postdating the emplacement of the Elshitsa

pluton. These rocks intruded the Upper Cretaceous volcanic rocks, as well as the Elshitsa pluton and the pre-Mesozoic basement (Fig. 2a). The subvolcanic bodies are subvertical, dacitic to rhyodacitic in composition and have a nearly linear-tabular shape. They are subparallel to the contacts of the Elshitsa pluton (~N120) and crop out within both the Elshitsa and the Radka volcanic strips (Fig. 2a). The subvolcanic-hypabyssal intrusions are predominantly granodioritic in composition, with variations to plagiogranite and granite porphyries. They host the porphyry-style mineralisation in the southern part of the Panagyurishte district (Kolkovski et al. 1977; Ignatovski et al. 1990; Popov 2001). The Vlaykov Vruh porphyry is the largest intrusion of this type.

As a result of the volcanic and tectonic activities, the Elshitsa magmatic complex was fragmented into four individual blocks (Fig. 2a; Popov 2001): (1) a central block limited by the Stefancho fault to the north and the Elshitsa fault to the south, uplifted by more than 1500 m, so that the Elshitsa pluton and the basement metamorphic rocks presently crop out at the same hypsometric level as the extrusive products of the Elshitsa stratovolcano; (2) volcanic rocks covered by the Chugovitsa and Mirkovo Formations are preserved in the northern block; (3) a southern block consisting of a narrow volcano-tectonic graben (Elshitsa graben) formed along a series of subparallel faults; and (4) a block occupying the southernmost part of the area, where basement lithologies and the Late Cretaceous Boshulia intrusion are in a highly uplifted position.

Geology of the Vlaykov Vruh porphyry Cu and the Elshitsa epithermal Cu-Au deposits

Among the different paired porphyry Cu and epithermal Cu-Au deposits in the Panagyurishte district, such as Chelopech-Elatsite (in the north part; Chambefort et al. 2007); Petelovo-Krassen (in the central part), Popovo Dere-Kaleto (in the south part; Ignatovski and Bayraktarov 1996), the Elshitsa epithermal Cu-Au deposit and the less than 1 km distant SE Vlaykov Vruh porphyry Cu deposit constitute the best example of the intimate spatial association of high-sulphidation epithermal and porphyry Cu systems in the district (Figs. 2b and 3). The two deposits are located along diagonal WNW-striking faults (Fig. 2b), which most probably played an important role for the lateral migration of the mineralising fluids. Despite the fact that the two mineralisation styles, occurring presently at the same crustal level, were formed at different P-T conditions, there is no evidence for any important vertical displacement between the two deposits. Due to their textural and geochemical similarities and field observations, the hypabyssal granodiorite intrusion hosting the Vlaykov Vruh porphyry deposit can be regarded as a deeper analogue of the subvolcanic dacite partially hosting the epithermal ores at Elshitsa (Bogdanov et al., 1972b; Ignatovski et al. 1990). Kouzmanov et al. (2003) provided stable isotope evidence in favour

of the close genetic relationship between the two systems, both being formed by oxidised fluids with almost identical oxygen and sulphur isotopic compositions, and evolving in time from purely magmatic in the early mineralisation stages to mixed magmatic-meteoric signatures during the later epithermal stage.

The Vlaykov Vruh porphyry Cu deposit

Geology: The Vlaykov Vruh deposit is one of the smallest porphyry Cu deposits in the Panagyurishte district (Table 1). The porphyry-copper mineralisation is mainly hosted by a granodioritic to quartz-monzodioritic stock (Figs 4a and 5a). The intrusion was emplaced along the contact between the southern parts of the Elshitsa effusive formation and the pre-Mesozoic metamorphic and granitic basement (Figs 2b and 4a; Bogdanov et al. 1972a; Strashimirov et al. 2003). The major axis of the intrusion trends N115-120; its length is over 2 km. The intrusion is cut by later dacitic dykes, which were interpreted by Bogdanov et al. (1972b) as ore-controlling. Up to 30 cm-large late-magmatic aplite-pegmatite dykes crosscut the intrusion (Fig. 5b). There is no clear evidence for time relationships between the two dyke generations.

Alteration and mineralisation: The mineral paragenesis of the Vlaykov Vruh deposit displays characteristic features of typical porphyry copper systems (Seedorff et al. 2005) with various generations of quartz-sulfide veinlets formed during successive stages of mineralisation (Fig. 6). According to Strashimirov et al. (2003), the mineralisation forms four vertical stockwork ore bodies in the deeper parts, coalescing into a single ore body in the upper part of the deposit. The main ore body was oxidised to a depth of 10-20 m into malachite-azurite ores, underlain by a 30-40 m-thick secondary enrichment zone of bornite-chalcocite-covellite ores (Bogdanov et al. 1972b; Fig. 6). Hypogene alteration including potassic, phyllic and propylitic alteration zones and mineralisation affected all rock varieties in the deposit. The Vlaykov Vruh granodiorite and the two-mica gneiss from the basement underwent potassic alteration (stage I in Figure 6), leading to the formation of shreddy biotite replacing mafic phenocrysts and the formation of tiny K-feldspar-magnetite veinlets (Fig. 5c). Aplite-pegmatite dykes, located at the southern border of the open pit were propylitically altered and cut by early stage quartz-rutile-molybdenite veins (Fig. 5d; stage II in Figure 6). Along the western and northwestern border of the open pit, both the relatively late NNW-striking quartz-pyrite veins (Fig. 5e) and the hydrothermal breccias with phyllic alteration (Fig. 5f) show very similar textures to that of the neighbouring Elshitsa epithermal deposit (see below). The main economic copper mineralisation (stage III in Figure 6) occurs as tiny veinlets of quartz – chalcopyrite – magnetite ± pyrite ± pyrrhotite or is disseminated (Fig. 5h). Polymetallic veins of quartz – sphalerite – galena – chalcopyrite ± barite (stage IV) were deposited as open space

filling in extensional fractures and were later cut by quartz-pyrite-chlorite veins (stage V; Fig. 5i). The latest hypogene mineral assemblage consists of sporadic carbonate-zeolite veinlets (stage VI).

The Elshitsa epithermal Cu-Au deposit

Geology: The Elshitsa epithermal deposit is located in the south-eastern part of the Elshitsa volcanic strip and occupies an area of 2-2.5 km length by 0.5-0.7 km width, NW of the Vlaykov Vruh porphyry Cu deposit (Fig. 3). The deposit was one of the first mines that was exploited in the Panagyurishte district, operating from 1922 until 1999. Until the 1960's, the main part of the ore was produced from massive pyrite bodies. For the last 30 years of mining activity, the underground mine production came mainly from copper and/or pyrite-copper-polymetallic bodies. A small open pit was developed in the sector Elshitsa-West in the 1990's. Elshitsa is one of the smallest epithermal deposits in the district (Table 1).

Figure 7 summarises representative Cu, Au and Ag ore grades in massive and disseminated pyrite and copper-pyrite bodies at Elshitsa. A good correlation exists between the gold and silver contents of the ore along a 1:5 line (Fig. 7a), with an average Ag/Au ratio of 10, as well as between copper and gold grades along a 1:1 line (Fig. 7b), with an average Cu/Au ratio of 1.3. The gold content in the hypogene ore is usually <5 g/t, mainly in the range 0.5-2 g/t. The copper content usually ranges between 0.5% and 2 %, with values up to 7-8 % in the richest copper bodies. The highest gold grades were registered in the richest copper ore bodies, while the massive pyrite bodies show average gold grades of 0.5-2 g/t (Fig. 7c).

The structure of the deposit is controlled by a subvolcanic dacitic body emplaced along the Elshitsa fault, elongated in NW direction (N110-115) and dipping almost vertically to the NE (75-80°). It cuts an andesitic and dacitic volcanic sequence dipping 10-30° to the N (Bogdanov et al. 1972a). The subvolcanic body is limited by the Elshitsa fault to the N and by an 80 to 120 m-large breccia zone to the S, which is the main host of the epithermal mineralisation (Fig. 3b). The latter dips steeply to the NE (65-70°) and is subparallel to the intrusive contact. At depth, it is cut by the Elshitsa fault. More than ten lenticular to columnar massive ore bodies are discordant with respect to the host rock and are structurally controlled by NW-trending normal faults (N110-120), subparallel to the contact of the dacitic subvolcanic body. Massive lens-like bodies generally show tectonic contacts, due to later Alpine reactivation of the main ore-controlling faults (Fig. 8a).

Two compositionally distinct types of ore bodies have been recognised at Elshitsa: (1) massive pyrite, mainly in the eastern sector of the deposit, closer to the Vlaykov Vruh porphyry Cu system, and (2) pyrite-chalcopyrite-quartz or copper-pyrite bodies – mainly in the central and western sectors of the deposit. The latter ore type results from the superposition of a copper-rich

mineral paragenesis upon the early massive pyrite ores. In the “Kanata” area, a small gossan (Fig. 8b) developed on the top of one of the massive pyrite bodies and was also mined out in a small open pit. It had a vertical extension of 25-30 m from the surface and total reserves of ~53000 t with average grades of 7.74 g/t Au and 21.24 g/t Ag.

Bogdanov et al. (1970a) reported massive pyrite clasts in a dacitic breccia matrix from the Radka and Elshitsa deposits (Fig. 8c) and interpreted them as epiclastic rocks resulting from the destruction of massive sulphide ore bodies, suggesting an early ore-formation in a shallow water environment coeval with the dacitic volcanism. Kouzmanov et al. (2004) described macro and micro-scale polymictic breccia-pipe and breccia-dyke structures with similar pyrite clasts, cutting the massive dacite volcanic rocks as well as the massive ore bodies at Radka, thus suggesting intra-mineralisation formation of the latter due to fluidisation of rocks and ore fragments along upward-escaping fluid channelways, which is a common feature to many hydrothermal systems formed in a porphyry-to-epithermal transitional environment (Corbett and Leach 1998).

Replacement textures at Elshitsa are common. Generally, laminated dacitic lapilli tufs are almost entirely replaced by fine-grained pyrite (Fig. 8d). By contrast, only the matrix of the breccia zone bordering the subvolcanic dacite body to the south has been replaced by a fine-grained pyrite ± silica assemblage, which preserves subangular to rounded dacite clasts (Fig. 8e). Veins are less abundant, but when present, they provide good textural evidence for time relationship between ore-forming stages (Fig. 8f). The most common textures are the massive ones (Fig. 8g), gradually replaced by disseminated mineralisation towards the periphery of the ore bodies.

Alteration and mineralisation: Wallrock alteration at Elshitsa is structurally and morphologically related to the ore-controlling faults (Fig. 4b). Radonova (1967a, 1967b, 1970) described the alteration zoning in the ore bodies consisting of inner silica-rich zone, phyllic zone with quartz-sericite-pyrite ± diaspore, quartz-sericite zone with dumortierite described at two localities, and an outer zone with chlorite-epidote-albite propylitic assemblage. The ore bodies are typically hosted by the phyllic alteration zone (± diaspore and/or dumortierite). Chipchakova and Stefanov (1974) noted the presence of a 2M₁ polytype of sericite associated with the early massive pyrite bodies and of lower-temperature 1M polytype associated with the later copper mineralisation. In addition to this time-zonation, Kouzmanov (2001) reported a lateral zonation of the polytypes of illite – from 2M₁-rich zones in the vicinity of the Vlaykov Vruh deposit to 1M-rich zones in the northwesternmost part of the deposit at Elshitsa-West. This illite zonation is most probably related to a temperature trend laterally decreasing away from the intrusion-centred porphyry copper system. Apart from this zonation, Dimitrov (1985) noted the presence of an alunite-bearing advanced argillic zone, around the uppermost level of a major massive pyrite body at Elshitsa.

The epithermal deposits in the Panagyurishte district display numerous paragenetic similarities including: (1) an early massive to disseminated pyrite stage; (2) an intermediate Au-bearing Cu-As-Sn-Bi-S stage; and (3) a later base-metal (Zn-Pb \pm Sb, Au) stage (Bogdanov and Bogdanova 1974b; Moritz et al. 2004). The mineral paragenesis at Elshitsa slightly differs from those of typical high-sulphidation epithermal deposits described worldwide (Fig. 9) and even from that of other epithermal Cu-Au deposits in the Panagyurishte district. In the deposit, typical high-sulphidation minerals such as enargite, covellite and bornite (Einaudi et al., 2003) are much less abundant than at Chelopech, Radka and Krassen, and advanced argillic and argillic alteration zones are only restricted to the uppermost levels of the system. These features are attributed to a greater depth of epithermal ore-formation at Elshitsa, directly linked to the porphyry Cu system at Vlaykov Vruh, and/or to a deeper erosional level in the southern part of the district, compared to the central and northern parts.

The stage-I massive pyrite bodies (Fig. 9) mainly consist of fine-grained and colloform pyrite (Fig. 8h), with rare rutile, cassiterite and covellite inclusions. Enargite and goldfieldite inclusions occur sporadically (Fig. 8i), indicating a high- to very high-sulphidation state of the mineralising fluids (Einaudi et al. 2003). The early pyrite generation is As-bearing, generally showing oscillatory growth zoning (Fig. 8j). Electrum and gold occur as submicroscopic to rare microscopic inclusions in pyrite, as well as invisible gold in the pyrite structure (Bogdanov et al. 1997). Following the formation of intramineralisation breccia-dykes with clasts of massive pyrite, the minerals of the Cu-stage II (Fig. 9) precipitated as replacement of the fine-grained and colloform pyrite of stage I (Figs. 8k and 8l) or formed separated ore bodies. The Cu-Bi-Te-Pb-As stage II assemblage mainly consists of chalcopyrite, tennantite and minor sphalerite and galena, with rare As-Sn-Bi-bearing sulphosalts and various Cu-Bi-Au-Ag tellurides (Fig. 9; Kouzmanov et al. 2005). Native gold and electrum occur mainly as inclusions in chalcopyrite (Fig. 8m). Sphalerite-galena-tetrahedrite-chalcopyrite-barite is the main mineral association of the polymetallic stage III (Figs. 8n and 9). It forms narrow veins (Fig. 8f) or selectively replaces the mineral assemblages of the previous mineralisation stages. Most of the microscopic gold at Elshitsa is associated with this assemblage (Fig. 8o). Bogdanov et al. (1997) reported a progressive increase of the gold fineness from the early massive pyrite to the late polymetallic mineralisation stage. Based on the mineral associations, an intermediate sulphidation state can be inferred for the mineralising fluids of stages II and III (Einaudi et al. 2003). Late massive anhydrite veins (stage V in the paragenetic sequence of Figure 9) crosscut the earlier formed sulphides (Fig. 8f) and are partially replaced by supergene gypsum in the uppermost parts of the ore bodies (Fig. 8p). Registered temperature and salinity intervals for the mineralising fluids at Elshitsa (Fig. 9) are typical for high-sulphidation epithermal systems (Arribas 1995).

The Radka epithermal Cu-Au deposit, located in the central part of the Radka volcanic strip (Fig. 2a) shares many common features with the Elshitsa epithermal deposit (Bogdanov and Bogdanova 1974a; Tsonev et al. 2000; Kouzmanov et al. 2002, 2004), therefore several samples of sulphides and sulphates from Radka were also analysed for comparison in the present study.

Whole rock geochemistry

The analytical techniques used in the present study are summarised in Appendix 1. Major oxide and trace and REE analyses of Late Cretaceous volcanic and intrusive rocks from the southern part of the Panagyurishte district are reported in Appendix 2 and presented in Figures 10 and 11. Published data for the same area are also plotted for comparison. The TAS diagram (Fig. 10a) illustrates a typical trend for the magmas in the Elshitsa magmatic complex – starting from basaltic and evolving to rhyolitic compositions, with few data plotting in the trachytic field. Most of the igneous rocks plot within the medium- and high-K fields in the K_2O versus SiO_2 diagram (Fig. 10b). The K_2O/Na_2O ratio varies between 0.2 and 2.1, but in most cases is <0.8 . The igneous rocks from Southern Panagyurishte district define a typical calc-alkaline trend on the AFM diagram (Fig. 10c).

All studied rocks show similar incompatible trace element patterns (Fig. 10d) with an enrichment of large-ion lithophile elements (LILE: K, Rb, Ba, Pb) and depletion of high field strength elements (HFSE: Nb and Ti), which are typical features of magmas from convergent margin tectonic settings (Saunders et al. 1980). For two of the andesite and one of the rhyodacite samples, concentrations of mobile elements (Rb, Ba and K) have been most probably affected by a hydrothermal overprint, even if in thin sections only minerals typical for the propylitic alteration assemblage have been observed. Consistent differences in the trace-elements pattern between andesites and more felsic varieties can be observed (Fig. 10d).

Chondrite-normalised REE patterns of studied rock varieties are presented in Figure 10e. Felsic rocks (rhyodacites including subvolcanic varieties, Vlaykov Vruh granodiorite and Elshitsa granite) show very similar REE patterns, with a moderate light REE enrichment, and a flat (andesites) to slightly concave (felsic rocks) heavy REE portion. All studied rocks from the Elshitsa magmatic complex are characterised by an absence of Eu anomaly.

As shown in Harker variation diagrams (Fig. 11), most of the igneous rocks from the Elshitsa magmatic complex form near-linear to curvilinear trends with decreasing TiO_2 , FeO^* , MgO , CaO and P_2O_5 concordant with increasing SiO_2 . The marked inflections of some patterns at approximately 55 wt% SiO_2 suggest fractionation of olivine, hornblende, a Ti-bearing phase (possibly magnetite), apatite and clinopyroxene during the process of magmatic crystallisation.

Extensive plagioclase fractionation most probably did not take place, as indicated by the absence of any Eu anomaly and negative Sr-SiO₂ correlation. The strong decrease of Al₂O₃ and Rb (\pm K₂O) detected in the felsic compositions is most probably due to K-feldspar fractionation at approximately 65-70 wt% SiO₂. The very similar curvilinear pattern of Zr, La and U suggests strong zircon fractionation at approximately 70 wt% SiO₂, typical for felsic magmas. Fractionation of La at this composition in parallel with a decrease of CaO could be also indicative for apatite fractionation.

Radiogenic (Pb, Sr, Nd) isotope geochemistry

Lead isotope data for whole-rock samples from the southern part of the Panagyurishte district are plotted in Figures 12a-b. For all samples [U], [Th] and [Pb] were measured and the ratios corrected at 85 Ma for in situ produced radiogenic Pb. The range of initial values for Late Cretaceous igneous rocks is $^{206}\text{Pb}/^{204}\text{Pb} = 18.13 - 18.64$, $^{207}\text{Pb}/^{204}\text{Pb} = 15.58 - 15.64$, and $^{208}\text{Pb}/^{204}\text{Pb} = 37.69 - 38.56$ (Table 2). Basement lithologies, including gneisses and Variscan intrusions, plot within the trends defined by Late Cretaceous igneous rocks. One sample of a Turonian sedimentary rock displays the most radiogenic composition within the studied suite of rocks. Within the Late Cretaceous rocks, the subvolcanic rhyodacites display the largest data scatter, while the data from the Vlaykov Vruh granodiorite, together with the Elshitsa granite, and andesites and rhyodacites form a tighter cluster. On the $^{208}\text{Pb}/^{204}\text{Pb}$ versus $^{206}\text{Pb}/^{204}\text{Pb}$ diagram (Fig. 12a), Late Cretaceous igneous rocks define a well expressed trend overlapping and subparallel to the orogenic and upper crust growth curves of Zartman and Doe (1981). On the $^{207}\text{Pb}/^{204}\text{Pb}$ versus $^{206}\text{Pb}/^{204}\text{Pb}$ diagram (Fig. 12b), our data points overlap with the orogenic growth curve or plot in a trend subparallel to it. Our whole-rock lead isotope data indicate Pb from mixed crustal-mantle or dominantly crustal sources. These data are partially overlapping with the trend of von Quadt et al. (2005) for Late Cretaceous igneous rocks from a north-south profile across the whole Panagyurishte district. Both trends crosscut the field of Variscan intrusions, defined by data of Amov et al. (1973). Age-uncorrected data for gneisses from the Central Srednogie (Amov et al. 1982) are also presented on the diagrams for comparison.

Lead isotopic compositions of sulphides from different mineralisation stages in the Elshitsa and Radka epithermal and Vlaykov Vruh porphyry Cu deposits are listed in Table 3 and plotted in Figures 12c-d together with data of Amov et al. (1974). The data points form a well differentiated cluster ($^{206}\text{Pb}/^{204}\text{Pb} = 18.432 - 18.534$, $^{207}\text{Pb}/^{204}\text{Pb} = 15.608 - 15.647$, and $^{208}\text{Pb}/^{204}\text{Pb} = 37.497 - 38.630$), approximately within the analytical error. Only one sample of pyrite from a gneiss-hosted quartz-chalcopyrite-pyrite porphyry vein at Vlaykov Vruh shows a lower $^{206}\text{Pb}/^{204}\text{Pb}$ ratio. There is

no correlation between stages of mineralisation and Pb isotopic composition of ore minerals from Elshitsa, Radka and Vlaykov Vruh. Ore samples from the Southern Panagyurishte district plot within the same range of $^{206}\text{Pb}/^{204}\text{Pb}$ ratios as the host Late Cretaceous intrusive and volcanic rocks, with slightly more radiogenic $^{207}\text{Pb}/^{204}\text{Pb}$ and $^{208}\text{Pb}/^{204}\text{Pb}$ compositions. Lead isotopic compositions of ore minerals from epithermal and porphyry Cu (-Au) deposits from the Central and Northern Panagyurishte are plotted on Figures 12c-d for comparison (see discussion below).

Strontium isotope analyses of various basement lithologies and Late Cretaceous igneous and sedimentary rocks from the southern part of the Panagyurishte district are given in Table 4. After age correction for 85 Ma, initial strontium isotope data were reported in Figure 13 together with some published data for the southern and northern parts of the district. $^{87}\text{Sr}/^{86}\text{Sr}$ ratios for basement lithologies range from 0.70846 to 0.71600. In the northern part of the district, the basement metamorphic rocks have $^{87}\text{Sr}/^{86}\text{Sr}$ ratios from 0.7090 up to 0.7329 (90 Ma age correction). Late Cretaceous sedimentary rocks, including a pre-magmatic Turonian marl, and marls from the Mirkovo and Chugovitsa Formations, yield very similar $^{87}\text{Sr}/^{86}\text{Sr}$ ratios that fall within the range of Late Cretaceous seawater $^{87}\text{Sr}/^{86}\text{Sr}$ ratios (Koepnick et al. 1985). Data points for Late Cretaceous igneous rocks from Southern Panagyurishte plot within a restricted interval of 0.70464 to 0.70612. The Elshitsa granite, representing the plutonic part of the Elshitsa magmatic complex, has a less radiogenic composition and clearly differs from the Vlaykov Vruh granodiorite, andesites and rhyodacites, including subvolcanic varieties (Fig. 13). There is no significant correlation among the initial $^{87}\text{Sr}/^{86}\text{Sr}$ ratio and SiO_2 content (Fig. 14a), neither with Sr and Rb concentration; nor with the Rb/Sr ratio.

The initial $^{143}\text{Nd}/^{144}\text{Nd}$ ratios for Late Cretaceous igneous rocks range from 0.51241 to 0.51255 (Table 4). On the SiO_2 versus $^{143}\text{Nd}/^{144}\text{Nd}_i$ diagram, the Southern Panagyurishte magmas plot between the well defined field of the Banatitic rocks of South Romania (Dupont et al. 2002), which are the most radiogenic and the Chelopech magmas (Stoykov et al. 2004), which show the least radiogenic composition (Fig. 14b).

In a $^{87}\text{Sr}/^{86}\text{Sr}_{(i)}$ versus $^{143}\text{Nd}/^{144}\text{Nd}_{(i)}$ diagram (Fig. 14c), the Southern Panagyurishte Late Cretaceous magmatic rocks lie on the mantle array and form a cluster between the EM-I and EM-II fields (Fig. 14c). Such data distribution could be due to primary magma source differences and/or contamination of mantellic magmas by crustal materials at middle/upper crustal levels. In comparison with the Chelopech magmas and the Capitan Dimitriev pluton, our samples show more radiogenic Sr isotopic signatures and Nd isotopic compositions intermediate between the two (Fig. 14c).

Carbonates and sulphates precipitated relatively late in the evolution of the hydrothermal systems in the southern part of the Panagyurishte district (Figs 6 and 9). The strontium isotopic

compositions of hydrothermal barite (Elshitsa, Radka and Vlaykov Vruh deposits), calcite (Elshitsa), and anhydrite (Elshitsa and Radka) fall between the Sr isotopic composition of the basement rocks and the one of the Late Cretaceous magmatic rocks partially overlapping the range of the latter (Fig. 13). They range from 0.70581 to 0.70729 (Table 5, Fig. 13). One sample of barite from the Vlaykov Vruh deposit has a composition very similar to the initial $^{87}\text{Sr}/^{86}\text{Sr}$ ratio of the host granodiorite intrusion. Two calcite samples from the polymetallic stage-III in Elshitsa yield less radiogenic $^{87}\text{Sr}/^{86}\text{Sr}$ ratios than cogenetic barite. Anhydrite from the Elshitsa and Radka epithermal deposits shows the largest scatter in composition. Compared to Late Cretaceous seawater and Upper Cretaceous marls from the Southern Panagyurishte, hydrothermal sulphates and carbonates have less radiogenic composition.

The Sr concentration of anhydrite from the Elshitsa and Radka deposits covaries with its $^{87}\text{Sr}/^{86}\text{Sr}$ ratio (Fig. 15a), and yields a linear trend that is explained by mixing of fluids from different sources during the late stages of ore-formation in the epithermal deposits. The hypothetical end-member fluids are attributed to: (A) a Sr-rich and radiogenic fluid source, and (B) a Sr-poor and less radiogenic fluid source (Fig. 15a). Extensive fluid circulation through basement lithologies (metamorphic rocks, Variscan granites and/or Turonian sedimentary rocks) could explain type (A) fluid, while type (B) fluid can be generated by fluid circulation through the Late Cretaceous igneous rocks or could simply represent Late Cretaceous magmatic ore-fluid. To better constrain the origin of the late-stage anhydrite-precipitating fluids in the epithermal deposits from the Southern Panagyurishte, their $\delta^{18}\text{O}$ value (Kouzmanov et al. 2003) has been plotted with respect to the $^{87}\text{Sr}/^{86}\text{Sr}$ composition of the studied anhydrite samples (Fig. 15b). The few analytical points fall between the isotopic composition of local meteoric water and the isotopic composition of primary magmatic water, thus suggesting anhydrite precipitation from isotopically exchanged meteoric fluids or most probably from mixed oxidised magmatic and meteoric fluids. No relationship between anhydrite precipitation and Senonian seawater can be inferred based on Figure 15b.

U-Pb geochronology

Five samples of Late Cretaceous igneous rocks hosting the Elshitsa and Vlaykov Vruh deposits have been used for the U-Pb geochronological study (Fig. 2b). They come from: the Elshitsa granite (AvQ 029); a subvolcanic rhyodacite body cut by epithermal veins in the southwestern part of the Elshitsa deposit (AvQ 032); one rhyodacite affected by quartz-sericite-pyrite alteration from the southwestern part of Elshitsa (AvQ 101); the Vlaykov Vruh granodiorite porphyry (AvQ 031), and one aplite-pegmatite dyke crosscutting the latter (AvQ 099). Dating the hydrothermal activity in the two neighbouring deposits has been attempted with one sample of

hydrothermal rutile from the Vlaykov Vruh deposit (VV 46/99) and two samples of hydrothermal rutile and anatase from the phyllic alteration zone of the Elshitsa-West deposit (E 63/99 and E 64/99; Fig. 2b).

Concordant ages have been obtained only for zircons from the Elshitsa granite and the Elshitsa subvolcanic dacite. Only air-abraded zircon grains yielded concordant dates, non-abraded zircons usually showing lead loss (Table 6). Four abraded grains from the Elshitsa granite, representing the plutonic part of the magmatic system, are concordant within analytical error and reveal an intrusive concordia age of 86.61 ± 0.31 Ma (mean $^{206}\text{Pb}/^{238}\text{U}$ age of 86.62 ± 0.02 Ma). The $\varepsilon\text{-Hf}$ values for concordant zircons range from +6.2 to +7.9 (Table 7). The Elshitsa subvolcanic dacite shows an intrusive concordia age (based on three concordant grains, within analytical error) of 86.11 ± 0.27 Ma (mean $^{206}\text{Pb}/^{238}\text{U}$ age of 86.11 ± 0.23 Ma) and $\varepsilon\text{-Hf}$ values ranging from +7.2 to +9.6 (Table 7). Three of the analysed zircons from this sample are significantly older suggesting the presence of an inherited component. Two of the discordant zircons define an upper intercept at 465 ± 7 Ma, indicating an important assimilation of Paleozoic crustal material by the Late Cretaceous magmas. For convenience, only concordant zircons are plotted in Figure 16. Because of the presence of inherited components in the zircons from the Elshitsa volcanic rocks, as well as from the Vlaykov Vruh granodiorite (Fig. 5a) and pegmatite (Fig. 5b), these three samples display discordant ages, not allowing us to determine precisely their crystallisation age.

Among the three studied samples of hydrothermal Ti-oxides, only one sample of rutile from the high-temperature porphyry mineralisation at Vlaykov Vruh (Fig. 5g) revealed a mean $^{206}\text{Pb}/^{238}\text{U}$ age of 85.6 ± 0.9 Ma (Fig. 16c). The low-temperature TiO_2 -varieties from the epithermal ore bodies at Elshitsa were not suitable for U-Pb analyses due to their very low U-content (< 5 ppm U) and hence very low radiogenic Pb addition.

Discussion

Origin of magmas in the southern part of the Panagyurishte district

At a district scale, the Elshitsa magmatic complex is the only one in which the deep plutonic, the subvolcanic and the extrusive facies are preserved and exposed at the present-day surface. For this reason several porphyry-epithermal pairs have been described here, mainly hosted by the Elshitsa and Radka volcanic strips, preserved along the SW and NE margins of the Elshitsa pluton, respectively. The Elshitsa magmatic complex shares geochemical characteristics with typical subduction-related magmatic series defined in other parts of the ABTS belt, such as the Eastern Srednogorie (Boccaletti et al. 1974, 1978; Stanisheva-Vassileva 1980), the Timok zone in

Serbia (Karamata et al. 1997; von Quadt et al. 2003) and the Banat zone in Romania (Berza et al. 1998; Dupont et al. 2002). The petrological and geochemical data reported in this paper suggest existence of continuous medium- to high-K, calc-alkaline magmatic series in the Elshitsa magmatic complex. The radiogenic isotope geochemical data presented in the previous sections provide fundamental constraints on the petrogenesis of the Elshitsa suite.

Many lines of evidence suggest a major role of fractional crystallisation in the evolution of magmas in the southern Panagyurishte district. These include: (1) the close spatial and probably temporal association of the eruptive products of the lower andesitic and the upper dacitic volcanic units; (2) the increase of alkalis, LILE and HFSE and the decrease of TiO_2 , FeO^* and MgO with increasing SiO_2 (Figs. 10 and 11); (3) variations of incompatible trace elements with trends from low abundances in andesites to higher abundances in dacites and rhyodacites and the felsic subvolcanic rocks; and (4) overall increase of total REE contents with increasing SiO_2 (Fig. 10e). Most probably fractionation of olivine, hornblende, Ti-bearing magnetite, apatite, clinopyroxene, zircon and possibly K-feldspar has played an important role in the process of magmatic crystallisation at Elshitsa. The general absence of Eu anomalies suggests a reduced fractionation of plagioclase.

The occurrence of cognate gabbroic inclusions and lenses in the most evolved rocks (granodiorite to granites) in the western parts of the Elshitsa pluton (Boyadjiev and Chipchakova 1962; Dabovski 1963) indicates that fractional crystallisation was accompanied to some extent by magma mixing and mingling at various stages of differentiation, as suggested by Ivanov et al. (2002) and Peytcheva et al. (2008) for the Boshulia pluton, located south of the Elshitsa magmatic complex.

The geochemical and isotopic variations in the Elshitsa magmatic complex also suggest involvement of various crustal components in the evolution of the magmatic series. In fact, many of the isotopic signatures of the felsic magmas seem to favour a model involving contamination of mantle-derived magmas during ascent through the continental crust. These include: (1) occurrence of pre-Mesozoic basement xenoliths in the subvolcanic intrusions and dykes, as well as in the periphery of the Elshitsa pluton (Boyadjiev and Chipchakova 1962, 1965; Dabovski 1963); (2) systematic inherited basement components in the magmatic zircons from dacites and rhyodacites and the subvolcanic felsic varieties; (3) the two samples from the Elshitsa granite show the strongest mantellic Sr isotopic signatures among the analysed suite of magmatic rocks, indicating that at deeper level (magma chamber), the contamination process was less pronounced, than in the subvolcanic and extrusive facies in which the magmas travelled across additional 3 to 4 km upper crust until they reached the site of crystallisation; (4) combination of Nd and Sr isotope characteristics of the Elshitsa magmas explicitly indicates mixed crustal-mantle signatures similar

to the Capitan Dimitrievo pluton and other Late Cretaceous magmatic rocks from the Banat area in Romania in the northern sector of the ABTS belt (Fig. 14c); and (5) ϵ -Hf values of +6.2 to +8.9 of magmatic zircons from the felsic intrusions and subvolcanic bodies also support a mixed crustal-mantle origin for the magmas at Elshitsa.

The Pb isotope data are more equivocal. In general, the southern Panagyurishte magmas show large variations in the $^{206}\text{Pb}/^{204}\text{Pb}$ ratio with only minor variations in $^{207}\text{Pb}/^{204}\text{Pb}$ and $^{208}\text{Pb}/^{204}\text{Pb}$ (Fig. 12). These Pb isotope signatures seem to be a general feature of the Central Srednogorie arc magmatism as a whole and cannot be explained by a single model of assimilation of the Variscan basement. In Figure 12, the Elshitsa suite samples form a nearly linear array, which most probably reflects mixing of two isotopically distinct components in terms of $^{206}\text{Pb}/^{204}\text{Pb}$ and similar or very close $^{207}\text{Pb}/^{204}\text{Pb}$ and $^{208}\text{Pb}/^{204}\text{Pb}$ ratios, however the limited database for the Pb isotopic compositions of the Balkanide-type metamorphic complex and the Variscan granitoids do not allow correct quantification. In Figure 12, fields for gneisses and Variscan intrusions from the Central Srednogorie (data from Amov et al. 1973, 1982) are not age-corrected for 85 Ma because the published Pb isotope data have no concentration values and thus the basement lithologies are difficult to define as potential mixing end-members. Alternatively, the existing data could be tentatively interpreted as a result of: (1) large-scale mantle source heterogeneities, given also the evidence that on the $^{87}\text{Sr}/^{86}\text{Sr}_{(i)}$ versus $^{143}\text{Nd}/^{144}\text{Nd}_{(i)}$ diagram, the Chelopech and the southern Panagyurishte magmatic rocks plot along a trend between EM-I and EM-II (Fig. 14c), and/or (2) assimilation by the parental mantellic magmas of dominantly upper continental crust material in the Southern Panagyurishte and of lower continental crust material in the northern part (Chelopech), possibly due to different depth of magma chamber formation.

In a regional context some particularities of the Late Cretaceous magmas from different parts of the ABTS belt can be inferred: (1) the available Sr-Nd isotopic data set for primitive magmas is very restricted - most of the published data is for relatively evolved compositions, thus making very difficult correct interpretation of the source effect; (2) banatites from South Romania (Dupont et al. 2002) have isotopically the most primitive compositions in the ABTS belt (generally higher $^{143}\text{Nd}/^{144}\text{Nd}_{(i)}$ and lower or partially overlapping $^{87}\text{Sr}/^{86}\text{Sr}_{(i)}$ compositions than the Late Cretaceous magmas in Central Srednogorie). Several scenarios are possible to explain such a difference: inhomogeneity of the metasomatised mantle edge along the subduction front during the Late Cretaceous; different thickness of the crust and/or assimilation of different middle-upper crustal material in different parts of the arc; (3) Southern Panagyurishte magmas display more primitive compositions compared to Chelopech in the Northern Panagyurishte, due to a difference in the magma source or a different nature of the assimilated pre-Mesozoic basement (Balkanide-

type high-grade metamorphic rocks in the south versus low-grade Diabase-Phyllitoid complex in the north; see above).

Origin of porphyry and epithermal deposits in the southern Panagyurishte district

In the southern Panagyurishte district, porphyry Cu and epithermal Cu-Au deposits are genetically and spatially related to the subvolcanic NW-trending intrusions and dykes, cropping out in the volcanic strips on both sides of the Elshitsa pluton (Fig. 2a). Both deposit types are structurally controlled and were formed along WNW-trending steeply dipping diagonal faults (N110-120). Mineralisation processes and stable isotopic signatures in both systems show many similarities, favouring the interpretation of a common origin of mineralising fluids and ore metals and a very probable close time relationship.

One of the still debated questions in the porphyry-epithermal models is the timing of ore formation in the porphyry and epithermal environment in complex magmatic-hydrothermal systems. Few case studies (e.g. Far Southeast-Lepanto, Philippines: Arribas et al. 1995; Hedenquist et al. 1998; Maricunga belt, northern Chile: Muntean and Einaudi 2001) tried to elucidate this question, reporting almost synchronous ($\leq 100\,000$ years) to clearly separate in time (as much as 900 000 years) porphyry and high-sulphidation ore formation in which the porphyry fluid and alteration-mineralisation is reported to be directly related to the spatially associated epithermal ore bodies.

In this contribution, we attempted to date the timing of magmatic and hydrothermal processes using high-precision U-Pb dating in the case of Vlaykov Vruh porphyry and Elshitsa epithermal pair. Unfortunately, only a few samples were suitable for correct dating and only the timing of porphyry ore formation at Vlaykov Vruh was determined with an age of 85.6 ± 0.9 Ma, immediately following or partially overlapping within analytical error the crystallisation of the subvolcanic dacitic bodies at Elshitsa dated at 86.11 ± 0.23 Ma (mean $^{206}\text{Pb}/^{238}\text{U}$ age). The Elshitsa granite, representing the deeper magma chamber level of this magmatic complex yielded the oldest age with 86.62 ± 0.02 Ma (mean $^{206}\text{Pb}/^{238}\text{U}$ age), but also overlapping within analytical error with the subvolcanic bodies. The timing of epithermal ore formation at Elshitsa still remains uncertain. However we believe that it was directly related to the porphyry ore formation at Vlaykov Vruh based on: (1) field evidence indicating a gradual transition between the Vlaykov Vruh granodiorite and the subvolcanic dacite and rhyodacite dykes at Elshitsa; (2) the observed lateral outward zonation from high- to lower-temperature alteration mineral assemblages from the porphyry Cu system at Vlaykov Vruh towards the Elshitsa-West sector of the epithermal system (Kouzmanov 2001); and (3) the almost identical sulphur and oxygen isotopic compositions of hydrothermal minerals in both systems, with a characteristic decrease of $\delta^{34}\text{S}$ values through time, attributed to

progressive oxidation of the fluids and sulphur isotope fractionation due to hydrolysis of magmatic SO₂ (Kouzmanov et al. 2003).

Lead isotopic compositions of sulphides from the studied porphyry and epithermal deposits have very similar values (Figs. 12c-d). This reveals that both deposit types belong to the same metallogenic event and have a similar source of Pb and most probably of the other metals. As the ore samples from Vlaykov Vruh, Elshitsa and Radka deposits plot exactly in the middle of the trend defined by the Late Cretaceous magmatic rocks on the Pb isotope diagrams (Figs. 12a-b), the most plausible interpretation is that Pb is of magmatic origin, and that it was directly released by the ore-related Late Cretaceous magmas. The lead isotopic signature of fine-grained pyrite from the early massive pyrite bodies at Elshitsa, and from “ore clasts” have identical compositions as the younger copper and polymetallic ores, thus supporting the common epithermal origin of the mineralisation at Elshitsa as a whole, and contradicts the two-stage model proposed earlier by Bogdanov (1984).

The Sr isotopic compositions of hydrothermal minerals from the epithermal Cu-Au deposits fall between the Sr isotope ratios of Upper Cretaceous magmatic rocks and the more radiogenic basement rocks (Fig. 13). From a genetic point of view, the rhyodacitic volcanic and subvolcanic rocks are the most likely source of Sr for the epithermal mineralisation, since they are the major host rocks. The more radiogenic composition of hydrothermal minerals with respect to the host felsic rocks shows that there was an additional Sr source. Figure 15a suggests an extensive circulation of the hydrothermal fluids through basement lithologies, such as metamorphic rocks and Variscan granitoids with radiogenic Sr, in addition to the Late Cretaceous magmatic source of Sr. Strontium and oxygen isotopic compositions of hydrothermal anhydrites from the Radka and Elshitsa deposits (Fig. 15b) allow us to exclude Late Cretaceous seawater as a potential source of external fluids involved in the magmatic-hydrothermal system.

The geochemical features, alteration, ore mineralogy and paragenesis, including lateral and vertical zonations in the Elshitsa epithermal Cu-Au deposit, described above, allows us to attribute the deposit to the deep part of a high-sulphidation epithermal system, according to the terminology of Hedenquist et al. (2000) and Sillitoe and Hedenquist (2003), genetically related to the adjacent Vlaykov Vruh porphyry Cu system. Local overprint of early stage typical high-sulphidation to very high-sulphidation mineral assemblage (in the sense of Einaudi et al. 2003) consisting of pyrite ± enargite ± covellite ± goldfieldite ± alunite and illite by intermediate-sulphidation assemblages (with characteristic Cu-Bi-Te-Pb-Zn signatures) constituting the main economic mineralisation at Elshitsa is observed. The latter telescoping feature is probably a common characteristic of high-sulphidation systems, no matter if they are hosted by magmatic or carbonate rocks, as described by Jannas et al. (1999) for the El Indio deposit in Chile, Baumgartner et al. (2008) for the Cerro de Pasco polymetallic deposit in Peru, and Bendezú (2007) for the Colquijirca district in Peru.

Southern versus northern Panagyurishte Late Cretaceous magmas and ore deposits

Progressive younging of the magmatism from ~92 Ma in the north (in the area of Elatsite – Chelopech) to ~78 Ma in the south (Capitan Dimitriev) correlates with a north-to-south geochemical trend of decreasing crustal input into the mantle-derived magmas (von Quadt et al. 2005) and is generally interpreted as a direct consequence of slab roll-back or hinge slab retreat during oblique subduction (Neubauer 2002; Lips 2002; Handler et al. 2004; Moritz et al. 2004; von Quadt et al. 2005; Kamenov et al. 2007).

The northern part of the Panagyurishte district hosts the largest porphyry Cu and epithermal Cu-Au deposits, compared to the southern part where only relatively small deposits were formed (Table 1). The most illustrative example for such N-S trend is given by the Pb isotope signatures of sulphide minerals from northern and southern deposits in the district (Figs. 12c-d), showing an almost linear tendency with increasing radiogenic compositions towards the north. Similar arguments are given by von Quadt et al. (2005) using ϵ -Nd and ϵ -Hf compositions of Late Cretaceous magmatic rocks. In contrast, the Sr isotopic compositions of magmas in the northern and southern parts of the district remain surprisingly identical (Fig. 14a). However, the hydrothermal sulphates at Chelopech show higher $^{87}\text{Sr}/^{86}\text{Sr}$ ratios compared to the Radka and Elshitsa sulphates and carbonates (Fig. 13), revealing a more intense interaction of the ore-forming fluids with radiogenic metamorphic basement rocks and their detrital sedimentary products. Similarly, the more radiogenic Pb isotopic signatures of sulphides from ore deposits in the Northern Panagyurishte compared to the southern part of the district (Fig. 12c) indicate more intense and extensive basement leaching at Elatsite and Chelopech than at Elshitsa and Vlaykov Vruh.

Besides the proposed slab retreat during subduction, an additional parameter that could have influenced the differences in magma chemistry is the thickness of the crust. The crustal thickness in the Panagyurishte corridor progressively decreases from the northernmost part (Elatsite-Chelopech area) to the area of the Elshitsa magmatic complex and then increases again towards the southernmost part of the transect (Capitan Dimitriev; Fig. 1b). This variation of the crustal thickness changes in parallel with the crustal-mantle signature of the magmas, with an increasing mantle component from Elatsite southwards down to the Elshitsa magmatic complex and the southerly adjacent Boshulia pluton both showing the strongest mantellic signatures (Peytcheva et al. 2008 and this study) and then again progressively decreasing southwards to the Capitan Dimitriev pluton (Kamenov et al. 2003b). The deeper erosional level in the southernmost part of the district, lacking economic mineralisation for this reason, has probably additionally reduced the crustal thickness in the area.

Conclusions

The studied Vlaykov Vruh and Elshitsa deposits are the best example of tight spatial association of porphyry Cu and high-sulphidation epithermal Cu-Au systems within the richly mineralised Late Cretaceous Panagyurishte ore district, Central Srednogorie. The southern part of the Panagyurishte magmato-tectonic corridor, where the studied deposits are located, displays some geochemical and metallogenic particularities compared to the central and northern parts, hosting the largest mineralised systems.

The medium- to high-K calc-alkaline magmas of the Elshitsa magmatic complex, with a well pronounced mantle geochemical signature, are the most probable sources for metals and mineralising fluids involved in the formation of Vlaykov Vruh porphyry Cu and Elshitsa epithermal Cu-Au deposits. Magma generation and evolution resulted from various processes such as fractional crystallisation, crustal assimilation and contamination, as well as magma mixing at different sites during the ascent of mantle-derived magmas across the crust. Strontium isotope evidence suggests large-scale interaction between mineralising fluids and basement lithologies at Vlaykov Vruh-Elshitsa. Lead isotope compositions of hydrothermal sulphides allow discrimination of the porphyry and epithermal deposits in the Southern Panagyurishte district as a product of one single metallogenic event with a common source of metals. All field evidence and prevailing mineralogical and geochemical data indicate an epigenetic epithermal character of the mineralisation in the different ore bodies at Elshitsa, excluding the scenario of two-stage massive sulphide-epithermal ore formation proposed in the past. At Elshitsa, an early-stage high-sulphidation mineral assemblage was overprinted by intermediate-sulphidation paragenesis forming the main economic parts of the ore bodies.

Based on their common geological and structural setting, the ore-mineralogy and paragenesis, as well as their alteration and geochemical zonation, the Elshitsa and Vlaykov Vruh deposits are interpreted as the deep part of a high-sulphidation epithermal system and its spatially and genetically related porphyry Cu counterpart, respectively.

Acknowledgements

This study was supported by the SCOPES Joint Research Project 7BUPJ62276.00/1, Swiss National Science Foundation grants 21-59041.99 and 200020-101853 and fellowship from the University of Geneva and grant for visit from the GEODE programme for KK. R. Petrunov, P. Ignatovski, L. Naftali, I. Velinov, I. Chambefort, P. Marchev, S. Stoykov, S. Strashimirov, V. Jeleu and M. Ovtcharova are gratefully acknowledged for discussions. The authors would like to thank F. Capponi (University of Geneva) – for XRF analyses, M. Falchéri (University of Geneva) – for

helping with the radiogenic isotope geochemistry, and P. Benoist-Julliot (ISTO-CNRS, Orléans) – for atomic absorption analyses of Sr. The journal reviewers Robert Frei and Volker Lüders are thanked for constructive comments, which improved the manuscript and Bernd Lehmann is thanked for efficient editorial handling. This is a contribution to the ABCD-GEODE research program supported by the European Science Foundation.

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Figure captions:

Figure 1 – **a)** Position of the Srednogorie zone within its Alpine tectonic framework; **b)** Tectonic units in Bulgaria. The Srednogorie zone is constituted of fragments of a Late Cretaceous island arc system. Crustal thickness in km is indicated as well (modified after Yossifov and Pčelarov 1977; Dabovski et al. 1991; Kamenov et al. 2003a); **c)** Simplified geological map of the Panagyurishte ore district (modified after Popov and Popov 2000). Location of major porphyry Cu-Au and epithermal Cu and Cu-Au deposits and occurrences is shown.

Figure 2 – **a)** Geological map of the southern part of the Panagyurishte district (modified after Ignatovski et al. 1990); **b)** Geological map of the area of Vlaykov Vruh and Elshitsa deposits (modified after Bogdanov et al. 1970b) with location of the samples used for U-Pb dating.

Figure 3 – View looking southwest toward the Vlaykov Vruh and Elshitsa deposits.

Figure 4 – **a)** Cross-section of the Vlaykov Vruh porphyry Cu deposit (after Strashimirov et al. 2003). Open pit outline as of 1979; **b)** Cross-section of the Elshitsa epithermal Cu-Au deposit (after Chipchakova and Stefanov 1974).

Figure 5 – Ore textures and mineralogy of the Vlaykov Vruh porphyry Cu deposit: **a)** Propylitically altered Vlaykov Vruh granodiorite; **b)** Aplite-pegmatite texture of felsic dyke cutting the granodiorite porphyry; **c)** K-silicate alteration assemblage affecting the host gneisses: shreddy biotite replacing the mafic rocks and K-feldspar-magnetite veinlets cut by later sulfide (chalcopyrite-pyrite) veinlets; **d)** Quartz-molybdenite vein with later pyrite crystals in vugs, cutting aplitic dyke; **e)** Late pyrite-chlorite vein along the contact between subvolcanic dacitic body and the host gneisses; **f)** Hydrothermal breccia cutting subvolcanic dacite. The breccia matrix consists of quartz, pyrite and minor chlorite and sericite; **g)** Hydrothermal rutile from quartz-rutile-molybdenite vein, BSE image; **h)** Chalcopyrite-magnetite-pyrrhotite-quartz (\pm pyrite) association, typical for the main economic mineralisation stage at Vlaykov Vruh, BSE image; **i)** Pyrite-chalcopyrite-chlorite-muscovite-quartz association from the main economic stage, BSE image. *Abbreviations:* ab – albite; bt – biotite; chl – chlorite; cp – chalcopyrite; hbl – hornblende; K-fs – K-feldspar; mol – molybdenite; ms – muscovite; mt – magnetite; plag – plagioclase; po – pyrrhotite; py – pyrite; qtz – quartz; rt – rutile; ttn – titanite.

Figure 6 – Paragenetic sequence of the Vlaykov Vruh porphyry Cu deposit (based on Bogdanov et al. 1972b; Popov et al. 2000; Kouzmanov 2001; and Strashimirov et al. 2003). Temperature intervals are based on: (*) Quartz-magnetite O-isotope thermometry (Kouzmanov et al. 2003); and (**) fluid inclusion microthermometry (Popov et al. 2000; Kouzmanov 2001). *Abbreviations:* Bt – biotite; Carb – carbonates; Cp – chalcopyrite; Gn – galena; Kfs – K-feldspar; Mol – molybdenite; Mt – magnetite; Py – pyrite; Qtz – quartz; Rt – rutile; Sp – sphalerite; Zeol - zeolites.

Figure 7 – Gold, silver and copper ore grades of the Elshitsa epithermal Cu-Au deposit. Data are representative for massive and disseminated pyrite and copper ore bodies from level 80 to level 320: **a)** Au (ppm) versus Ag (ppm); **b)** Au (ppm) versus Cu (wt %); **c)** Au (ppm) versus S (wt %). The Au, Ag and Cu average ore grades for the deposit are indicated on a) and b).

Figure 8 – Ore textures and mineralogy of the Elshitsa epithermal Cu-Au deposit. Microphotographs are taken in reflected light (/N) except p) which is taken in transmitted light (xN): **a)** Tectonic contact of massive copper-pyrite ore body with subvolcanic rhyodacite (level 80; ore body n°1); **b)** Boxwork texture typical for the small gossan body developed in the SE part of the Elshitsa deposit; **c)** Ore clasts of stage-I massive fine-grained pyrite in intramineralisation breccia-dyke with fine-grained matrix and clasts of volcanic rhyodacites, tuffs and subvolcanic dacites; **d)** Stage-I fine-grained pyrite replacing laminated dacitic ash tuff; **e)** Breccia-dyke with hydrothermal matrix (quartz and pyrite) and sub-rounded dacitic clasts; **f)** Vein of stage-I pyrite, cut by polymetallic sphalerite-galena vein, with later anhydrite infill; **g)** Late anhydrite veins cutting massive pyrite-polymetallic ore; **h)** Massive fine-grained and colloform pyrite; **i)** Fine-grained pyrite from the massive pyrite bodies, in association with enargite and goldfieldite; **j)** Massive coarse-grained As-bearing pyrite. Darker zones correspond to higher As contents. Microphotograph is taken after etching of the polished section with HNO_{3conc}; **k)** Stage-II chalcopyrite replacing the stage-I colloform pyrite; **l)** Colloform structures in stage-I massive pyrite, selectively replaced by chalcopyrite and tennantite from stage-II; **m)** Massive chalcopyrite with inclusions of native gold, galena and aikinite. Quartz and sericite constitute the alteration assemblage; **n)** Fine-grained polymetallic ore from the ore stage-III: sphalerite, galena, chalcopyrite and tetrahedrite in various proportions are the dominant minerals; **o)** Native gold in association with the main minerals of the polymetallic stage; **p)** Supergene gypsum replacing hypogene anhydrite along fractures and cleavage planes.

Abbreviations: aik – aikinite; anh – anhydrite; Au – gold; cp – chalcopyrite; en – enargite; gdf – goldfieldite; gn – galena; gy – gypsum; py – pyrite; qtz – quartz; ser – sericite; sp – sphalerite; td – tetrahedrite; tn – tennantite.

Figure 9 – Paragenetic sequence of the Elshitsa epithermal Cu-Au deposit (based on Popov et al. (2000) and Kouzmanov (2001) and references therein). Temperature intervals are according to Strashimirov and Kovachev (1992) and were determined by fluid inclusion microthermometry. Salinities of the mineralising fluids are from Kouzmanov (2001).

Figure 10 – Major and trace element composition of unaltered Late Cretaceous volcanic and intrusive rocks from the southern part of the Panagyurishte district. Data from this study are presented in colour and published data are presented as open dots: **a)** (Na₂O + K₂O) versus SiO₂ (TAS diagram; division after Le Maitre et al. 1989). Data from the Appendix are used after normalisation to 100% of the 10 major oxides. The grey field shows the general trend for the Southern Panagyurishte magmas; **b)** K₂O versus SiO₂ diagram (division after Le Maitre et al. 1989). Most of the igneous rocks from the southern part of the district plot within the medium-K and high-K fields; **c)** AFM diagram (division after Kuno 1968); **d)** Trace element concentrations normalised to the primitive mantle composition (after Sun and McDonough 1989); **e)** Rare earth element abundances normalised to chondritic meteorite values (after McDonough and Sun 1995).

Figure 11 – Variation of wt % TiO₂, FeO*, MgO, CaO, P₂O₅ and Al₂O₃, and ppm Sc, Sr, Rb, Zr, La and U versus wt % SiO₂ for Late Cretaceous volcanic and intrusive rocks from the Elshitsa magmatic complex. Dashed lines show trends typical for fractional crystallisation of magnetite, amphibole, pyroxene, zircon and apatite. Symbols – as in Figure 10.

Figure 12 – **a)** ²⁰⁸Pb/²⁰⁴Pb versus ²⁰⁶Pb/²⁰⁴Pb and **b)** ²⁰⁷Pb/²⁰⁴Pb versus ²⁰⁶Pb/²⁰⁴Pb plots of whole rocks from the Southern Panagyurishte district. Data for the Variscan Vejen granite, K-feldspars from pegmatites in other Variscan intrusions from Central Srednogorie, gneisses from Central Srednogorie and ores from Vlaykov Vruh, Elshitsa and Radka deposits are presented for comparison. All Pb isotope data have been age corrected (85 Ma; von Quadt et al. 2005 – for 90 Ma), except the Pb data of Amov et al. (1973, 1982); **c)** ²⁰⁸Pb/²⁰⁴Pb versus ²⁰⁶Pb/²⁰⁴Pb and **d)** ²⁰⁷Pb/²⁰⁴Pb versus ²⁰⁶Pb/²⁰⁴Pb plots of sulphide ore minerals from the Panagyurishte district. The evolution curves of the mantle (M), upper crust (UC) and orogen (OR) are from Zartmann and Doe (1981).

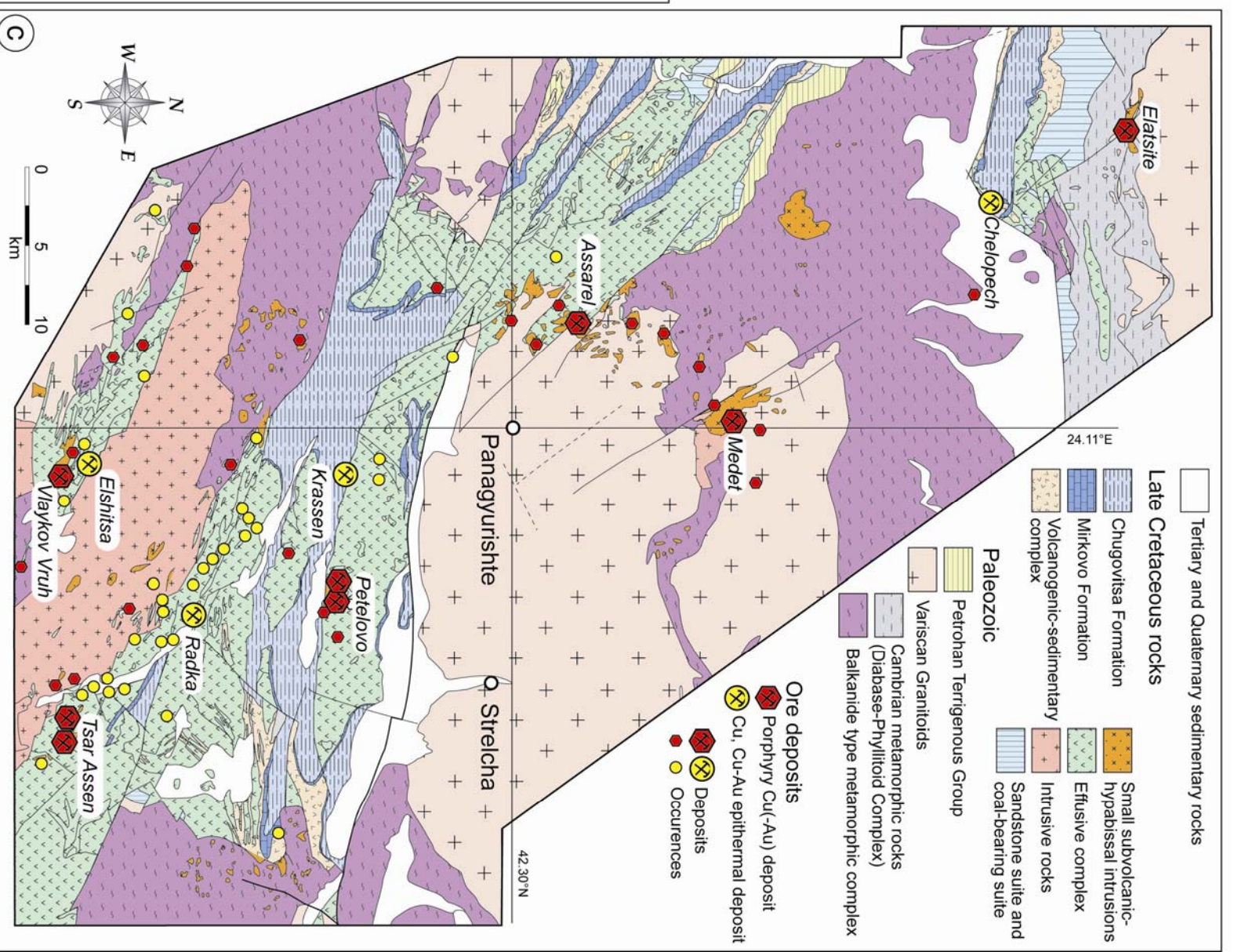
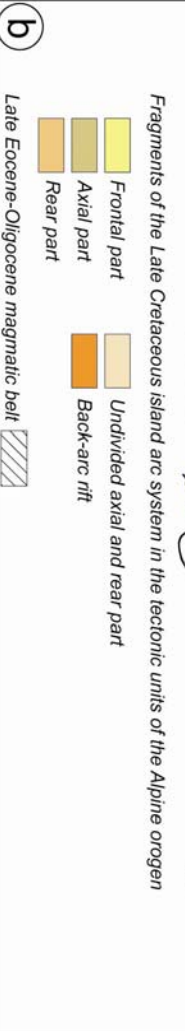
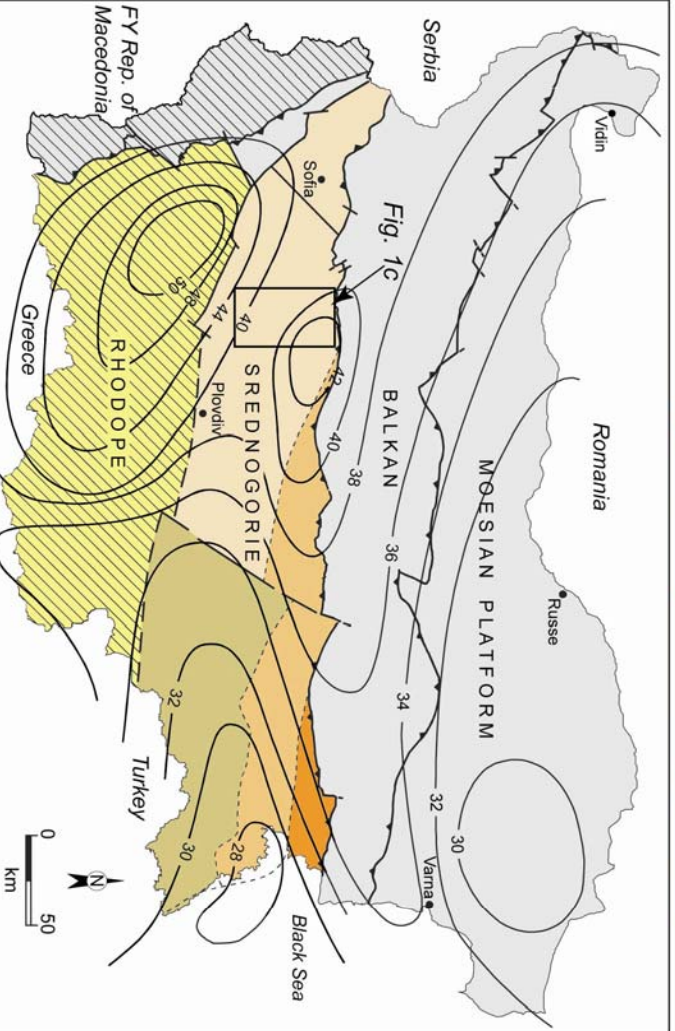
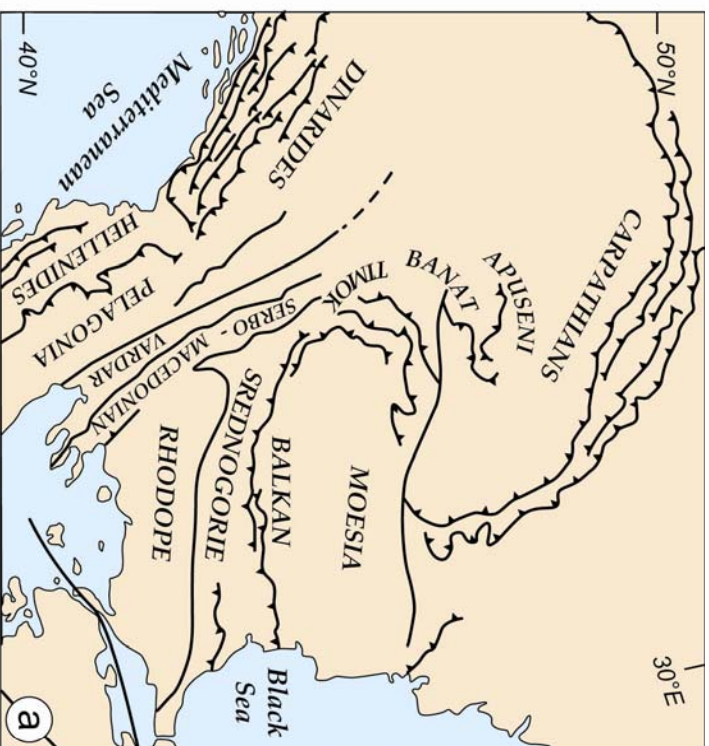
Figure 13 – Strontium isotope data for host rocks and hydrothermal minerals in southern and northern parts of the Panagyurishte district. Data from this study are presented in colour and published data are presented as light-grey symbols. Late Cretaceous igneous rocks are marked with

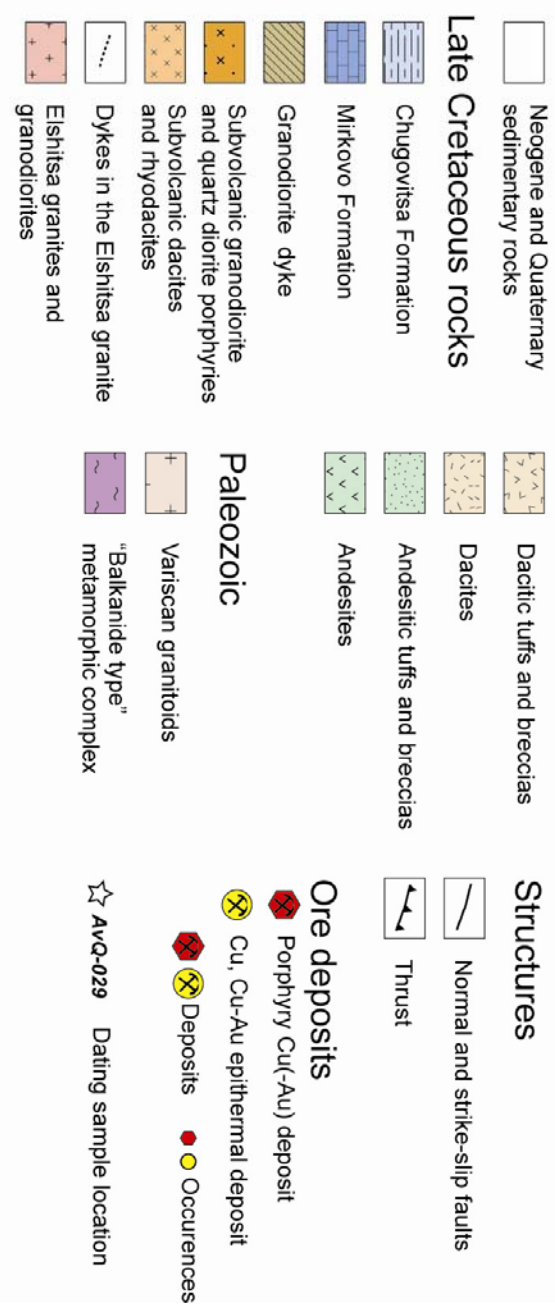
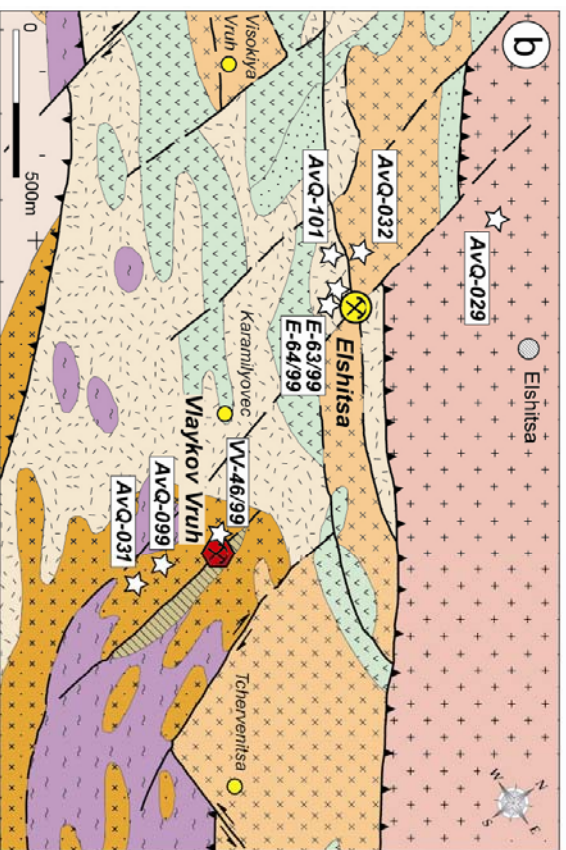
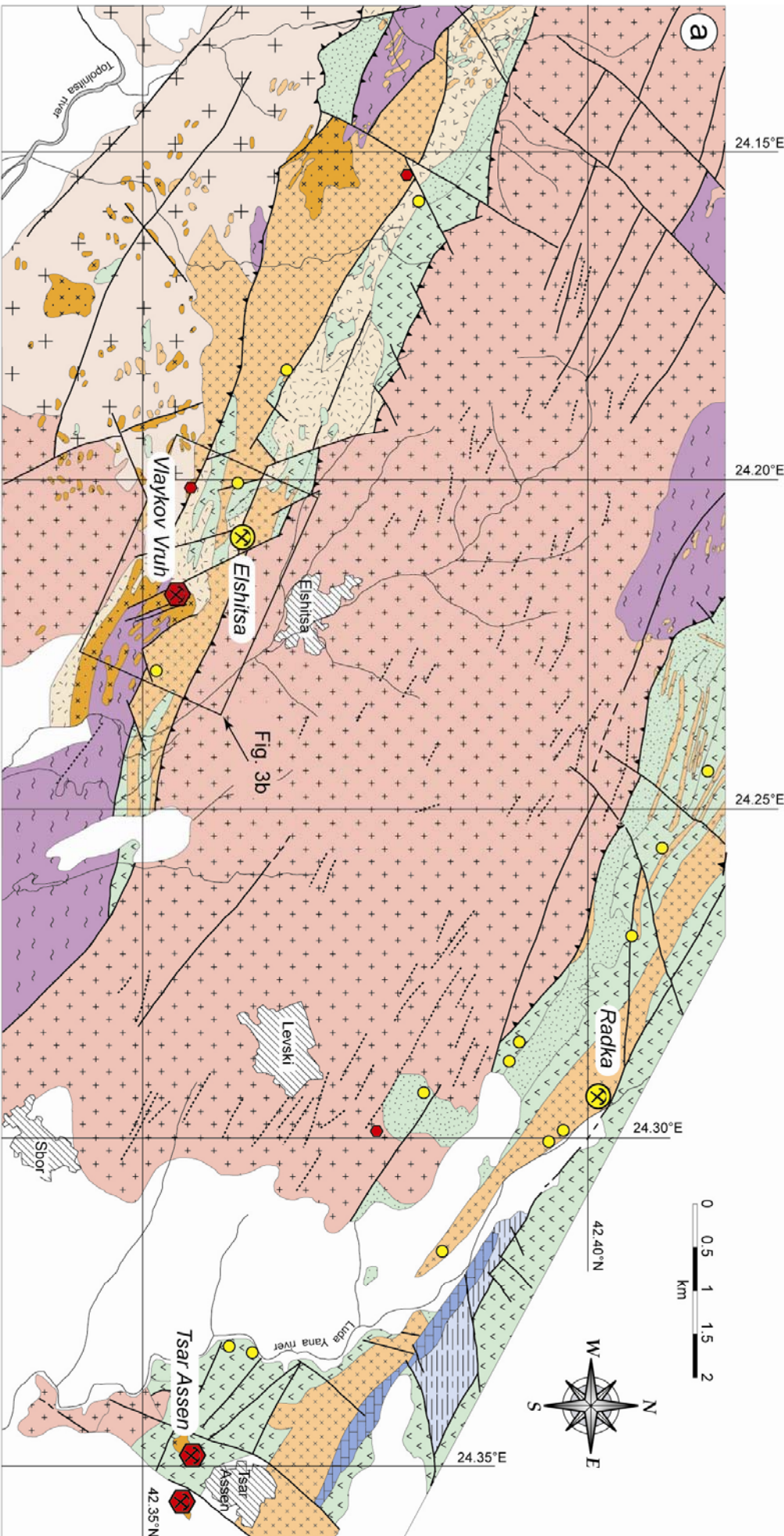
asterisk. The sources for the published data are: Chelopech hydrothermal minerals (Moritz et al. 2001, 2003; Jacquat 2003); Chelopech igneous rocks (Stoykov et al. 2003, 2004); Elatsite hydrothermal minerals and igneous rocks (von Quadt et al. 2002, 2005); Vejen and Koprivshtitsa Variscan intrusions in Northern Panagyurishte (Kamenov et al. 2002; von Quadt et al. 2002, 2005; Jacquat 2003); metamorphic basement in Northern Panagyurishte (Jacquat 2003); Southern Panagyurishte Late Cretaceous igneous rocks from various magmatic centres (von Quadt et al. 2005); Capitan Dimitriev pluton in the southernmost part of the district (Kamenov et al. 2003b; von Quadt et al. 2005); Late Cretaceous seawater (Koepnick et al. 1985); few data for the Elshitsa granite, subvolcanic rhyodacites and the Lesitchevo Variscan intrusion from Southern Panagyurishte (Peytcheva et al. 2003) are plotted as well. Data for the Northern Panagyurishte are corrected for 90 Ma, and data for the Southern Panagyurishte – for 85 Ma.

Figure 14 – **a)** SiO₂ versus (⁸⁷Sr/⁸⁶Sr)_i, **b)** SiO₂ versus (¹⁴³Nd/¹⁴⁴Nd)_i and **c)** (⁸⁷Sr/⁸⁶Sr)_i versus (¹⁴³Nd/¹⁴⁴Nd)_i diagrams for rocks from the Southern Panagyurishte district. Fields of the EM1 and 2 (enriched mantle 1 and 2) and BSE (bulk silicate earth) are given (corrected for 85 Ma) according to Zindler and Hart (1986) and Hart and Zindler (1989). Dashed lines correspond to the Sr and Nd isotope values of the undifferentiated reservoir, corrected for 85 Ma.

Figure 15 – Strontium isotope data for hydrothermal anhydrite from Elshitsa and Radka epithermal deposits – **a)** ⁸⁷Sr/⁸⁶Sr versus 1/Sr (ppm)*10⁵ diagram. The observed negative correlation is interpreted in terms of fluid mixing between two hypothetical end-member fluids – A and B (for discussion see text); **b)** Oxygen and strontium isotopic compositions of the hydrothermal fluid that precipitated anhydrite at Elshitsa and Radka. Oxygen isotopic composition of primary magmatic water (Taylor 1974) and Late Cretaceous local meteoric water (Kouzmanov et al. 2003), as well as strontium and oxygen isotopic compositions of Late Cretaceous seawater (Koepnick et al. 1985; Claypool et al. 1980) are indicated for comparison.

Figure 16 – U-Pb concordia diagrams of the Elshitsa granite AvQ 029 (**a**), the Elshitsa subvolcanic rhyodacite AvQ 032 (**b**) and the hydrothermal rutile from Vlaykov Vruh deposit VV 46/99 (**c**). Individual analyses are shown as 2σ error ellipses for zircons and 1σ ellipses for rutile. Given ages are weighted mean ²⁰⁹Pb/²³⁸U ages at 95% confidence level. Numbers correspond to the zircon numbers from Table 6. Only concordant zircons, not showing Pb loss or inheritance are presented on the plots (see text for explanation). Measured ε-Hf values for single zircons are indicated as well.





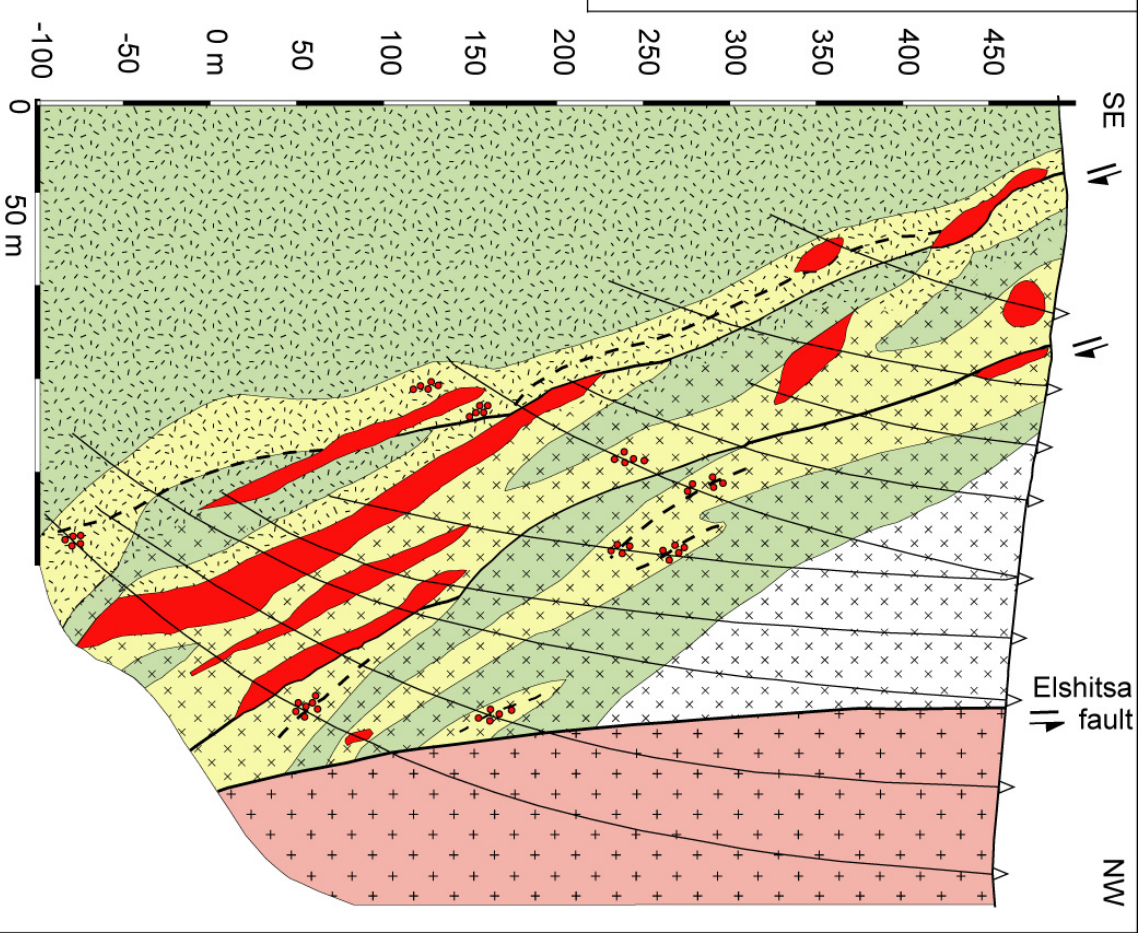
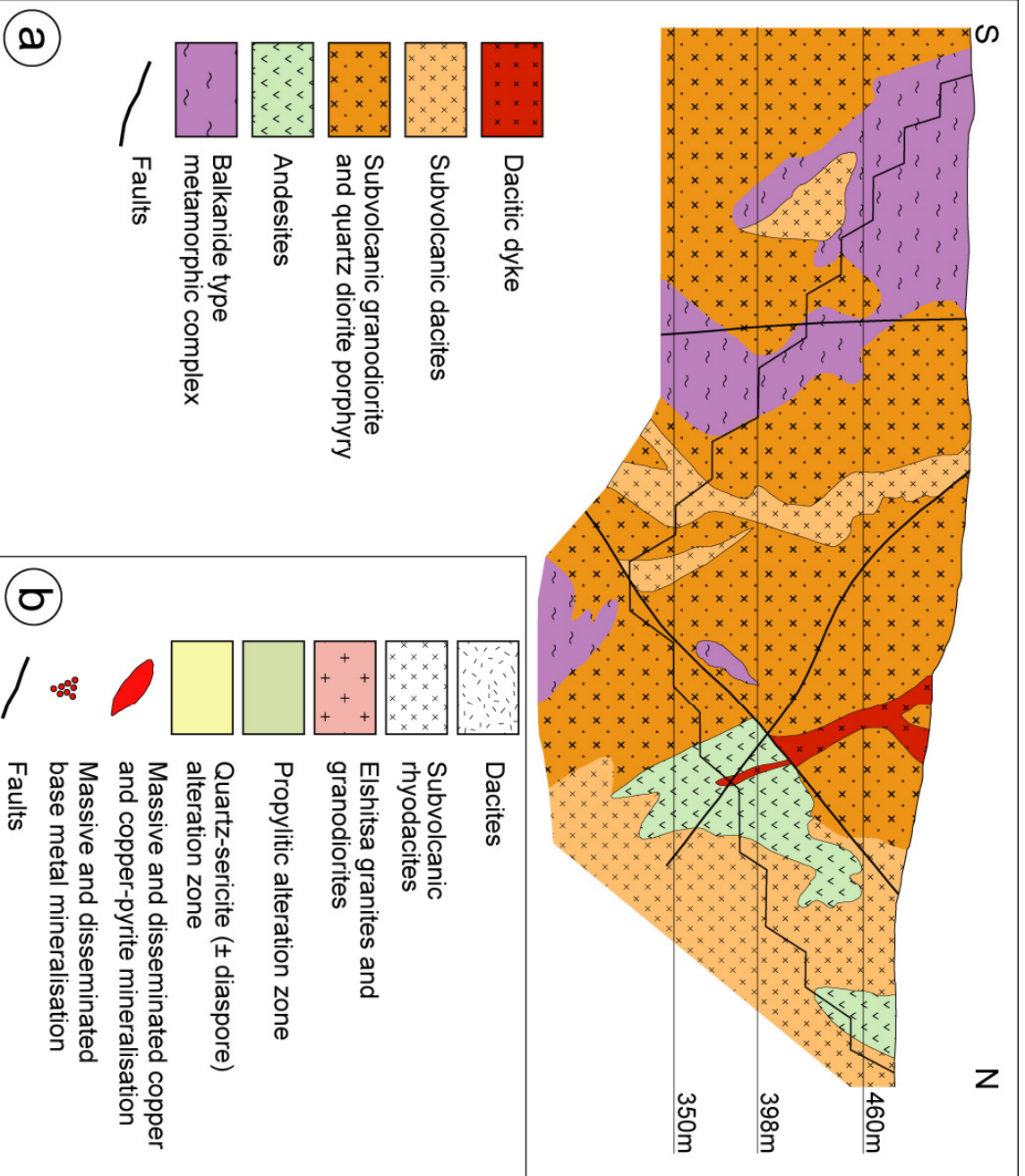
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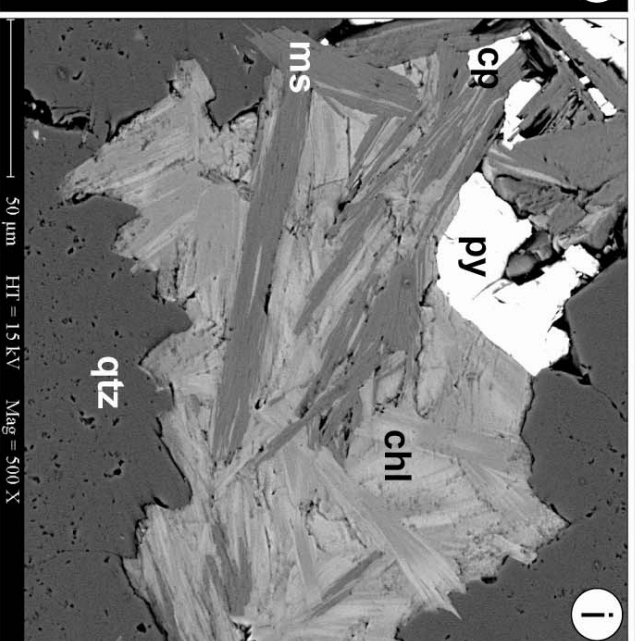
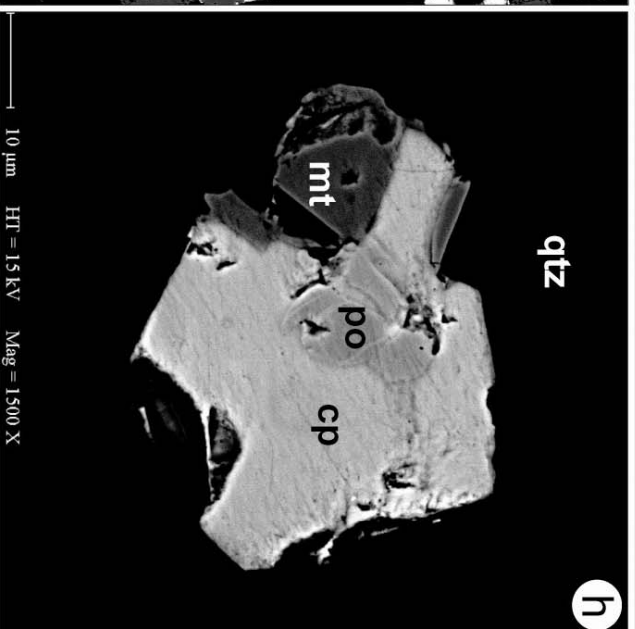
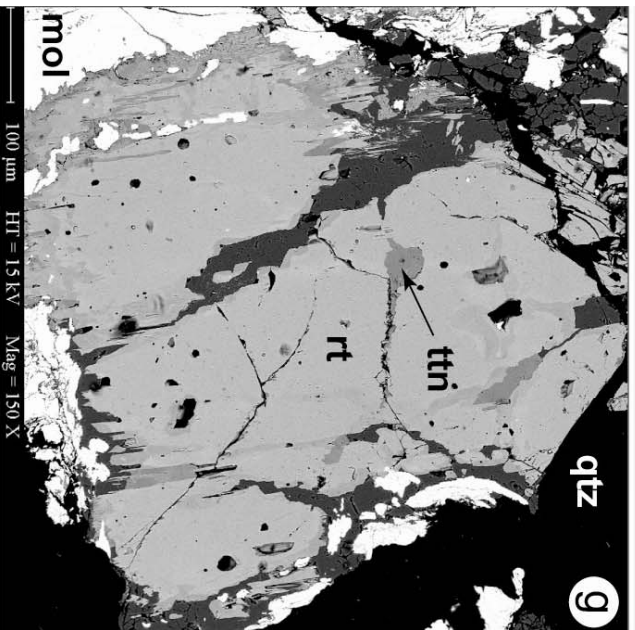
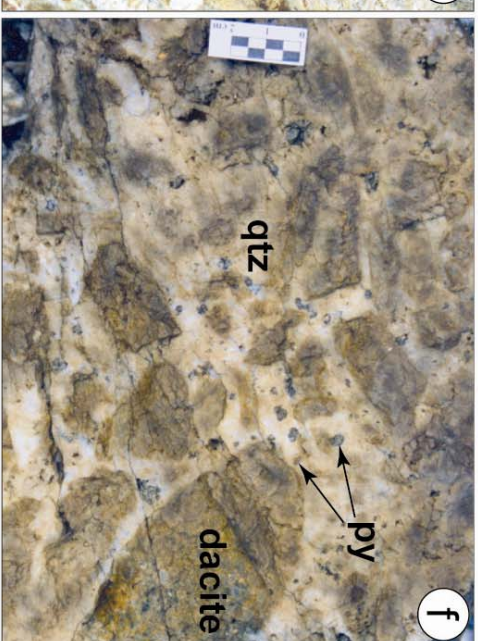
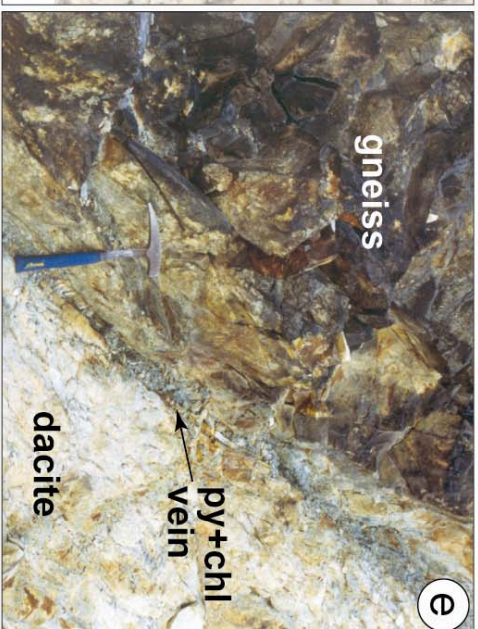
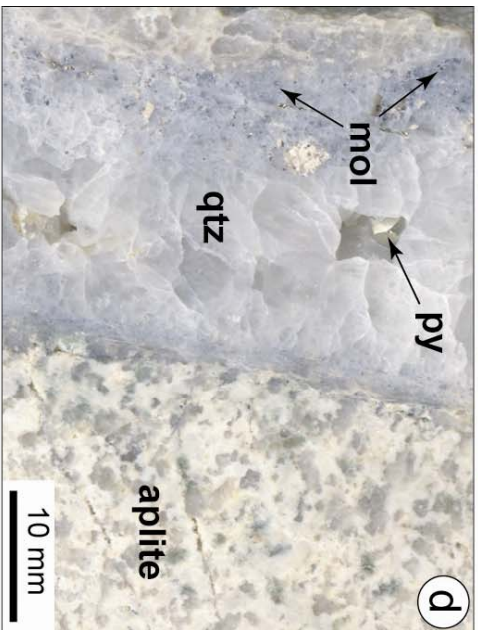
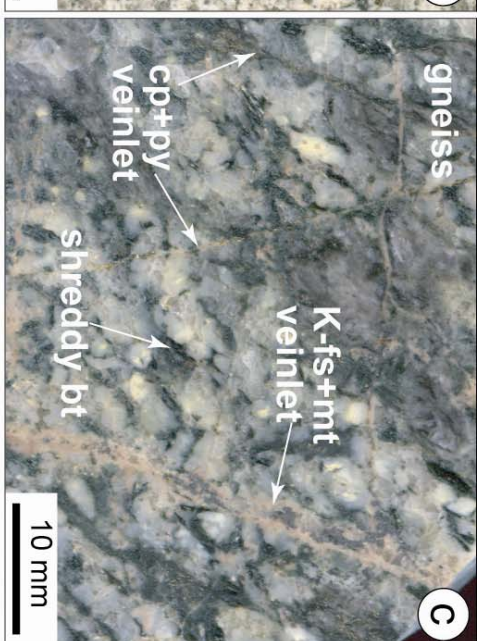
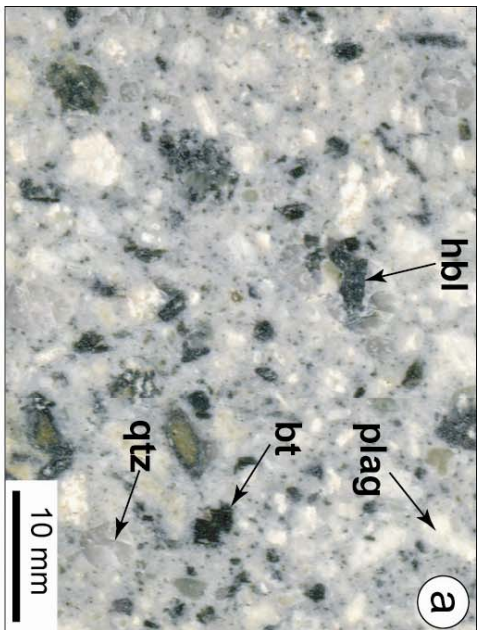
Wlyakov Vruh porphyry Cu deposit



Elshitsa epithermal Cu(-Au) deposit

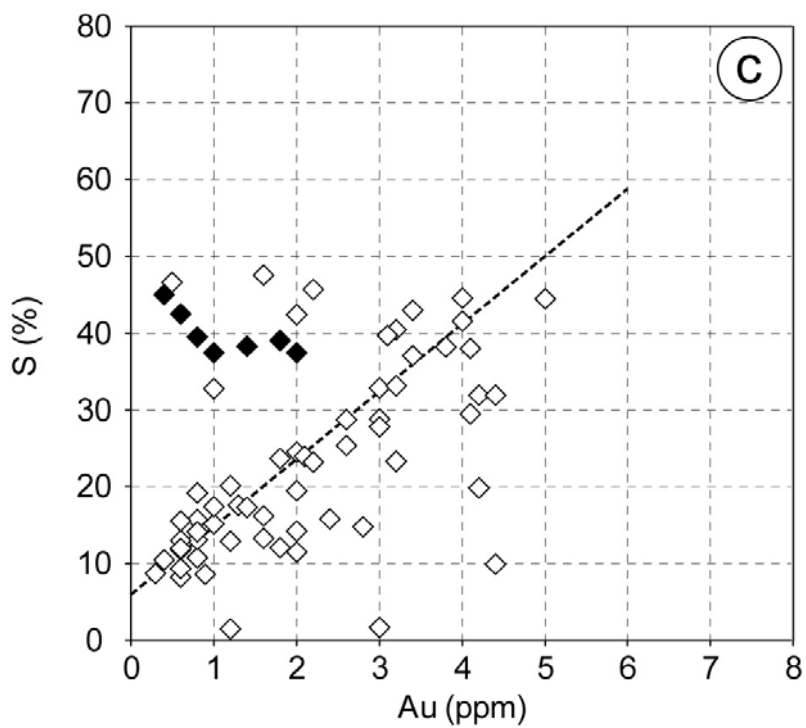
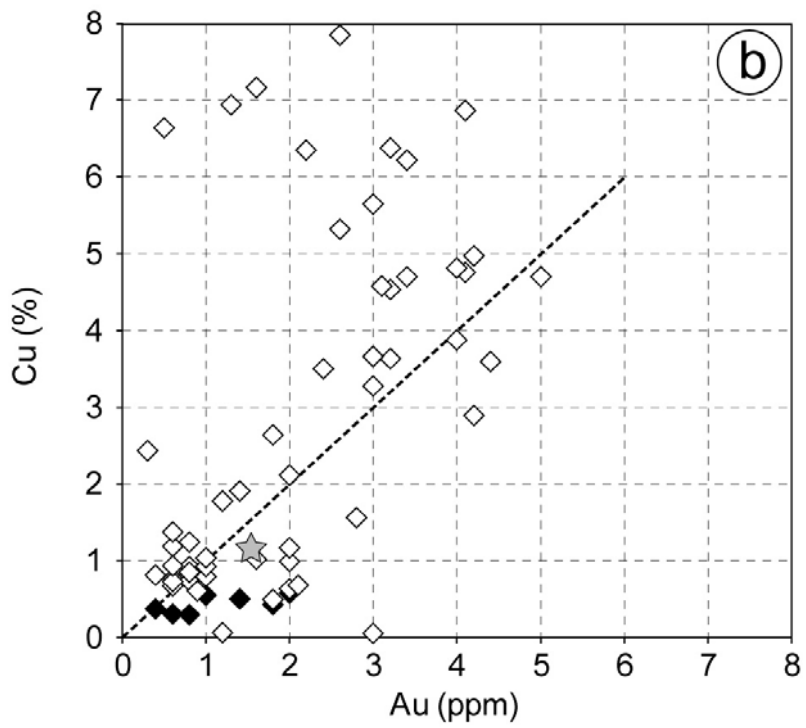
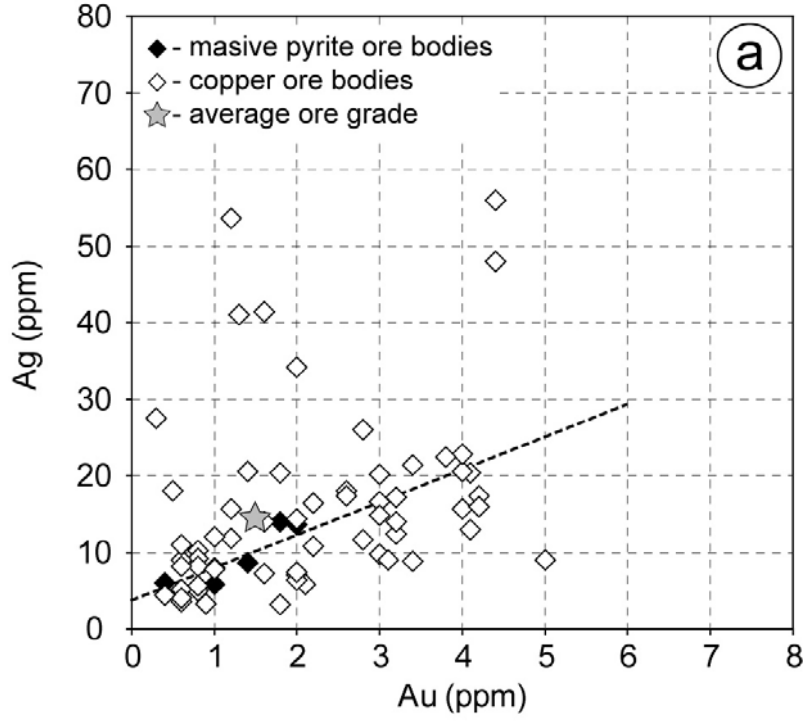
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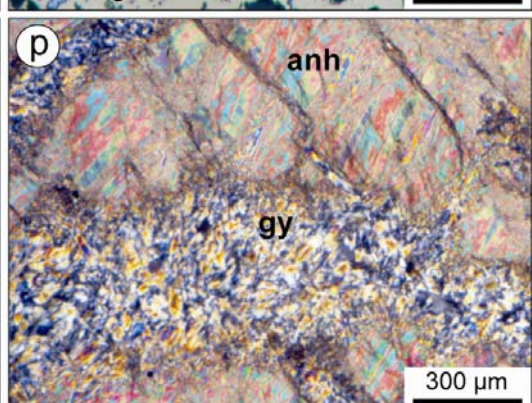
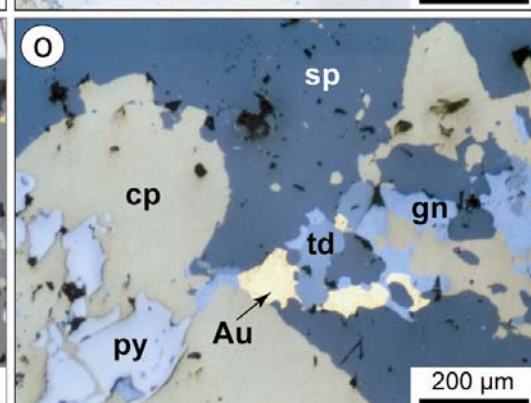
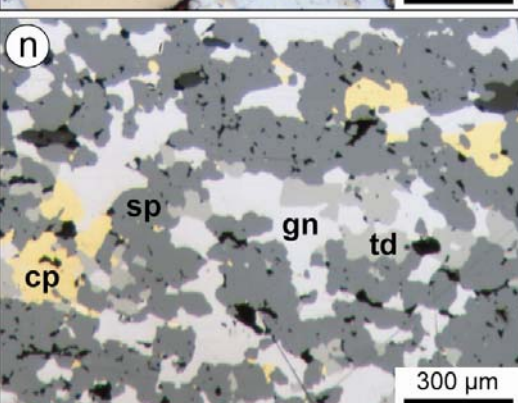
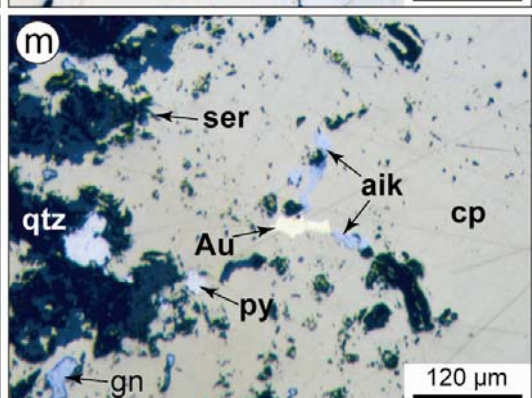
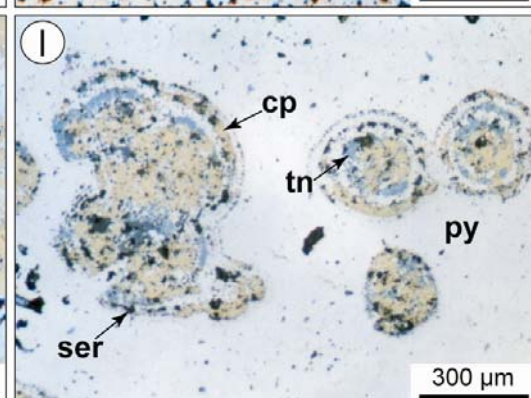
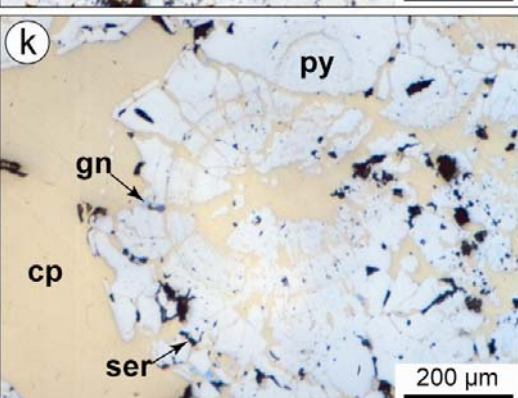
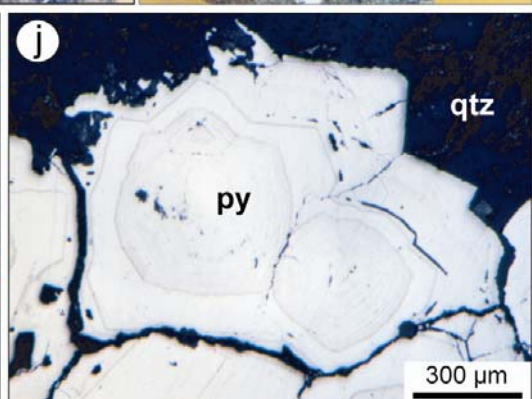
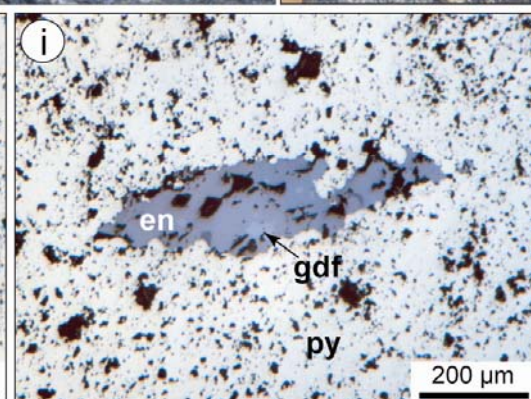
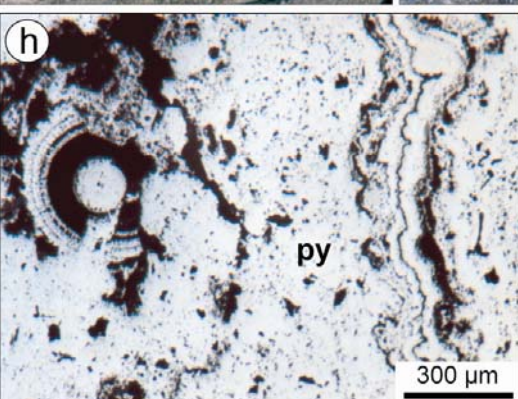
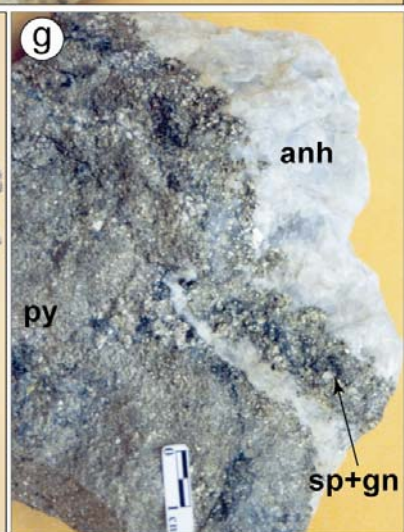
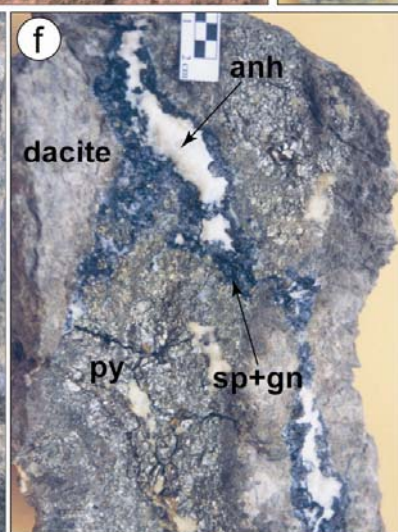
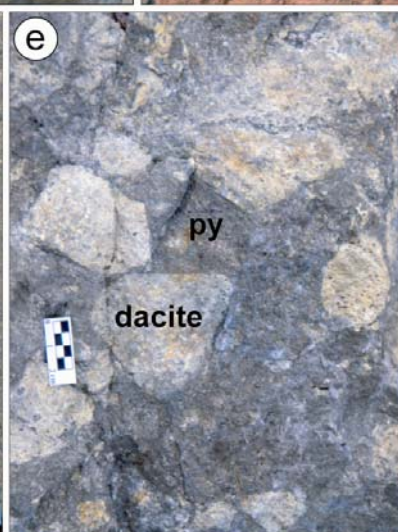
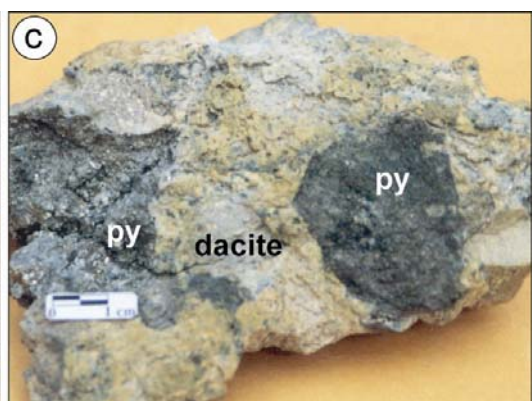
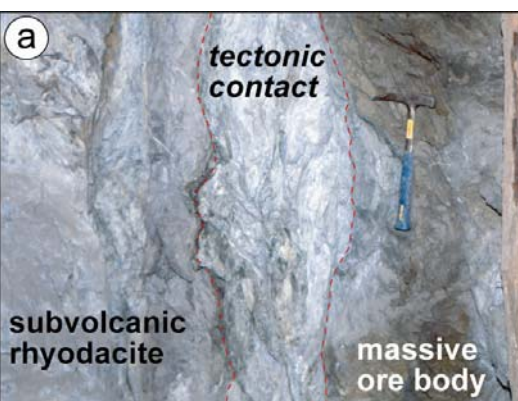




| <i>Mineralisation stages</i> | I | II | III | IV | V | VI | VII | VIII |
|------------------------------|--------------------|------------|--------------|------------------|---------------------|----------------|--------------------|-------|
| <i>Minerals</i> | Kfs-Qtz- -Mt-Bt | Qtz-Rt-Mol | Qtz-Cp-Mt-Py | Qtz-Sp -Gn-Cp | Qtz-Py- -Chl-Ser | Carb- -Zeol | Second. enrich. | Oxid. |
| Quartz | ▬ | ▬ | ▬ | ▬ | ▬ | | | |
| K-feldspar | ▬ | | | | | | | |
| Biotite | ▬ | | | | | | | |
| Magnetite | ▬ | | ▬ | | | | | |
| Rutile | | ▬ | | | | | | |
| Molybdenite | ▴ | ▬ | | | | | | |
| Titanite | | ▴ | | | | | | |
| Chlorite | | | | | | | | |
| Chalcopyrite | | ▴ | ▬ | | | | | |
| Pyrite | | ▴ | ▬ | | | | | |
| Pyrrhotite | | ▴ | ▬ | | | | | |
| Muscovite / Illite | | | ▬ | ▬ | ▬ | | | |
| Tennantite / Tetrahedrite | | | ▬ | ▬ | | | | |
| Bornite | | | ▬ | | | | ▬ | |
| Aikinite | | | ▴ | | | | | |
| Tetradymite | | | ▴ | | | | | |
| Gold | | | ▬ | | | | | |
| Anatase | | | | ▴ | ▴ | | | |
| Sphalerite | | | | ▬ | | | | |
| Galena | | | | ▬ | | | | |
| Barite | | | | ▬ | | | | |
| Hematite | | | | ▬ | | | | |
| Calcite | | | | ▬ | | ▬ | | |
| Zeolites | | | | | | ▬ | | |
| Chalcocite | | | | | | | ▬ | |
| Covellite | | | | | | | ▬ | |
| Fe-oxides & hydroxides | | | | | | | | ▬ |
| Azurite | | | | | | | | ▬ |
| Malachite | | | | | | | | ▬ |
| Chrysocolla | | | | | | | | ▬ |
| Native Cu | | | | | | | | ▴ |
| Tenorite | | | | | | | | ▬ |
| Melanterite | | | | | | | | ▬ |
| Temperature (°C) | 600-500°C* | | 390-270°C** | | <200°C** | | | |

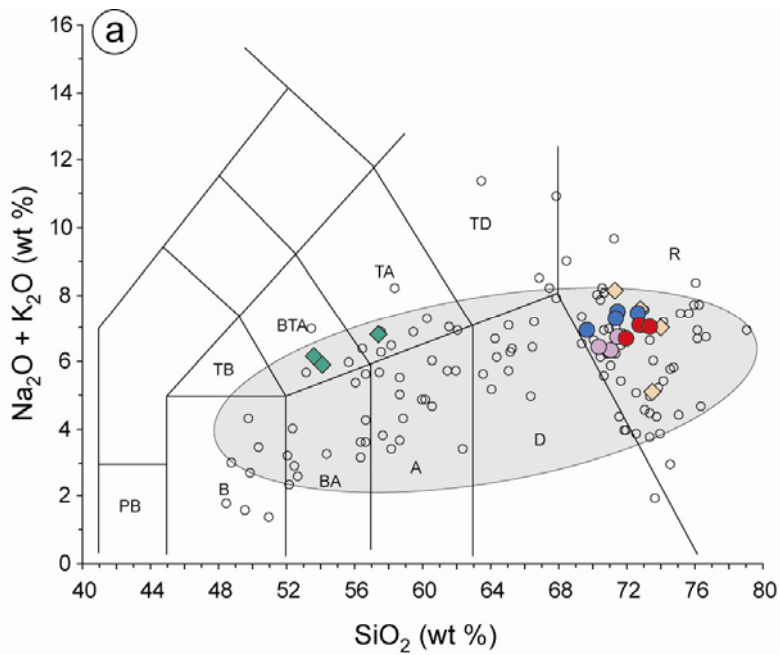
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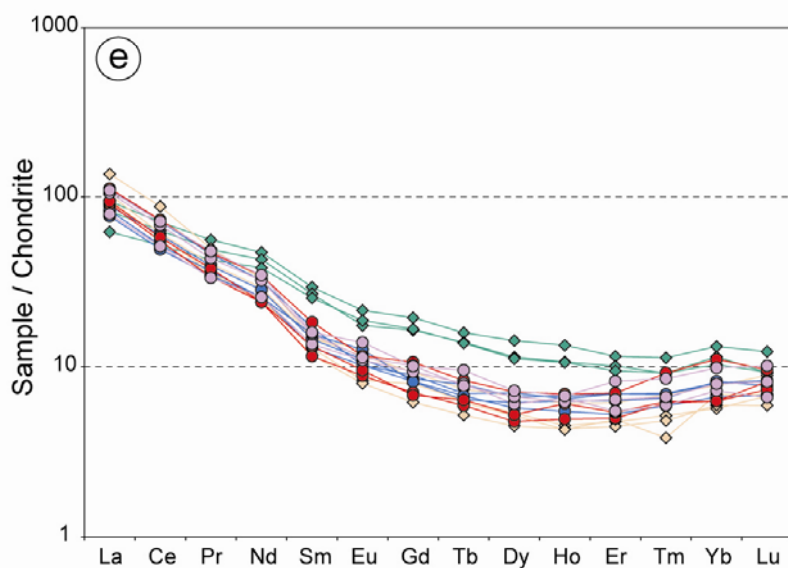
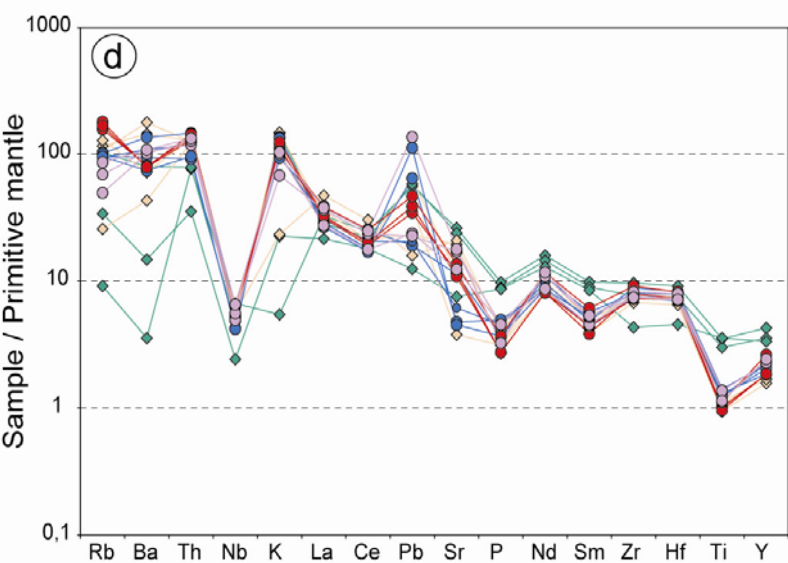
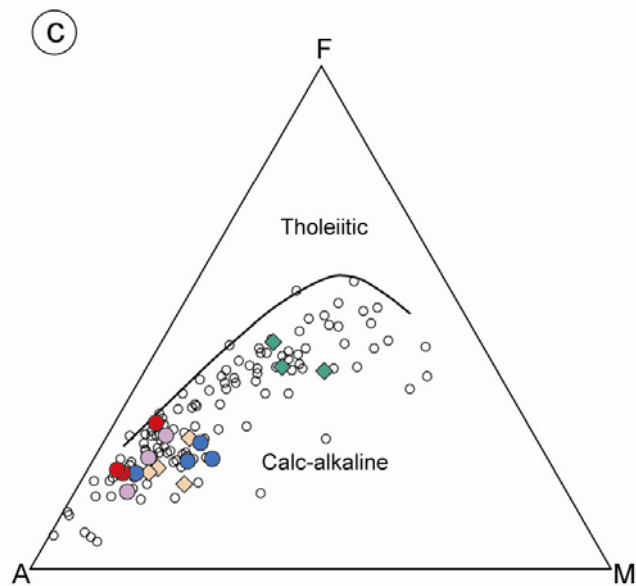
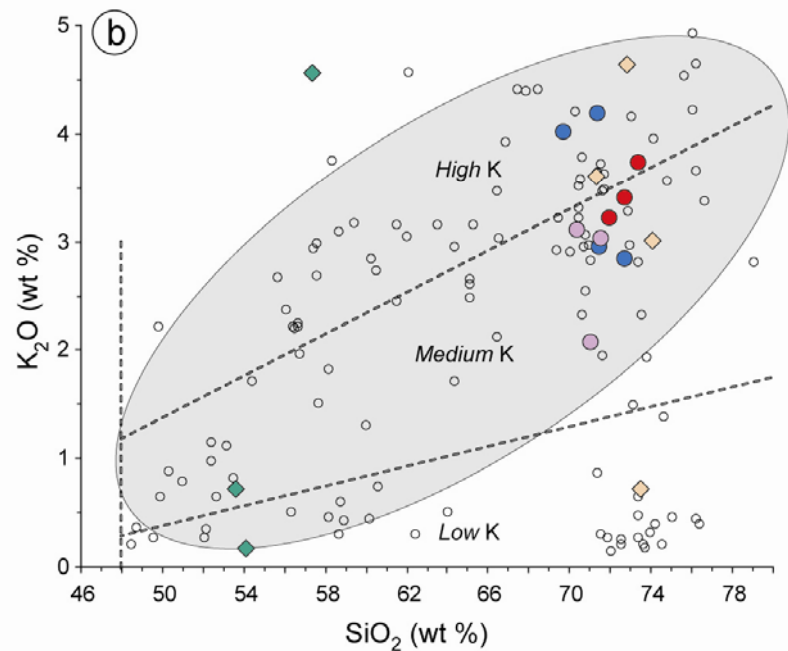


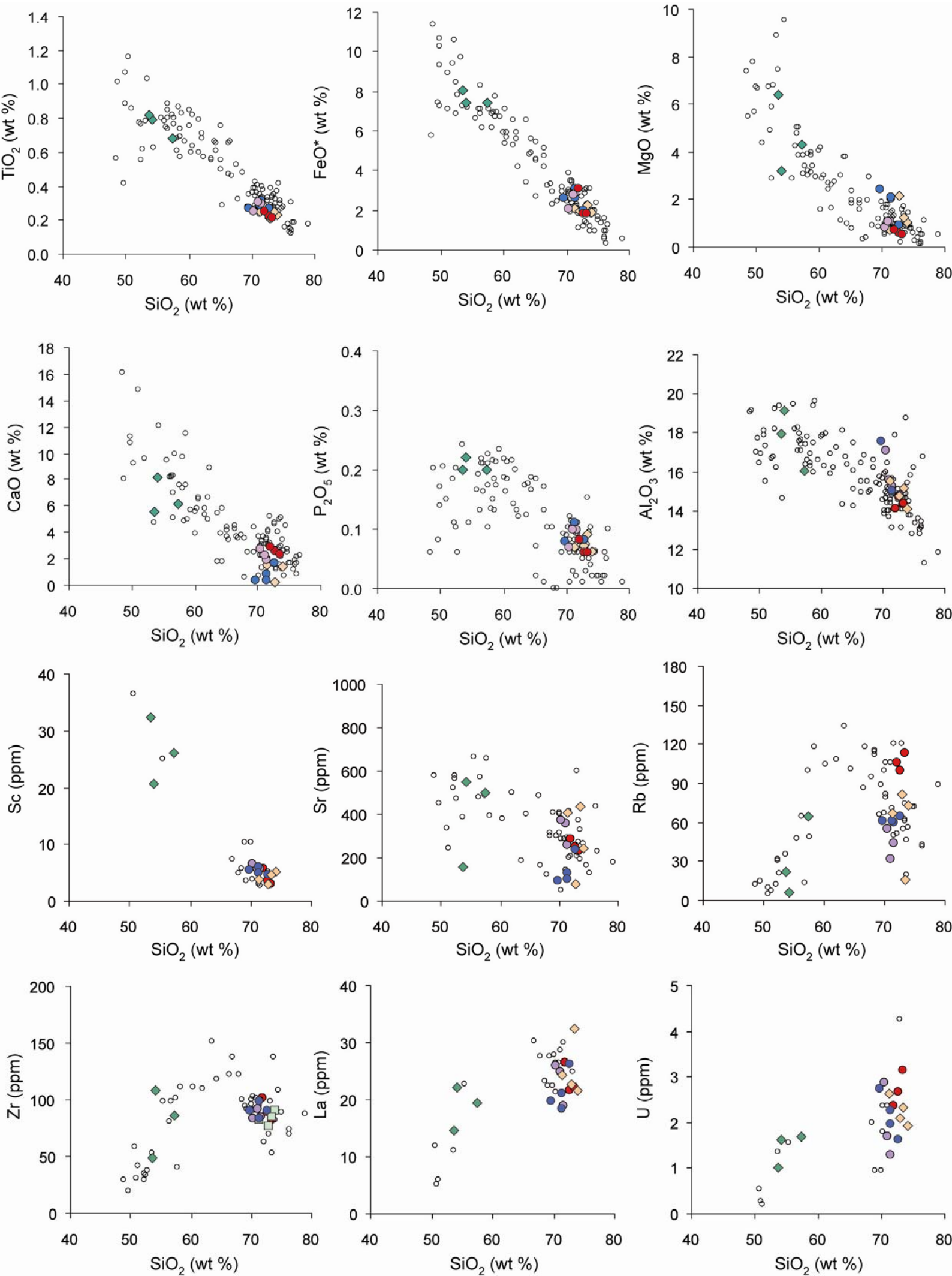
| <i>Mineralisation stages</i> | I | II | III | IV | V | VI |
|--------------------------------|----------------|----------------------|---------------------------|-------|-------------|-------------------|
| <i>Minerals</i> | Fe (±As,Cu) | Cu (±Bi,Te,Pb,As) | Zn-Pb-Cu (±Sb, As, Au) | Si-Fe | Ca (±Fe) | Remob. & Oxid. |
| Quartz | | | | | | |
| Illite | | | | | | |
| Alunite | | | | | | |
| Diaspore | | | | | | |
| Dumortierite | | | | | | |
| Rutile | | | | | | |
| Anatase | | | | | | |
| Cassiterite | | | | | | |
| Pyrite | | | | | | |
| Enargite | | | | | | |
| Goldfieldite | | | | | | |
| Covellite | | | | | | |
| Chalcopyrite | | | | | | |
| Tennantite | | | | | | |
| Sphalerite | | | | | | |
| Galena | | | | | | |
| Marcasite | | | | | | |
| Bornite | | | | | | |
| Aikinite-Bismuthinite | | | | | | |
| Tetradymite | | | | | | |
| Various Cu-Bi-Au-Ag tellurides | | | | | | |
| Native Bi | | | | | | |
| Stannoidite | | | | | | |
| Gold | | | | | | |
| Tetrahedrite | | | | | | |
| Arsenosulvanite | | | | | | |
| Sulvanite | | | | | | |
| Betekhtinite | | | | | | |
| Barite | | | | | | |
| Native Ag | | | | | | |
| Chalcocite | | | | | | |
| Calcite | | | | | | |
| Anhydrite | | | | | | |
| Fluorite | | | | | | |
| Gypsum | | | | | | |
| Fe-oxides & hydroxides | | | | | | |
| Jarosite | | | | | | |
| Melanterite | | | | | | |
| Temperature (°C) | 400-390°C | 240-220°C | 220-210°C | | | |
| Salinity (wt% NaCl equiv) | | 4.6-11.5 | 1.0-5.1 | | | |

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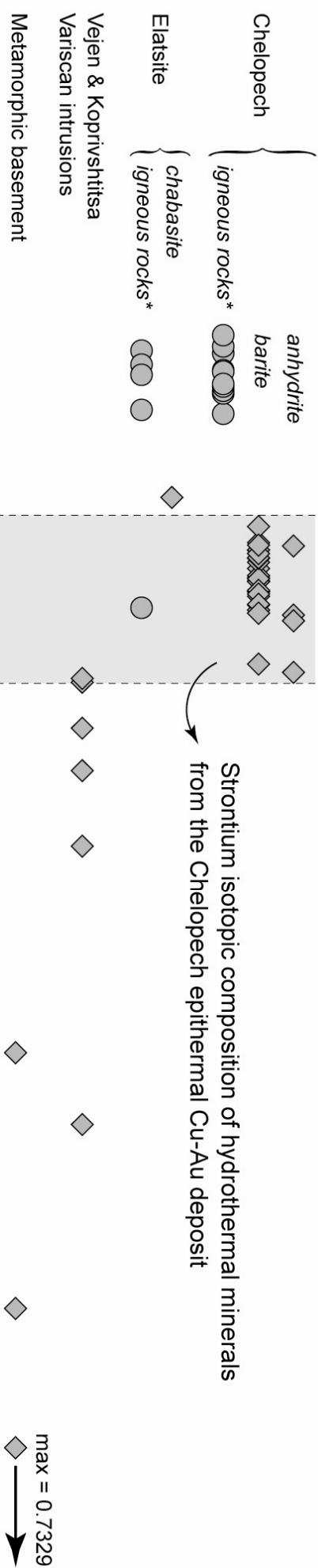


- ◆ - Andesitic volcanic rocks
- ◇ - Dacitic and rhyodacitic volcanic rocks
- - Subvolcanic rhyodacites
- - Elshitsa granite
- - Vlaykov Vruh granodiorite
- - Late Cretaceous magmatic rocks from the southern part of the Panagyurishte district (after Bojadjiev & Chipchakova, 1962; Bojadjiev & Chipchakova, 1965; Ignatovski et al., 1990; Daieva & Chipchakova, 1997)

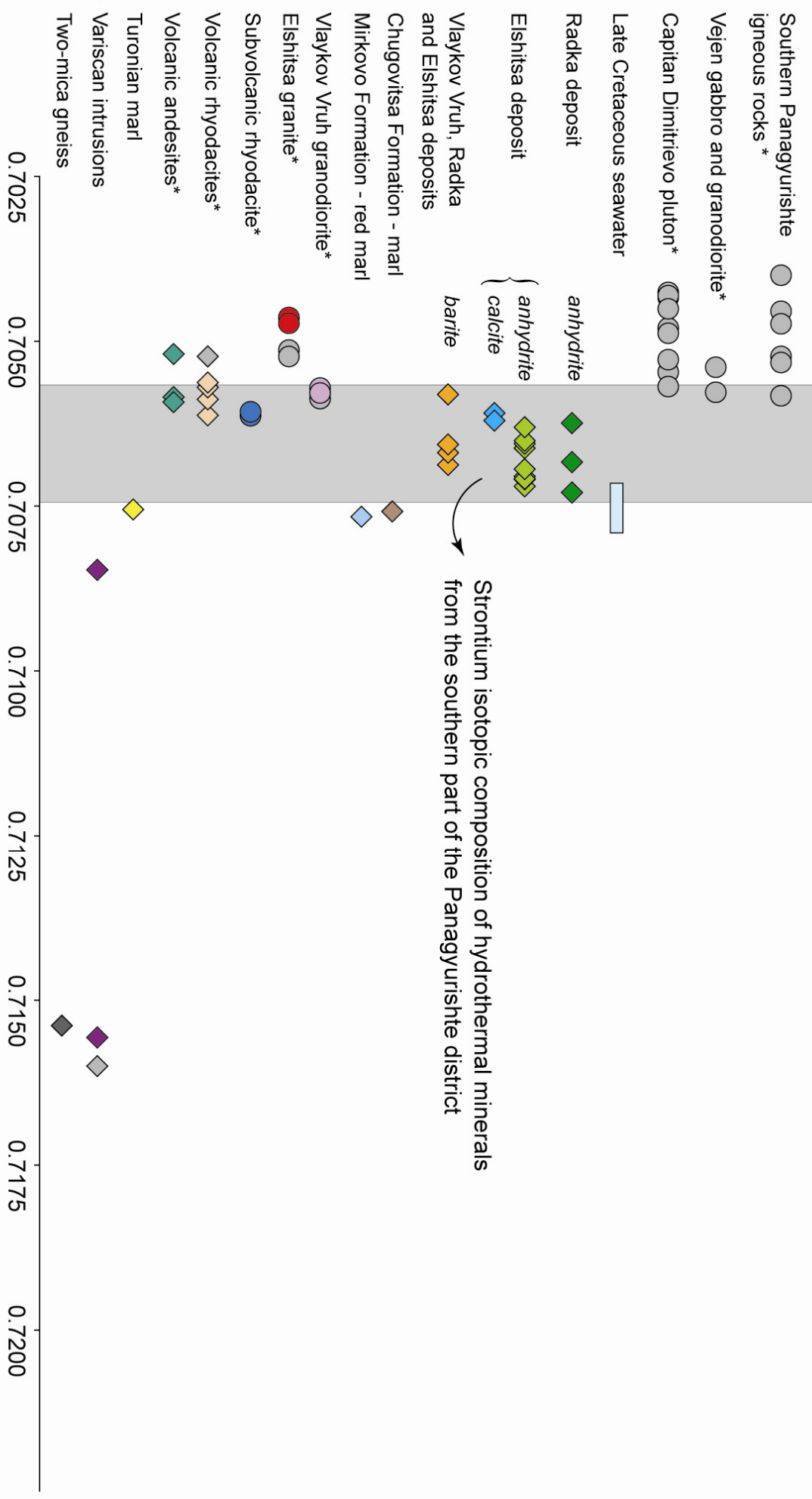




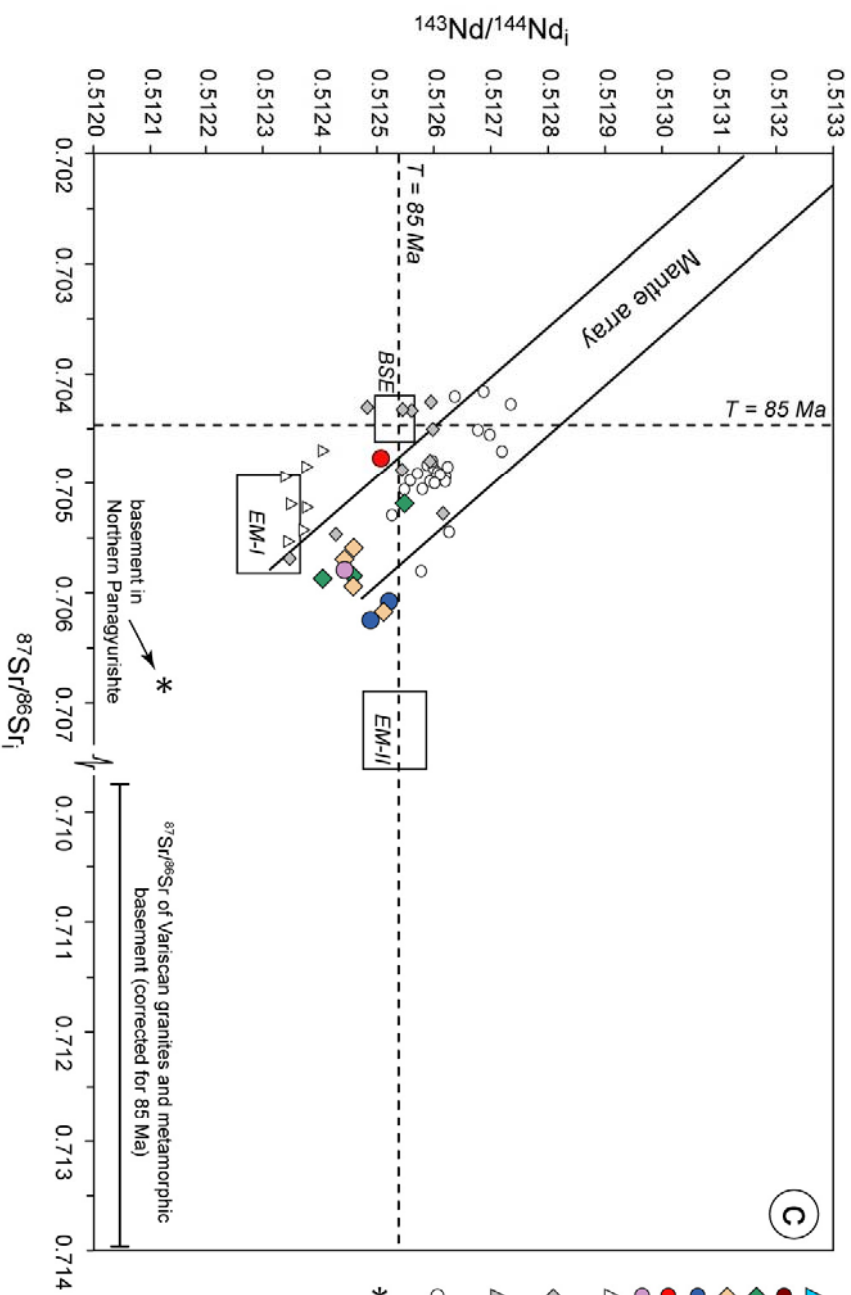
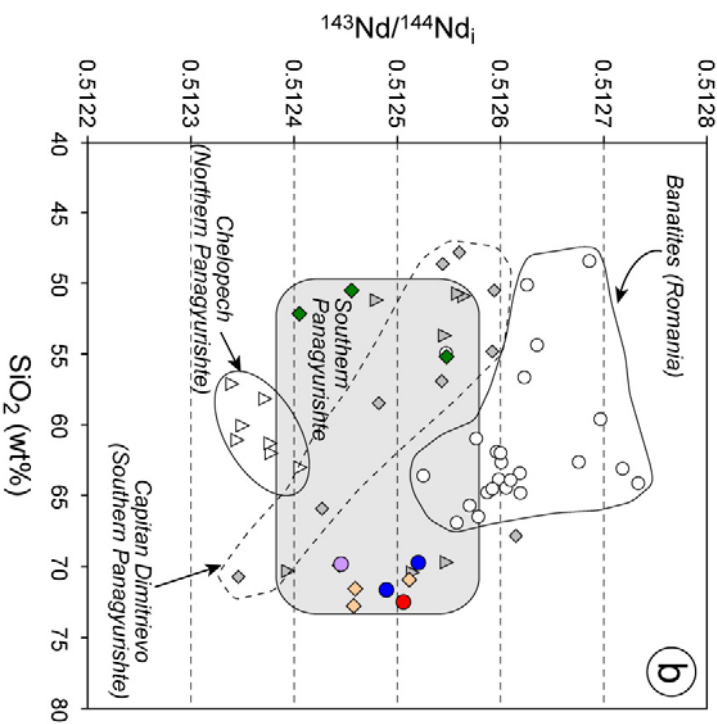
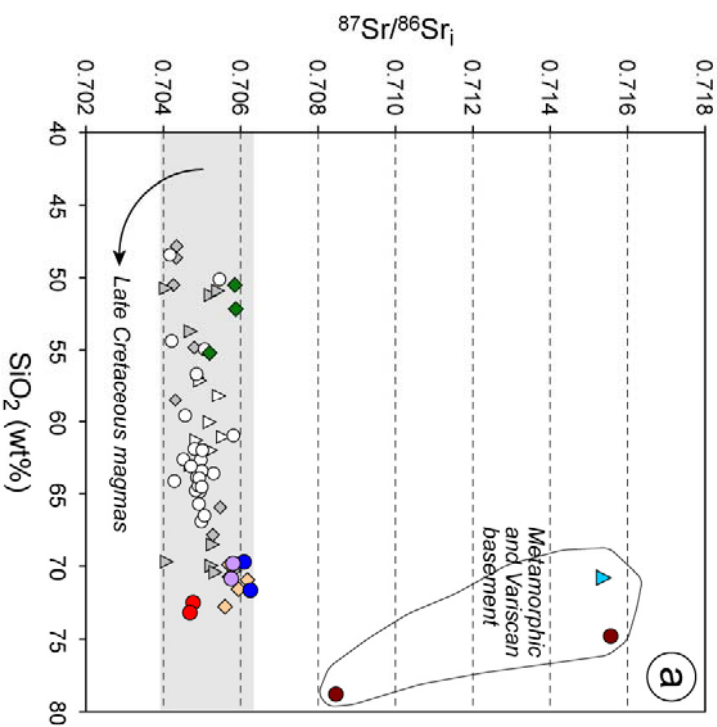
Northern Panagyurishte



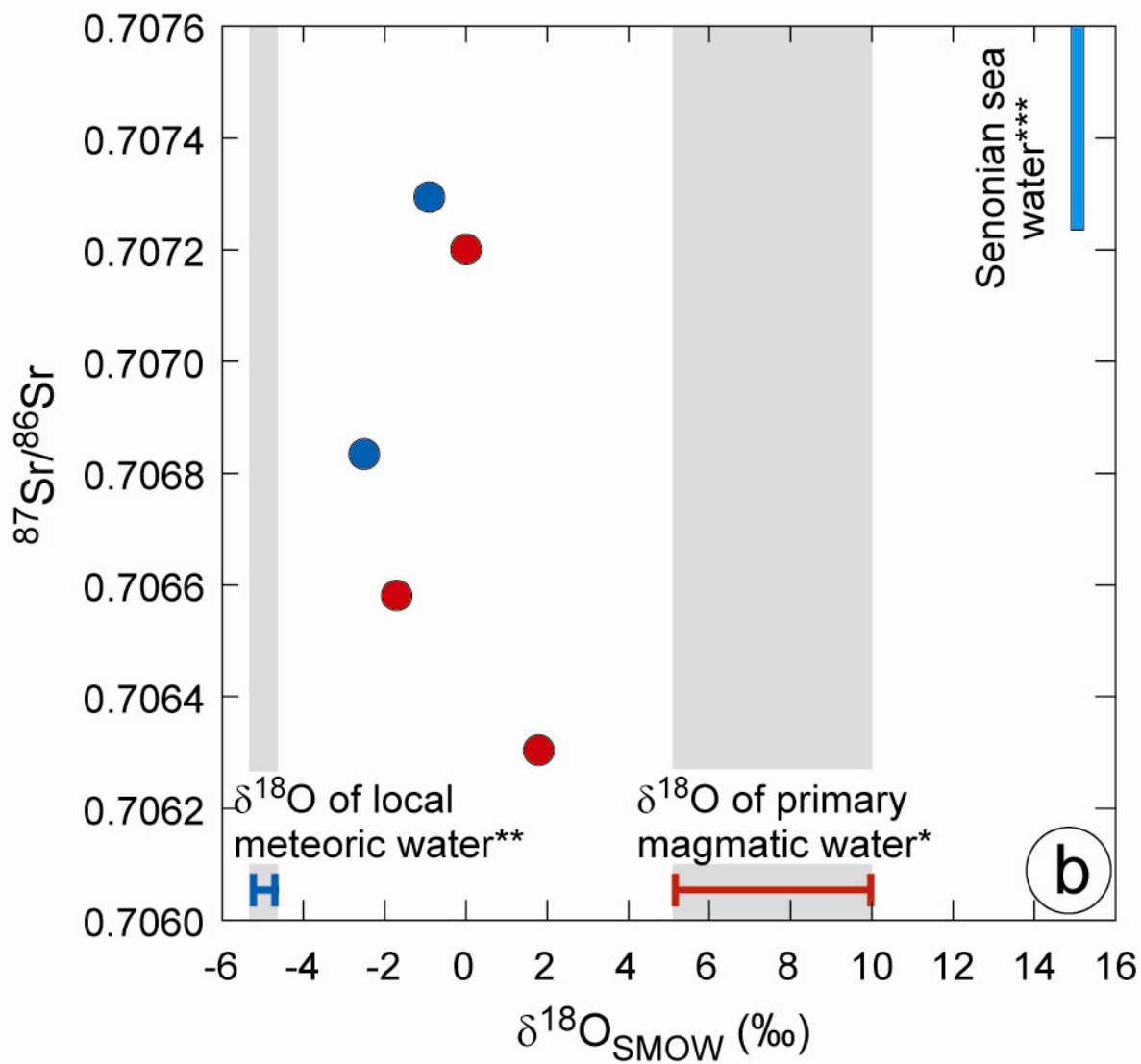
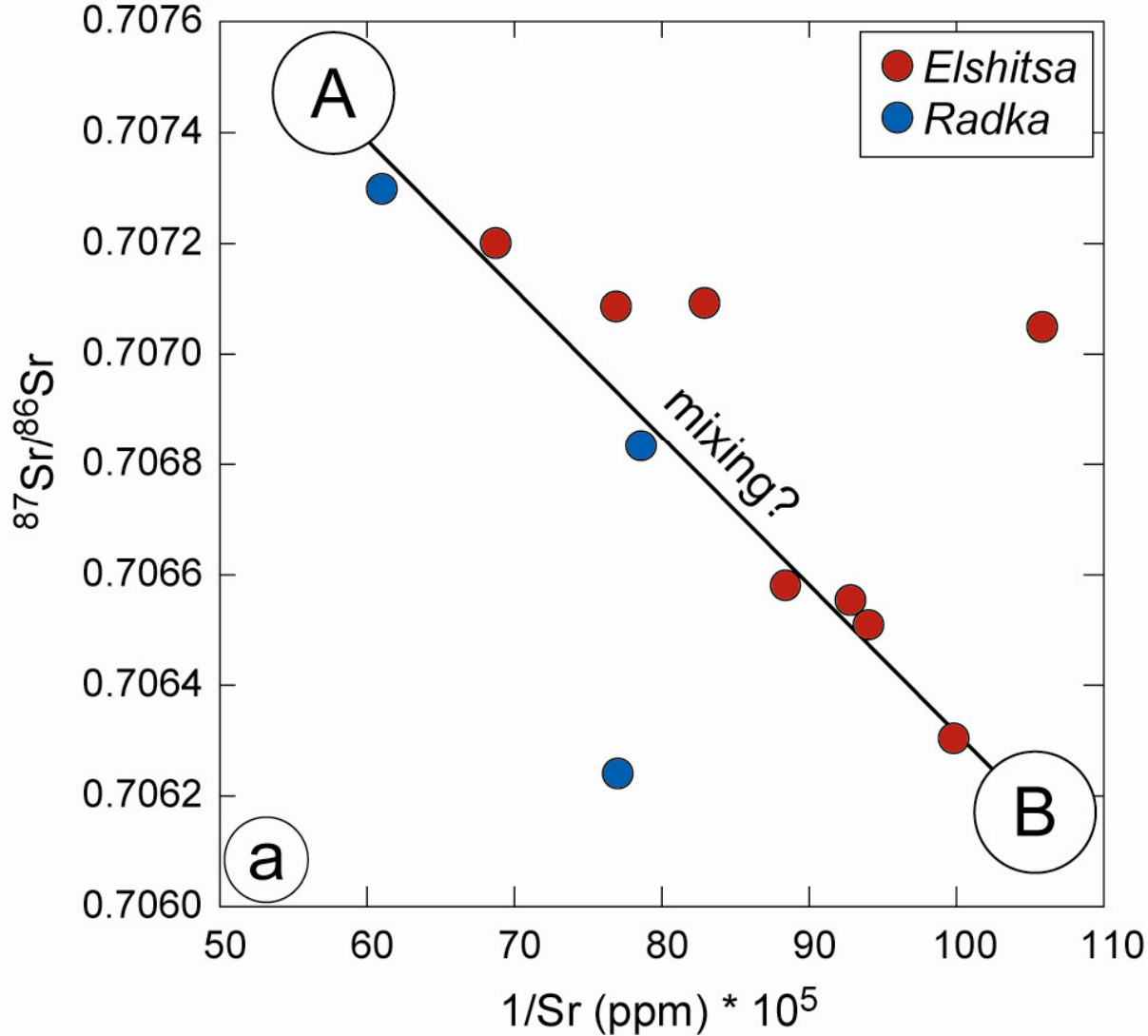
Southern Panagyurishte

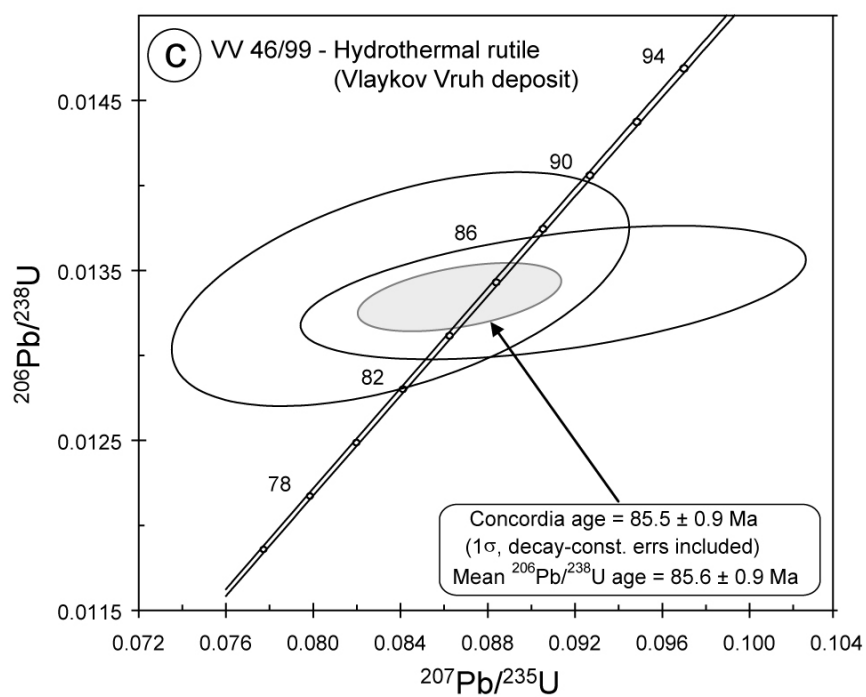
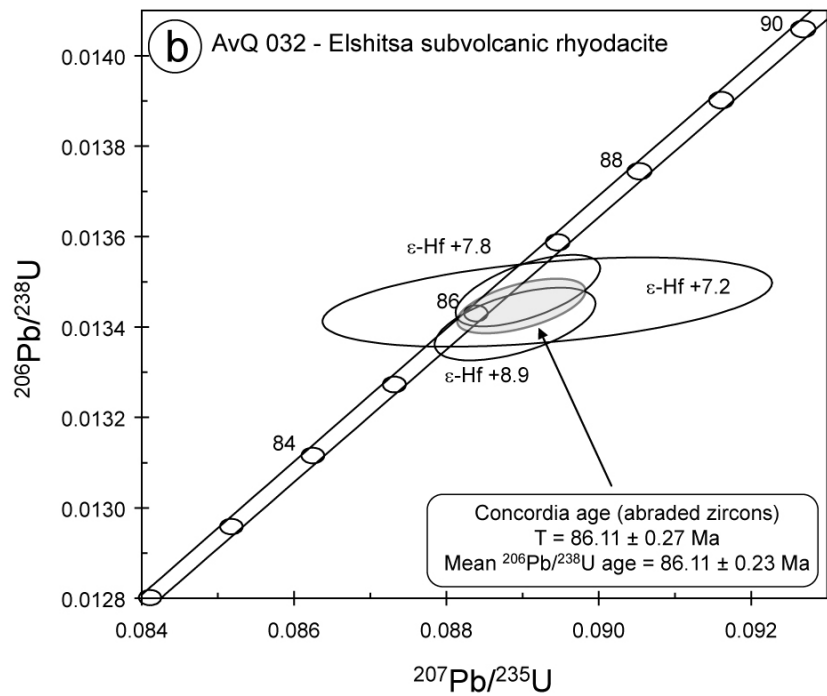
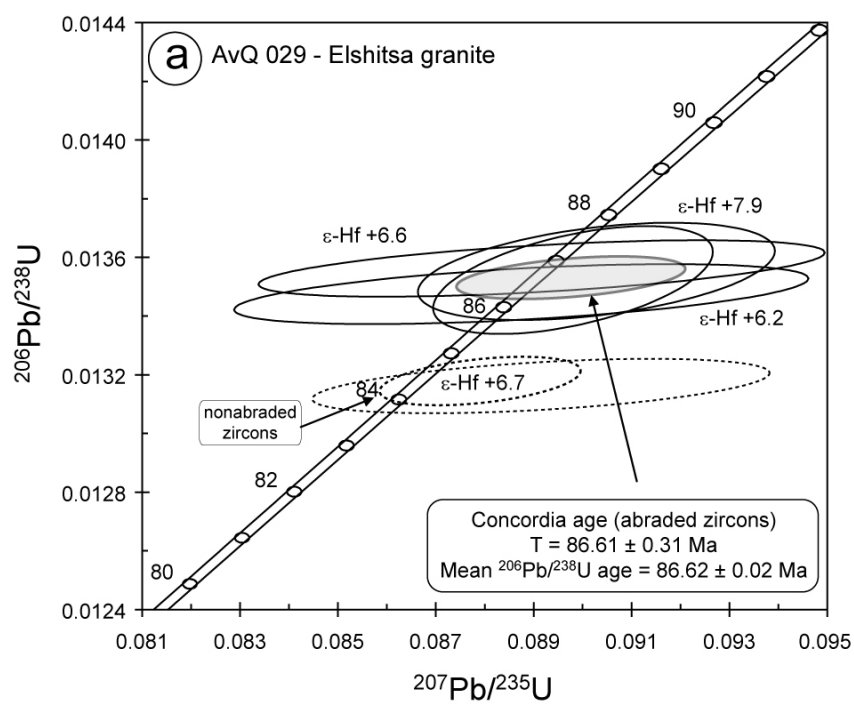


$^{87}\text{Sr}/^{86}\text{Sr}_i$



- ▲ - Two-mica gneiss
- - Variscan granite
- ◆ - Andesitic volcanic rocks
- ◇ - Dacitic and rhyodacitic volcanic rocks
- - Subvolcanic rhyodacites
- - Eishitsa granite
- - Vlyakov Vruh granodiorite
- △ - Chelopech andesites (Stoykov et al. 2004; corrected for 90 Ma)
- ◇ - Captain Dimitrevo pluton (Kamenov et al. 2003; corrected for 80 Ma)
- △ - Southern Panagyurishte (Peytcheva et al. 2008; corrected for 85 Ma)
- - Banaites (Dupont et al. 2002; corrected for 80 Ma)
- * - Variscan granite (von Quadt et al. 2005; corrected for 85 Ma)





Appendix 1 - Analytical techniques *(could be printed in smaller case – font 9)*

Whole-rock major oxides were analysed by X-ray fluorescence spectrometry on powdered samples fused with $\text{Li}_2\text{B}_4\text{O}_7$ on a Philips PW2400 spectrometer with Rh anode at the “Centre d’Analyses Minérales”, University of Lausanne, Switzerland. Trace elements and Rare Earth elements (REE) were analysed by laser-ablation inductively-coupled-plasma mass-spectrometry on the same pellets at ETH - Zurich, Switzerland, applying the methodology described in Günther et al. (2001). Analytical results for major, REE and trace elements are presented in the Appendix 2.

Lead isotope analyses on whole-rocks were carried out at ETH – Zurich, Switzerland. Samples of about 50 mg of finely ground whole rock were washed in 3 N HNO_3 at 80°C for 30 minutes. The residues were digested in 10-ml PFA-Teflon beakers with a mixture of 2 ml HF (21 N) and 4 ml HNO_3 (7.5 N) at 160°C. After 5 days, the samples were dried and transferred to a nitrate form. Lead was separated by Sr Resin[®] (Eichrom, 50-100 micron) in 100 μl columns. The total procedure blank for lead was better than 10pg. Pb was loaded on outgassed Re-filaments using the silica gel technique and measured on a MAT Finnigan 262 mass spectrometer in static mode. The fractionation for lead was based on repeated measurements of the SRM982 standard. The fractionation factor was 0.095% per atomic mass unit. The analytical errors (2σ) were 0.03 % for $^{206}\text{Pb}/^{204}\text{Pb}$, 0.04 % for $^{207}\text{Pb}/^{204}\text{Pb}$ and 0.08 % for $^{208}\text{Pb}/^{204}\text{Pb}$.

Lead isotope analyses on separates of chalcopyrite, pyrite, galena, bornite, enargite, and sphalerite-galena mixtures were carried out at the Department of Mineralogy, University of Geneva, Switzerland. Samples of 50 mg of ore minerals (except for galena - 10 μg) were digested in sealed 20-ml Teflon beakers with a mixture of 3 ml 7 M HCl and 1 ml conc. HNO_3 at 180°C. After conversion of the samples to bromide form, lead was separated by ion exchange chromatography. Procedural blanks were less than 120 pg Pb. Fractions of the purified lead were loaded on Re-filaments using the silica gel technique and lead isotope ratios were measured on a MAT Finnigan 262 thermal ionization mass spectrometer in static mode. Lead isotope ratios were corrected for fractionation by a +0.10 % atomic mass unit correction factor based on repeated analyses of the SRM981 international standard. The analytical errors (2σ) were 0.07 % for $^{206}\text{Pb}/^{204}\text{Pb}$, 0.10 % for the $^{207}\text{Pb}/^{204}\text{Pb}$, and 0.14 % for the $^{208}\text{Pb}/^{204}\text{Pb}$.

Strontium and neodymium isotope analyses on whole-rock samples were conducted at the Department of Mineralogy, University of Geneva, Switzerland. About 200 mg of finely ground rock were digested with concentrated HF/ HNO_3 in stainless steel bomb with Teflon liners. All solutions were evaporated and the dry residues were dissolved in 2.5 N HCl, followed by Sr and Nd separation in ion exchange columns. Strontium and neodymium isotope analyses were performed on a seven-collector MAT Finnigan 262 thermal ionization mass spectrometer with extended

geometry and stigmatic focusing using double Re-filaments. Strontium isotopic analysis was conducted in a semidynamic mode (triple collectors, measurement in jumping mode). The Sr isotope ratios were mass fractionation corrected to $^{88}\text{Sr}/^{86}\text{Sr} = 8.375209$ and normalized to the Eimer and Amend[®] SrCO₃ standard, with $^{87}\text{Sr}/^{86}\text{Sr} = 0.708000$ using an average of $0.708028 \pm 5 \times 10^{-6}$ (2SE; n = 52) measured during the period of analyses. $^{143}\text{Nd}/^{144}\text{Nd}$ was measured in a semidynamic mode (quadruple collectors, measurement in jumping mode), mass fraction corrected to $^{146}\text{Nd}/^{144}\text{Nd} = 0.721903$, and normalized to the La Jolla standard = 0.511835. An average of $0.511838 \pm 6 \times 10^{-6}$ (2SE; n = 28) was measured during the analyses.

For strontium isotope analyses of hydrothermal minerals, approximately 100 mg of finely ground anhydrite, barite and calcite have been used. Barite and anhydrite were leached overnight in 6 N HCl at 110°C, and calcite samples were dissolved in 1.5 N HCl for 20 min. After evaporation of the solutions, the dry residues were dissolved in 2.5 N HCl, followed by Sr and Nd separation in cation exchange columns. Analytical conditions during the analyses on the mass spectrometer were similar to those, used for the whole-rock samples. The Sr-content of anhydrite, barite and calcite samples was determined by atomic absorption with an analytical precision of $\pm 5\%$, using a GBS 905AA spectrophotometer at the Atomic Absorption Laboratory of ISTO-CNRS, Orléans, France.

U-Pb dating of magmatic zircons and hydrothermal rutiles was performed at ETH – Zurich, Switzerland. High-precision “conventional” U-Pb zircon analyses were carried out on single zircon grains (except for 3 analyses, for which a multigrain technique was used). Selected zircons were air-abraded to remove marginal zones with lead loss, washed in warm 4 N HNO₃ and rinsed several times with distilled water and acetone in an ultrasonic bath. Dissolution and chemical extraction of U and Pb was performed using miniaturised bombs and anion exchange columns. Blanks for the entire procedure were < 2 pg Pb and 0.5 pg U. A mixed $^{205}\text{Pb}/^{235}\text{U}$ tracer solution was used for all analyses. Both Pb and U were loaded with 1 μl of silica gel-phosphoric acid mixture on outgassed single Re-filaments and measured on a MAT Finnigan 262 thermal ionisation mass spectrometer using an ion counter system. The performance of the ion counter system was checked by repeated measurements of the NBS 982 standard solution. The reproducibility of the $^{207}\text{Pb}/^{206}\text{Pb}$ ratio (0.467070) was better than 0.05 %. The calculations of the U/Pb ratios include uncertainties of the spike calibration; Pb blank measurements, common Pb correction, U and Pb fractionation and U decay constant errors. All uncertainties are included in the error propagation for each individual analysis. Mean age values are given at the 2σ level.

Hf isotope ratios of zircons were measured on a Nu Instruments multiple collector inductively coupled plasma mass spectrometer at ETH – Zurich. During analysis, the $^{176}\text{Hf}/^{177}\text{Hf}$ ratio of the JMC 475 standard was measured at 0.282141 ± 5 (1σ) using the $^{179}\text{Hf}/^{177}\text{Hf} = 0.7325$ ratio for normalisation (exponential law for mass correction). For the calculation of the ϵ -Hf values

the following present-day ratios $(^{176}\text{Hf}/^{177}\text{Hf})_{\text{CH}} = 0.28286$ and $(^{176}\text{Lu}/^{177}\text{Hf})_{\text{CH}} = 0.0334$ were used, and for 85 Ma an average $^{176}\text{Lu}/^{177}\text{Hf}$ ratio of 0.005 for all zircons was taken into account.

Appendix 2 - Major- and trace-element analyses of whole rock samples from the southern part of the Panagyurishte district

| Sample | E 59/99 | E 60/99 | E 17/00 | E 18/00 | E 19a/00 | E 19/00 | E 20/00 | R 01/00 | E 53/99A |
|--------------------------------|------------------|-----------------|------------------|------------------|--------------------|---------------|----------|----------|----------|
| Rock type | Muscovite gneiss | Two-mica gneiss | Variscan granite | Variscan granite | Turonian sandstone | Turonian marl | Andesite | Andesite | Andesite |
| SiO ₂ (%) | 76,78 | 70,77 | 78,80 | 74,79 | 39,67 | 19,38 | 52,16 | 55,19 | 50,51 |
| TiO ₂ | 0,05 | 0,56 | 0,08 | 0,10 | 0,44 | 0,24 | 0,76 | 0,65 | 0,77 |
| Al ₂ O ₃ | 15,01 | 14,06 | 12,70 | 14,46 | 9,34 | 5,84 | 18,49 | 15,47 | 16,95 |
| Fe ₂ O ₃ | 0,96 | 4,56 | 0,50 | 1,31 | 4,03 | 5,35 | 7,95 | 7,89 | 8,47 |
| MnO | 0,01 | 0,13 | 0,01 | 0,01 | 0,08 | 0,16 | 0,14 | 0,12 | 0,29 |
| MgO | 0,58 | 0,89 | 0,09 | 0,19 | 1,02 | 1,11 | 3,11 | 4,18 | 6,06 |
| CaO | 0,22 | 1,72 | 1,95 | 1,15 | 23,42 | 36,56 | 7,87 | 5,98 | 5,20 |
| Na ₂ O | 0,05 | 2,51 | 4,31 | 3,83 | 0,53 | 0,09 | 5,55 | 2,16 | 5,12 |
| K ₂ O | 4,62 | 1,80 | 0,54 | 3,55 | 1,46 | 0,75 | 0,16 | 4,38 | 0,68 |
| P ₂ O ₅ | 0,15 | 0,41 | 0,09 | 0,09 | 0,08 | 0,19 | 0,21 | 0,19 | 0,19 |
| LOI | 1,91 | 2,85 | 1,27 | 0,60 | 15,80 | 24,28 | 3,56 | 4,09 | 5,89 |
| Total | 100,33 | 100,25 | 100,34 | 100,08 | 95,88 | 93,92 | 99,97 | 100,32 | 100,13 |
| Be (ppm) | 3,6 | 2,1 | 3,0 | 3,2 | 2,3 | 2,3 | 1,4 | 1,1 | 1,9 |
| Sc | 5,4 | 10,7 | 2,3 | 2,6 | 9,6 | 4,6 | 21 | 26 | 32 |
| V | 31 | 56 | 36 | 31 | 65 | 58 | 254 | 240 | 243 |
| Cr | 10,1 | 22 | 9,3 | 11 | 41 | 25 | 18 | 44 | 30 |
| Co | 1,0 | 3,9 | 0,60 | 1,0 | 5,6 | 2,3 | 17 | 20 | 28 |
| Ni | 4,2 | 11 | 9,6 | 6,2 | 25 | 12 | 9,4 | 13 | 16 |
| Cu | 9,2 | 25 | 8,9 | 7,2 | 23 | 18 | 72 | 63 | 11 |
| Rb | 144 | 49 | 16 | 121 | 53 | 34 | 6 | 65 | 22 |
| Sr | 8,7 | 126 | 717 | 169 | 834 | 1139 | 550 | 503 | 159 |
| Y | 9,4 | 134 | 9,9 | 11 | 25 | 14 | 19 | 16 | 15 |
| Zr | 30 | 336 | 59 | 53 | 160 | 38 | 108 | 86 | 48 |
| Nb | 7,4 | 9,4 | 5,0 | 7,7 | 6,6 | 3,8 | 4,8 | 3,4 | 1,7 |
| Cs | 1,8 | 0,44 | 0,34 | 0,92 | 1,8 | 1,9 | 0,07 | 0,51 | 0,19 |
| Ba | 515 | 506 | 258 | 479 | 317 | 213 | 25 | 556 | 103 |
| La | 4,0 | 72 | 22 | 17 | 31 | 14 | 22 | 20 | 15 |
| Ce | 7,3 | 143 | 40 | 30 | 57 | 25 | 44 | 39 | 32 |
| Pr | 0,89 | 18 | 4,4 | 3,3 | 6,4 | 2,8 | 5,2 | 4,6 | 4,0 |
| Nd | 3,4 | 78 | 16 | 12 | 26 | 11 | 22 | 20 | 17 |
| Sm | 0,92 | 18 | 3,0 | 2,3 | 5,0 | 2,3 | 4,4 | 4,0 | 3,8 |
| Eu | 0,18 | 3,1 | 0,53 | 0,44 | 0,72 | 0,86 | 1,2 | 0,98 | 1,0 |
| Gd | 1,0 | 19 | 2,3 | 1,9 | 3,9 | 2,0 | 3,9 | 3,3 | 3,3 |
| Tb | 0,27 | 3,3 | 0,40 | 0,33 | 0,69 | 0,36 | 0,57 | 0,50 | 0,50 |
| Dy | 1,8 | 21 | 2,0 | 1,8 | 3,8 | 1,9 | 3,5 | 2,8 | 2,7 |
| Ho | 0,28 | 4,4 | 0,39 | 0,37 | 0,86 | 0,43 | 0,74 | 0,59 | 0,58 |
| Er | 0,86 | 12 | 0,76 | 1,1 | 2,3 | 1,0 | 1,8 | 1,6 | 1,5 |
| Tm | 0,13 | 1,7 | 0,15 | 0,18 | 0,30 | 0,15 | 0,28 | 0,23 | 0,23 |
| Yb | 1,2 | 12 | 0,91 | 1,2 | 2,3 | 1,1 | 2,1 | 1,8 | 1,7 |
| Lu | 0,22 | 1,7 | 0,15 | 0,16 | 0,30 | 0,14 | 0,31 | 0,23 | 0,23 |
| Hf | 1,3 | 8,5 | 1,9 | 1,7 | 4,0 | 1,2 | 2,8 | 2,3 | 1,4 |
| Ta | 0,96 | 0,24 | 0,66 | 0,68 | 0,38 | 0,30 | 0,32 | 0,19 | 0,07 |
| Pb | 1,6 | 125 | 5,6 | 12 | 14 | 10,7 | 10,3 | 10,7 | 2,3 |
| Th | 1,6 | 21 | 6,1 | 6,8 | 7,8 | 4,6 | 6,5 | 6,7 | 3,0 |
| U | 1,2 | 3,6 | 2,0 | 1,8 | 1,6 | 1,3 | 1,6 | 1,7 | 1,0 |

Appendix 2 (continuation)

| Sample | E 53/99B | R 17/99 | VV 35/99 | R 19/99 | VV 36/99 | E 52/99 | E 66/99 | AVQ032 | E 16/97 |
|--------------------------------|------------|------------|------------|------------|------------------------|------------------------|------------------------|------------------------|------------------|
| Rock type | Rhyodacite | Rhyodacite | Rhyodacite | Rhyodacite | Subvolcanic rhyodacite | Subvolcanic rhyodacite | Subvolcanic rhyodacite | Subvolcanic rhyodacite | Elshitsa granite |
| SiO ₂ (%) | 72,75 | 69,88 | 70,91 | 71,55 | 71,63 | 70,59 | 69,71 | 69,68 | 72,48 |
| TiO ₂ | 0,22 | 0,20 | 0,25 | 0,24 | 0,27 | 0,31 | 0,28 | 0,27 | 0,22 |
| Al ₂ O ₃ | 13,77 | 14,09 | 15,36 | 14,67 | 14,54 | 14,72 | 14,69 | 17,59 | 14,64 |
| Fe ₂ O ₃ | 2,06 | 2,09 | 2,28 | 2,39 | 2,12 | 2,83 | 3,35 | 2,32 | 2,00 |
| MnO | 0,09 | 0,03 | 0,15 | 0,04 | 0,05 | 0,08 | 0,06 | 0,06 | 0,06 |
| MgO | 1,06 | 2,05 | 0,71 | 1,23 | 0,88 | 1,99 | 2,05 | 2,44 | 0,58 |
| CaO | 1,37 | 0,26 | 1,58 | 2,21 | 1,61 | 0,79 | 0,31 | 0,37 | 2,55 |
| Na ₂ O | 3,91 | 2,83 | 4,52 | 4,27 | 4,48 | 4,47 | 3,01 | 2,91 | 3,65 |
| K ₂ O | 2,96 | 4,45 | 3,59 | 0,70 | 2,80 | 2,91 | 4,08 | 4,01 | 3,39 |
| P ₂ O ₅ | 0,06 | 0,07 | 0,07 | 0,09 | 0,08 | 0,10 | 0,11 | 0,08 | 0,06 |
| LOI | 2,21 | 2,94 | 0,97 | 2,65 | 1,05 | 1,57 | 2,65 | n.a. | 0,41 |
| Total | 100,47 | 98,89 | 100,38 | 100,04 | 99,51 | 100,35 | 100,30 | 99,73 | 100,04 |
| Be (ppm) | 5,7 | 2,0 | 3,4 | 4,1 | 5,1 | 2,1 | 2,4 | n.a. | 2,0 |
| Sc | 5,2 | 3,0 | 3,8 | 4,6 | 4,7 | 5,9 | 5,0 | 5,5 | 3,6 |
| V | 65 | 62 | 59 | 62 | 59 | 72 | 71 | n.a. | 56 |
| Cr | 14 | 9,5 | 11 | 7,85 | 12 | 9,89 | 11 | 25 | 16 |
| Co | 3,9 | 3,3 | 3,1 | 3,5 | 5,2 | 4,1 | 4,8 | 4,4 | 2,9 |
| Ni | 118 | 4,8 | 3,9 | <4,1 | 4,8 | 3,5 | 7,7 | 6,1 | 4,5 |
| Cu | 52 | 13 | 24 | 5,0 | 9,7 | 16 | 776 | 1064 | 10 |
| Rb | 73 | 81 | 67 | 16 | 64 | 59 | 61 | 60 | 100 |
| Sr | 241 | 80 | 408 | 438 | 238 | 130 | 100 | 95 | 248 |
| Y | 8,4 | 7,1 | 11 | 7,7 | 10 | 10 | 8,5 | 9,3 | 8,4 |
| Zr | 90 | 76 | 81 | 84 | 90 | 99 | 83 | 90 | 89 |
| Nb | 3,6 | 3,4 | 3,8 | 4,0 | 3,9 | 3,8 | 3,0 | 3,0 | 3,6 |
| Cs | 0,53 | 0,36 | 0,35 | 0,16 | 0,39 | 0,21 | 0,42 | 0,52 | 0,96 |
| Ba | 1006 | 500 | 1236 | 300 | 946 | 759 | 655 | 518 | 563 |
| La | 22 | 23 | 24 | 32 | 26 | 21 | 18 | 20 | 22 |
| Ce | 34 | 36 | 37 | 54 | 44 | 37 | 30 | 32 | 34 |
| Pr | 3,2 | 3,3 | 3,8 | 4,6 | 4,3 | 3,7 | 3,2 | 3,4 | 3,1 |
| Nd | 11 | 11 | 13 | 15 | 15 | 13 | 12 | 12 | 11 |
| Sm | 1,9 | 1,7 | 2,2 | 2,2 | 2,5 | 2,2 | 1,9 | 2,3 | 1,7 |
| Eu | 0,47 | 0,45 | 0,65 | 0,71 | 0,57 | 0,62 | 0,57 | 0,71 | 0,50 |
| Gd | 1,6 | 1,2 | 1,9 | 1,7 | 1,8 | 1,7 | 1,6 | 1,6 | 1,4 |
| Tb | 0,23 | 0,19 | 0,27 | 0,22 | 0,25 | 0,28 | 0,24 | 0,23 | 0,21 |
| Dy | 1,3 | 1,1 | 1,7 | 1,3 | 1,7 | 1,6 | 1,4 | 1,5 | 1,2 |
| Ho | 0,25 | 0,24 | 0,33 | 0,24 | 0,36 | 0,37 | 0,30 | 0,35 | 0,27 |
| Er | 0,76 | 0,71 | 0,98 | 0,79 | 1,11 | 1,11 | 0,84 | 1,01 | 0,80 |
| Tm | 0,13 | 0,12 | 0,17 | 0,10 | 0,17 | 0,17 | 0,15 | 0,16 | 0,15 |
| Yb | 0,92 | 0,96 | 1,2 | 1,1 | 1,3 | 1,3 | 1,1 | 1,3 | 1,0 |
| Lu | 0,17 | 0,15 | 0,22 | 0,17 | 0,20 | 0,21 | 0,17 | 0,19 | 0,18 |
| Hf | 2,3 | 2,0 | 2,3 | 2,5 | 2,4 | 2,6 | 2,2 | 2,2 | 2,3 |
| Ta | 0,29 | 0,29 | 0,31 | 0,30 | 0,33 | 0,33 | 0,23 | 0,23 | 0,30 |
| Pb | 11,7 | 7,2 | 4,3 | 2,9 | 3,5 | 3,8 | 11,9 | 20,6 | 6,4 |
| Th | 10,8 | 11 | 11 | 10 | 12 | 9,9 | 7,7 | 8,1 | 11 |
| U | 1,9 | 2,1 | 2,6 | 2,3 | 1,6 | 2,3 | 2,0 | 2,7 | 2,7 |

Appendix 2 (continuation)

| Sample | E 17/94 | AvQ 029 | VV 06/99 | VV 23/99 | AVQ 031 | R 16/99 | R 15/99 |
|--------------------------------|------------------|------------------|---------------------------|---------------------------|---------------------------|----------|------------------------|
| Rock type | Elshitsa granite | Elshitsa granite | Vlaykov Vruh granodiorite | Vlaykov Vruh granodiorite | Vlaykov Vruh granodiorite | Red marl | Senonian flysch (marl) |
| SiO ₂ (%) | 73,17 | 70,30 | 70,84 | 69,81 | 70,40 | 17,94 | 12,43 |
| TiO ₂ | 0,21 | 0,24 | 0,30 | 0,30 | 0,25 | 0,12 | 0,11 |
| Al ₂ O ₃ | 14,32 | 13,80 | 15,11 | 15,31 | 17,11 | 2,92 | 2,77 |
| Fe ₂ O ₃ | 2,03 | 2,69 | 3,10 | 3,05 | 1,84 | 1,46 | 1,19 |
| MnO | 0,05 | 0,14 | 0,04 | 0,06 | 0,06 | 0,14 | 0,20 |
| MgO | 0,52 | 0,69 | 1,08 | 1,12 | 0,87 | 0,52 | 0,60 |
| CaO | 2,37 | 2,85 | 1,83 | 2,27 | 2,77 | 43,83 | 45,79 |
| Na ₂ O | 3,29 | 3,39 | 3,65 | 4,17 | 3,33 | 0,04 | 0,06 |
| K ₂ O | 3,72 | 3,13 | 2,99 | 2,04 | 3,10 | 0,61 | 0,49 |
| P ₂ O ₅ | 0,06 | 0,08 | 0,10 | 0,10 | 0,07 | 0,06 | 0,08 |
| LOI | 0,65 | n.a. | 1,22 | 1,47 | n.a. | 22,61 | 27,82 |
| Total | 100,38 | 97,31 | 100,26 | 99,69 | 99,80 | 90,27 | 91,54 |
| Be (ppm) | 2,0 | n.a. | 1,8 | 3,0 | n.a. | <2,3 | 2,7 |
| Sc | 3,1 | 5,8 | 5,6 | 5,6 | 6,5 | 3,0 | 3,2 |
| V | 51 | n.a. | 78 | 78 | n.a. | 25 | 26 |
| Cr | 11 | 39 | 15 | 11 | 30 | 20 | 19 |
| Co | 3,0 | 5,6 | 3,5 | 3,3 | 4,9 | 3,3 | 2,2 |
| Ni | 6,1 | 9,8 | 6,3 | 3,7 | 4,2 | 13 | 12 |
| Cu | 353 | 28 | 233 | 63 | 50 | 26 | 22 |
| Rb | 113 | 106 | 44 | 31 | 54 | 22 | 18 |
| Sr | 228 | 286 | 260 | 359 | 374 | 401 | 436 |
| Y | 8,4 | 12 | 10 | 10 | 11 | 11 | 7,9 |
| Zr | 81 | 102 | 85 | 92 | 83 | 18 | 15 |
| Nb | 3,7 | 4,5 | 3,5 | 4,0 | 4,7 | 1,5 | 1,3 |
| Cs | 1,1 | 1,8 | 0,76 | 0,23 | 0,50 | 0,89 | 0,84 |
| Ba | 548 | 550 | 729 | 703 | 752 | 239 | 121 |
| La | 22 | 27 | 19 | 25 | 26 | 13 | 9,1 |
| Ce | 35 | 45 | 31 | 42 | 44 | 17 | 13 |
| Pr | 3,5 | 4,4 | 3,1 | 4,0 | 4,5 | 2,3 | 1,6 |
| Nd | 11 | 16 | 12 | 15 | 16 | 9,2 | 6,7 |
| Sm | 2,0 | 2,7 | 2,0 | 2,3 | 2,4 | 1,8 | 1,3 |
| Eu | 0,53 | 0,65 | 0,61 | 0,64 | 0,78 | 0,39 | 0,27 |
| Gd | 1,4 | 2,1 | 1,9 | 2,0 | 2,0 | 1,8 | 1,2 |
| Tb | 0,23 | 0,30 | 0,29 | 0,28 | 0,34 | 0,28 | 0,20 |
| Dy | 1,3 | 1,8 | 1,5 | 1,6 | 1,8 | 1,4 | 1,1 |
| Ho | 0,33 | 0,38 | 0,36 | 0,34 | 0,37 | 0,35 | 0,23 |
| Er | 0,86 | 1,1 | 0,88 | 1,0 | 1,3 | 0,84 | 0,59 |
| Tm | 0,16 | 0,23 | 0,15 | 0,17 | 0,21 | 0,13 | 0,08 |
| Yb | 1,0 | 1,8 | 1,2 | 1,3 | 1,6 | 0,81 | 0,56 |
| Lu | 0,20 | 0,24 | 0,17 | 0,20 | 0,25 | 0,12 | 0,08 |
| Hf | 2,2 | 2,5 | 2,3 | 2,5 | 2,2 | 0,47 | 0,38 |
| Ta | 0,34 | 0,35 | 0,30 | 0,34 | 0,37 | 0,09 | 0,08 |
| Pb | 7,2 | 8,6 | 4,4 | 4,2 | 25,08 | 12,39 | 5,1 |
| Th | 11 | 12 | 10 | 11 | 11 | 2,2 | 1,6 |
| U | 3,2 | 2,4 | 1,3 | 1,7 | 2,9 | 0,31 | 0,27 |

Table 1 – Geological setting and ore deposit types in the Panagyurishte district

| Pre-Mesozoic basement | Mesozoic sedimentary cover | Late-Cretaceous magmatism | Ore deposits (from N to S) | Size of the deposits | | | |
|--|---|---|---------------------------------|--|--|---|---|
| <p>- Paleozoic Balkanide type high-grade amphibolite facies metamorphic complex (Ivanov 1989a, b) or Srednogorie type metamorphic rocks (Cheshitev et al. 1995)</p> <p>- Cambrian-Ordovician low-grade metamorphic Diabase-Phyllitoid Complex or Berkovitsa Group (Haydoutov 2001)</p> <p>- Variscan (300-350 Ma) gabbrodiorites, quartz-diorites, tonalites, and granodiorites-granites intruding the metamorphic rocks (Dabovski et al. 1972; Peytcheva and von Quadt 2004; Carrigan et al. 2005).</p> | <p>- Late Carboniferous to Jurassic clastic and carbonate sedimentary rocks - Petrohan Terrigenous Group – unconformably covering the basement lithologies;</p> <p>- Sandstone suite – Turonian conglomerates and sandstones (Moev and Antonov 1978)</p> <p>- Coniacian to Maastrichtian clastic terrigenous sequences with subordinate pelagic marls, shales, turbidites, and volcanic intercalations (Karagjuleva et al. 1974; Aiello et al. 1977; Nachev and Nachev 1986).</p> | <p>- Andesites predominate in the northern and central parts of the district; dacitic volcanic rocks are more abundant in the southern sector, whereas rhyodacites and rhyolites occur exclusively in the central and southern parts (Dimitrov 1960; Ignatovski et al. 1990; Daieva and Chipchakova 1997; Lilov and Chipchakova 1999; Kouzmanov et al. 2001; Stoykov et al. 2004; Popov 2005; von Quadt et al. 2005; Kamenov et al. 2007). Due to a deeper erosional level, intrusive rocks become more abundant in the south (Ivanov et al., 2002).</p> <p>- Magmas are dominantly metaluminous and LILE-enriched with prevailing high-K calc-alkaline and shoshonitic trends (Daieva and Chipchakova 1997; Kamenov et al. 2007). The only exception is the Elshitsa magmatic structure in the southern part of the district which shows a typical calc-alkaline trend (Ignatovski et al. 1990; Kamenov et al. 2007). Geochemical signatures are typical of island-arc magmas in a subduction-related geodynamic setting (Boccaletti et al. 1974; Stanisheva-Vassileva 1980; von Quadt et al. 2005; Kamenov et al. 2007).</p> <p>- Recent radiogenic (Pb, Sr, Nd, Hf) isotope data along with U-Pb geochronological studies (von Quadt et al. 2005 and references therein) reveal the mixed crustal-mantle source of the magmas, with a progressively increasing mantle component towards the Southern Panagyurishte district in parallel with the migration of a 14-m.y.-long protracted magmatic and ore-forming activity, starting at ca. 92 Ma in the north at Elatsite porphyry Cu-Au (PGE) deposit and ending at ca. 78 Ma in the south at Capitan Dimitriev.</p> <p>- The Late-Cretaceous volcano-sedimentary complex is overlain by Santonian-Campanian red marly limestones of the Mirkovo Formation, and Campanian-Maastrichtian calcarenite and mudstone flysch of the Chugovitsa Formation (Moev and Antonov 1978). The Senonian flysch units were deposited as outer-fan lobes in several basins, which were possibly related to different volcanic centres (Aiello et al. 1977).</p> | Northern Panagyurishte district | Elatsite porphyry Cu-Au (-PGE) (in operation since 1981) | mined out 165 Mt @ 0.38 % Cu and 0.21 g/t Au; reserves 154 Mt @ 0.33 % Cu, 0.96 g/t Au, 0.19 g/t Ag, 0.007 g/t Pd and 0.002 g/t Pt (Strashimirov et al. 2002; Tarkian et al. 2003) | | |
| | | | | Chelopech HS epithermal Cu-Au (in operation since 1954) | total of past production and remaining resources 42.5 Mt @ 1.28 % Cu, 3.4 g/t Au and 8.4 g/t Ag (Moritz et al. 2004) | | |
| | | | | Medet porphyry Cu (1964-1993) | 163 Mt @ 0.32 % Cu, 0.1 g/t Au (Strashimirov et al. 2002) | | |
| | | | Southern Panagyurishte district | | | Assarel porphyry Cu (in operation since 1976) | mined out 100 Mt @ 0.53% Cu ; reserves 254 Mt @ 0.41% Cu (Strashimirov et al. 2002) |
| | | | | | | Krassen HS epithermal Cu-Au (1962-1973) | 0.30 Mt @ 0.76% Cu (Strashimirov et al. 2002) |
| | | | | | | Petelovo porphyry Au | 8.09 Mt @ 0.88 g/t Au (Vassilev and Fargova 1998) |
| Radka HS epithermal Cu-Au (1942-1997) | 8.9 Mt @ 1.06 % Cu, 1.5 g/t Au, 25 g/t Ag (Moritz et al. 2004) | | | | | | |
| Tsar Assen porphyry Cu (1980-1995) | 6.6 Mt @ 0.47 % Cu (Strashimirov et al. 2002) | | | | | | |
| Elshitsa HS epithermal Cu-Au (1922-1999) | 4.5 Mt @ 1.13% Cu, 1.5 g/t Au and 15 g/t Ag (Moritz et al. 2004) | | | | | | |
| Vlaykov Vruh porphyry Cu (1962-1979) | mined out 9.793 Mt @ 0.46% Cu; reserves 22 700 t @ 0.32% Cu (Strashimirov et al. 2003) | | | | | | |

Abbreviations: HS - high-sulphidation

Table 2 - Pb, U and Th concentrations, measured Pb isotopic ratios and calculated Pb initial ratios of whole rock samples from the southern part of the Panagyurishte district

| Sample | Rock type | Pb (ppm) | U (ppm) | Th (ppm) | $^{206}\text{Pb}/^{204}\text{Pb}$ measured | $^{207}\text{Pb}/^{204}\text{Pb}$ measured | $^{208}\text{Pb}/^{204}\text{Pb}$ measured | $^{206}\text{Pb}/^{204}\text{Pb}_i$ (t=85 Ma) | $^{207}\text{Pb}/^{204}\text{Pb}_i$ (t=85 Ma) | $^{208}\text{Pb}/^{204}\text{Pb}_i$ (t=85 Ma) |
|----------|---------------------------|-------------|------------|-------------|---|---|---|--|--|--|
| E 60/99 | Two-mica gneiss | 125 | 3,6 | 21 | 18,505 | 15,622 | 38,561 | 18,48 | 15,62 | 38,51 |
| E 17/00 | Variscan granite | 5,6 | 2,0 | 6,1 | 18,855 | 15,612 | 38,648 | 18,56 | 15,60 | 38,35 |
| E 19/00 | Turonian marl | 10,7 | 1,3 | 4,6 | 18,905 | 15,612 | 38,906 | 18,80 | 15,61 | 38,79 |
| E 20/00 | Andesite | 10,3 | 1,6 | 6,5 | 18,606 | 15,620 | 38,686 | 18,47 | 15,61 | 38,51 |
| R 01/00 | Andesite | 10,7 | 1,7 | 6,7 | 18,581 | 15,601 | 38,602 | 18,45 | 15,59 | 38,43 |
| E 53/99A | Andesite | 2,3 | 1,0 | 3,0 | 18,776 | 15,599 | 38,212 | 18,40 | 15,58 | 37,86 |
| E 53/99B | Rhyodacite | 11,8 | 1,9 | 10,8 | 18,591 | 15,634 | 38,717 | 18,45 | 15,63 | 38,47 |
| R 17/99 | Rhyodacite | 7,2 | 2,1 | 11,2 | 18,720 | 15,609 | 38,768 | 18,48 | 15,60 | 38,34 |
| VV 35/99 | Rhyodacite | 4,3 | 2,6 | 10,6 | 18,885 | 15,638 | 38,891 | 18,37 | 15,61 | 38,21 |
| R 19/99 | Rhyodacite | 2,9 | 2,3 | 10,5 | 19,248 | 15,643 | 39,284 | 18,57 | 15,61 | 38,29 |
| VV 36/99 | Subvolcanic rhyodacite | 3,5 | 1,6 | 12,4 | 18,591 | 15,613 | 38,645 | 18,21 | 15,59 | 37,69 |
| E 52/99 | Subvolcanic rhyodacite | 3,8 | 2,3 | 9,9 | 18,864 | 15,642 | 39,041 | 18,36 | 15,62 | 38,31 |
| E 66/99 | Subvolcanic rhyodacite | 11,9 | 2,0 | 7,7 | 18,782 | 15,647 | 38,736 | 18,64 | 15,64 | 38,56 |
| AvQ 032 | Subvolcanic rhyodacite | 2,6 | 1,9 | 8,1 | 18,739 | 15,668 | 38,895 | 18,13 | 15,64 | 38,03 |
| E 16/97 | Elshitsa granite | 6,4 | 2,7 | 11,5 | 18,752 | 15,614 | 38,837 | 18,40 | 15,60 | 38,34 |
| AvQ 029 | Elshitsa granite | 12,3 | 2,5 | 9,0 | 18,585 | 15,588 | 38,642 | 18,42 | 15,58 | 38,44 |
| VV 06/99 | Vlaykov Vruh granodiorite | 4,4 | 1,3 | 10,0 | 18,698 | 15,615 | 38,796 | 18,46 | 15,60 | 38,17 |
| VV 23/99 | Vlaykov Vruh granodiorite | 4,2 | 1,7 | 10,8 | 18,885 | 15,639 | 38,891 | 18,55 | 15,62 | 38,18 |
| AvQ 031 | Vlaykov Vruh granodiorite | 3,4 | 2,0 | 11,3 | 18,892 | 15,641 | 38,992 | 18,40 | 15,62 | 38,07 |

2 σ error: $^{206}\text{Pb}/^{204}\text{Pb} = 0.006$; $^{207}\text{Pb}/^{204}\text{Pb} = 0.006$; $^{208}\text{Pb}/^{204}\text{Pb} = 0.031$

Table 3 - Pb isotopic compositions of ore minerals from the Elshitsa, Radka and Vlaykov Vruh deposits. Mineralisation stages are according to Figures 6 and 9.

| Sample | Deposit | Mineral | Mineralisation stage | $^{206}\text{Pb}/^{204}\text{Pb}$ | $^{207}\text{Pb}/^{204}\text{Pb}$ | $^{208}\text{Pb}/^{204}\text{Pb}$ |
|-----------|--------------|---------|----------------------|-----------------------------------|-----------------------------------|-----------------------------------|
| E 58/97 | Elshitsa | py | stage-I | 18,493 | 15,608 | 38,497 |
| E G5 | Elshitsa | py | stage-I | 18,495 | 15,608 | 38,498 |
| E 33/99 | Elshitsa | cp | stage-II | 18,492 | 15,615 | 38,522 |
| E 36/98 | Elshitsa | cp | stage-II | 18,518 | 15,613 | 38,518 |
| BU10A | Elshitsa | cp | stage-II | 18,534 | 15,632 | 38,615 |
| E 500 | Elshitsa | bn | stage-II | 18,513 | 15,622 | 38,551 |
| E 01/94 | Elshitsa | sp-gn | stage-III | 18,525 | 15,635 | 38,590 |
| E 06/99 | Elshitsa | gn | stage-III | 18,494 | 15,610 | 38,514 |
| BU10B | Elshitsa | gn | stage-III | 18,505 | 15,620 | 38,543 |
| VV 124/70 | Vlaykov Vruh | py | stage-III | 18,475 | 15,610 | 38,502 |
| VV 125/70 | Vlaykov Vruh | cp | stage-III | 18,481 | 15,618 | 38,531 |
| VV 17/99 | Vlaykov Vruh | py | stage-IV | 18,432 | 15,636 | 38,519 |
| VV 24/99 | Vlaykov Vruh | py | stage-V | 18,531 | 15,637 | 38,621 |
| R 10/97 | Radka | py | stage-I | 18,491 | 15,616 | 38,533 |
| R 79/97 | Radka | cp | stage-II | 18,490 | 15,621 | 38,567 |
| R 614 | Radka | en | stage-II | 18,488 | 15,619 | 38,537 |
| R 201/93 | Radka | sp-gn | stage-III | 18,497 | 15,627 | 38,573 |
| R 64/97 | Radka | py | stage-IV | 18,518 | 15,647 | 38,630 |

Abbreviations:

py - pyrite; cp - chalcopyrite; bn - bornite; sp - sphalerite; gn - galena; en - enargite

2 σ error: $^{206}\text{Pb}/^{204}\text{Pb} = 0.013$; $^{207}\text{Pb}/^{204}\text{Pb} = 0.016$; $^{208}\text{Pb}/^{204}\text{Pb} = 0.054$

Table 4 - Rb, Sr, Sm, and Nd concentrations and isotopic compositions of whole rock samples from the southern part of the Panagyurishte district

| Sample | Rock type | Rb (ppm) | Sr (ppm) | Rb/Sr | $^{87}\text{Rb}/^{86}\text{Sr}$ | $^{87}\text{Sr}/^{86}\text{Sr}$ | Error ($\pm 2\text{SE}$)* | $^{87}\text{Sr}/^{86}\text{Sr}_i$ ($t=85\text{ Ma}$) | Sm (ppm) | Nd (ppm) | Sm/Nd | $^{147}\text{Sm}/^{144}\text{Nd}$ | $^{143}\text{Nd}/^{144}\text{Nd}$ | Error ($\pm 2\text{SE}$)* | $^{143}\text{Nd}/^{144}\text{Nd}_i$ ($t=85\text{ Ma}$) | ϵ_{Nd} ($t=85\text{ Ma}$) |
|----------|---------------------------|-------------|-------------|--------|---------------------------------|---------------------------------|--------------------------------|---|-------------|-------------|--------|-----------------------------------|-----------------------------------|--------------------------------|---|--|
| E 60/99 | Two-mica gneiss | 49 | 126 | 0,3896 | 1,0913 | 0,716704 | 9 | 0,71539 | | | | | | | | |
| E 17/00 | Variscan granite | 16 | 717 | 0,0227 | 0,0637 | 0,708537 | 7 | 0,70846 | | | | | | | | |
| E 18/00 | Variscan granite | 121 | 169 | 0,7171 | 2,0091 | 0,717985 | 12 | 0,71556 | | | | | | | | |
| E 19/00 | Turonian marl | 34 | 1139 | 0,0296 | 0,0829 | 0,707653 | 26 | 0,70755 | | | | | | | | |
| E 20/00 | Andesite | 5,8 | 550 | 0,0105 | 0,0295 | 0,705908 | 11 | 0,70587 | 4,4 | 21,6 | 0,2019 | 0,122108 | 0,512473 | 11 | 0,51241 | -2,4 |
| R 01/00 | Andesite | 65 | 503 | 0,1288 | 0,3603 | 0,705627 | 7 | 0,70519 | 4,0 | 19,6 | 0,2031 | 0,122838 | 0,512616 | 8 | 0,51255 | 0,4 |
| E 53/99A | Andesite | 22 | 159 | 0,1357 | 0,3796 | 0,706307 | 11 | 0,70585 | 3,8 | 17,5 | 0,2153 | 0,130223 | 0,512528 | 14 | 0,51246 | -1,4 |
| E 53/99B | Rhyodacite | 73 | 241 | 0,3026 | 0,8468 | 0,706618 | 10 | 0,70560 | 1,9 | 11,4 | 0,1672 | 0,101135 | 0,512514 | 23 | 0,51246 | -1,4 |
| R 17/99 | Rhyodacite | 81 | 80 | 1,0174 | 2,8477 | 0,709138 | 12 | 0,70570 | 1,7 | 11,0 | 0,1576 | 0,095324 | 0,512497 | 23 | 0,51244 | -1,7 |
| VV 35/99 | Rhyodacite | 67 | 408 | 0,1643 | 0,4599 | 0,706728 | 8 | 0,70617 | 2,2 | 13,3 | 0,1646 | 0,099540 | 0,512567 | 8 | 0,51251 | -0,3 |
| R 19/99 | Rhyodacite | 16 | 438 | 0,0372 | 0,1042 | 0,706068 | 7 | 0,70594 | 2,2 | 15,0 | 0,1453 | 0,087842 | 0,512508 | 5 | 0,51246 | -1,4 |
| VV 36/99 | Subvolcanic rhyodacite | 64 | 238 | 0,2685 | 0,7515 | 0,707153 | 10 | 0,70625 | 2,5 | 14,7 | 0,1715 | 0,103734 | 0,512547 | 14 | 0,51249 | -0,8 |
| E 66/99 | Subvolcanic rhyodacite | 61 | 100 | 0,6055 | 1,6945 | 0,708129 | 8 | 0,70608 | 1,9 | 11,8 | 0,1628 | 0,098470 | 0,512575 | 27 | 0,51252 | -0,2 |
| E 16/97 | Elshitsa granite | 100 | 248 | 0,4016 | 1,1239 | 0,706123 | 8 | 0,70477 | 1,7 | 11,0 | 0,1552 | 0,093885 | 0,512558 | 9 | 0,51251 | -0,4 |
| E 17/94 | Elshitsa granite | 113 | 228 | 0,4966 | 1,3896 | 0,706365 | 9 | 0,70469 | | | | | | | | |
| VV 06/99 | Vlaykov Vruh granodiorite | 44 | 260 | 0,1692 | 0,4736 | 0,706325 | 12 | 0,70575 | | | | | | | | |
| VV 23/99 | Vlaykov Vruh granodiorite | 31 | 359 | 0,0871 | 0,2437 | 0,706094 | 9 | 0,70580 | 2,3 | 14,7 | 0,1565 | 0,094656 | 0,512498 | 6 | 0,51245 | -1,6 |
| R 16/99 | Red marl | 22 | 401 | 0,0539 | 0,1508 | 0,707844 | 9 | 0,70766 | | | | | | | | |
| R 15/99 | Senonian flysch (marl) | 18 | 436 | 0,0413 | 0,1155 | 0,707719 | 8 | 0,70758 | | | | | | | | |

* - 2SE represents two standard errors on the least significant digits on the analytical mean, based on within-run statistics.

Table 5 - Sr concentrations and isotopic compositions of hydrothermal calcite, barite and anhydrite from the Elshitsa, Radka and Vlaykov Vruh deposits

| Sample | Deposit | Mineral | Mineralisation stage | Sr (ppm) | $^{87}\text{Sr}/^{86}\text{Sr}$ | Error ($\pm 2\text{SE}$)* |
|-----------|--------------|-----------|----------------------|-------------|---------------------------------|--------------------------------|
| E 46/98 | Elshitsa | Calcite | stage-III | 642 | 0,706115 | 7 |
| E 39/99 | Elshitsa | Calcite | stage-III | 622 | 0,706165 | 13 |
| E 50/97 | Elshitsa | Barite | stage-III | 1076 | 0,706691 | 8 |
| E MPI404 | Elshitsa | Barite | stage-III | 1689 | 0,706874 | 6 |
| R 506 | Radka | Barite | stage-III | 1252 | 0,706565 | 7 |
| VV 39/99 | Vlaykov Vruh | Barite | stage-IV | 1582 | 0,705805 | 10 |
| R 561 | Radka | Anhydrite | stage-V | 1273 | 0,706835 | 8 |
| R 570/67 | Radka | Anhydrite | stage-V | 1640 | 0,707295 | 12 |
| R 29/98 | Radka | Anhydrite | stage-V | 1298 | 0,706241 | 6 |
| E 32/1967 | Elshitsa | Anhydrite | stage-V | 1131 | 0,706581 | 7 |
| E 49/97 | Elshitsa | Anhydrite | stage-V | 1063 | 0,706510 | 6 |
| E V1 | Elshitsa | Anhydrite | stage-V | 1078 | 0,706554 | 8 |
| E 13/98 | Elshitsa | Anhydrite | stage-V | 1456 | 0,707201 | 9 |
| E 17/98 | Elshitsa | Anhydrite | stage-V | 945 | 0,707050 | 7 |
| E 22/98 | Elshitsa | Anhydrite | stage-V | 1002 | 0,706305 | 7 |
| E 04/89 | Elshitsa | Anhydrite | stage-V | 1207 | 0,707093 | 10 |
| E 79/99 | Elshitsa | Anhydrite | stage-V | 1300 | 0,707087 | 8 |
| E 40/98 | Elshitsa | Gypsum | stage-V | 593 | 0,706940 | 6 |

* - 2SE represents two standard errors on the least significant digits on the analytical mean, based on within-run statistics.

Table 6 - U-Pb zircon isotope data for the Elshitsa granite and Elshitsa subvolcanic rhyodacite, and rutile data from the Vlaykov Vruh deposit

| N | meas. N° | size fraction, μm | weight, mg | Description | U, ppm | Pb, ppm | $^{206}\text{Pb}/^{204}\text{Pb}$ | $^{206}\text{Pb}/^{238}\text{U}$ | 2σ | $^{207}\text{Pb}/^{235}\text{U}$ | 2σ | $^{207}\text{Pb}/^{206}\text{Pb}$ | 2σ | $^{206}\text{Pb}/^{238}\text{U}$ apparent age | Rho |
|--|--------------------|------------------------------|------------|----------------------------|--------|---------|-----------------------------------|----------------------------------|-----------|----------------------------------|-----------|-----------------------------------|-----------|---|------|
| AvQ 029 (magmatic zircon - Elshitsa granite) | | | | | | | | | | | | | | | |
| 1 | IP29 | 75-100 | 0,0033 | »isometric | 121 | 3,74 | 70,3 | 0,01348 | 0,00008 | 0,089 | 0,005 | 0,048 | 0,002 | 86,28 | 0,52 |
| 2 | IP33 | 75-100 | 0,0109 | prism abr pale beige | 103 | 2,28 | 154,8 | 0,01356 | 0,00008 | 0,089 | 0,005 | 0,0477 | 0,0002 | 86,84 | 0,55 |
| 3 | IP86 | 75-100 | 0,0080 | prism abr pale beige | 62,7 | 1,15 | 284,1 | 0,01352 | 0,00015 | 0,090 | 0,002 | 0,0482 | 0,0011 | 86,59 | 0,52 |
| 4 | IP87 | 75-100 | 0,0020 | prism abr pale beige | 163 | 3,20 | 206,8 | 0,01355 | 0,00014 | 0,090 | 0,003 | 0,0483 | 0,0015 | 86,78 | 0,42 |
| 5 | IP89* | 75-100 | 0,0033 | prism pale beige | 189 | 3,61 | 215,1 | 0,01316 | 0,00008 | 0,089 | 0,004 | 0,049 | 0,002 | 84,30 | 0,51 |
| 6 | IP90* | 75-100 | 0,0051 | prism pale beige | 167 | 2,87 | 415,6 | 0,01318 | 0,00007 | 0,088 | 0,002 | 0,0484 | 0,0009 | 84,40 | 0,43 |
| AvQ 032 (magmatic zircon - Elshitsa subvolcanic rhyodacite) | | | | | | | | | | | | | | | |
| 7 | IP175 | 75-100 | 0,0102 | 3 long prism beige, abr | 114 | 1,90 | 559,6 | 0,01348 | 0,00006 | 0,0891 | 0,0008 | 0,0479 | 0,0003 | 86,32 | 0,62 |
| 8 | IP176 | 75-100 | 0,0103 | 4 long prism beige, abr | 97,2 | 1,62 | 542,8 | 0,01342 | 0,00006 | 0,0892 | 0,0009 | 0,0485 | 0,0003 | 85,91 | 0,66 |
| 9 | IP177 | 75-100 | 0,0154 | 4 long prism beige, abr | 99,1 | 1,78 | 356,0 | 0,01346 | 0,00008 | 0,089 | 0,002 | 0,0481 | 0,0012 | 86,16 | 0,43 |
| 10 | IP103* | 75-100 | 0,0045 | long prism beige | 118 | 2,89 | 107,4 | 0,01316 | 0,00008 | 0,089 | 0,006 | 0,049 | 0,004 | 84,31 | 0,62 |
| 11 | IP104* | 75-100 | 0,0090 | long prism beige | 118 | 2,56 | 133,0 | 0,01256 | 0,00008 | 0,082 | 0,005 | 0,048 | 0,003 | 80,43 | 0,56 |
| 12 | IP105 ⁱ | 75-100 | 0,0120 | isometric water clear | 94,2 | 9,57 | 1969,8 | 0,09653 | 0,00060 | 0,991 | 0,010 | 0,0744 | 0,0006 | 594,1 | 0,64 |
| 13 | 2573 ⁱ | 63-125 | 0,0085 | prism transp | 80,0 | 1,54 | 374,6 | 0,01375 | 0,00010 | 0,139 | 0,003 | 0,0733 | 0,0011 | 88,01 | 0,55 |
| 14 | 2574 ⁱ | 63-125 | 0,0049 | prism transp | 318 | 21,80 | 1181,8 | 0,06833 | 0,00034 | 0,529 | 0,003 | 0,0561 | 0,0001 | 426,1 | 0,89 |
| 15 | 2575* | 63-125 | 0,0058 | prism transp | 137 | 3,57 | 87,8 | 0,01207 | 0,00015 | 0,081 | 0,005 | 0,049 | 0,003 | 77,32 | 0,42 |
| VV 46 / 99 (hydrothermal rutile - Vlaykov Vruh porphyry Cu deposit) | | | | | | | | | | | | | | | |
| 16 | 2564 | 125-250 | 0,0101 | | 31,1 | 1,09 | 55,0 | 0,01337 | 0,00016 | 0,091 | 0,005 | 0,049 | 0,002 | 85,59 | 0,49 |
| 17 | 2565 | 125-250 | 0,0100 | | 28,7 | 0,64 | 97,1 | 0,01339 | 0,00028 | 0,084 | 0,004 | 0,0456 | 0,0009 | 85,72 | 0,52 |

Abbreviations : abr.- abraded; prism.- prismatic; transp. – transparent; Rho - correlation coefficient $^{206}\text{Pb}/^{238}\text{U} - ^{207}\text{Pb}/^{235}\text{U}$; * - sample with lead loss; ⁱ - zircons with inherited core
Errors (%): 2σ - for the two zircon samples (AvQ 029 and AvQ 032) and 1σ - for the rutile sample (VV 46 / 99)

Table 7 - Hf isotope data for zircons from the Elshitsa granite and Elshitsa subvolcanic rhyodacite (numbers are as in Table 6)

| N | meas. N° | $^{176}\text{Hf}/^{177}\text{Hf}$ | 2s error | e Hf today | e Hf (t=85 Ma) |
|---|--------------------|-----------------------------------|----------|------------|----------------|
| AvQ-029 (Elshitsa granite) | | | | | |
| 1 | IP33 | 0,282901 | 0,000006 | 4,6 | 6,2 |
| 2 | IP86 | 0,282951 | 0,000002 | 6,3 | 7,9 |
| 3 | IP87 | 0,282912 | 0,000011 | 5,0 | 6,6 |
| 4 | IP90* | 0,282917 | 0,000006 | 5,1 | 6,7 |
| AvQ-032 (Elshitsa subvolcanic rhyodacite) | | | | | |
| 5 | IP175 | 0,282947 | 0,000007 | 6,2 | 7,8 |
| 6 | IP176 | 0,282977 | 0,000019 | 7,2 | 8,9 |
| 7 | IP177 | 0,282929 | 0,000004 | 5,6 | 7,2 |
| 8 | IP104* | 0,282983 | 0,000005 | 7,5 | 9,1 |
| 9 | IP105 ⁱ | 0,281937 | 0,000004 | -30 | -28 |
| 10 | 2575* | 0,282997 | 0,000009 | 8,0 | 9,6 |

Legend: * - sample with lead loss; ⁱ - zircon with inherited core.