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Petrology of the Luingo caldera (SE margin of the Puna plateau): A middle Miocene window of the arc-back arc configuration

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ABSTRACT

We describe the petrographic characteristics, whole-rock geochemistry and mineral chemistry of rocks from the Pucarilla–Cerro Tipillas Volcanic Complex with emphasis on the rocks belonging to the middle Miocene Luingo caldera, located in the south-eastern portion of the Central Volcanic Zone (CVZ) of the Andes. We modelled the petrogenesis of the Luingo caldera rocks as a mixture of ca. 20% crustal magmas and 80% of mantle magmas by AFC with recharge processes. A comparison of Luingo geochemical data with the composition of Miocene–Pliocene volcanic rocks from the broad area, points to major thickening events during the middle Miocene for the western portion and during the upper Miocene for the eastern portion of the Southern CVZ. In the eastern sector (~66°W) the mantle source appears to change from a spinel–lherzolite type for the middle Miocene to a garnet–lherzolite type for the upper Miocene–Pliocene magmas. The areal distribution of the volcanic products led to the recognition of approximately equivalent areas covered by volcanic rocks both in the eastern and in the western Puna borders. This indicates a broad arc, which was structurally controlled at the proto–Puna/Puna margins, whose geochemical differences are related with variations in crustal thicknesses and heterogeneous mantle sources from west to east.

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1. Introduction

The Central Volcanic Zone (CVZ) of the Central Andes is developed between 14 and 28°S (Thorpe et al., 1984; de Silva and Francis, 1991) and is characterized by abundant Miocene to Recent ignimbrites, lavas and subvolcanic igneous bodies (e.g., Baker and Francis, 1978). The CVZ includes the Altiplano–Puna Volcanic Complex (APVC; de Silva, 1989) between 21 and 24°S, where large volumes of ignimbrites and abundant calderas were interpreted as the result of an "ignimbrite flare-up" (de Silva, 1989), bringing the region to the focus of various subsequent studies (e.g., de Silva, 1989; Ort et al., 1993; Lindsay et al., 2001a,b; Soler et al., 2007; Petrinovic et al., 2010). The APVC ignimbrites are predominantly crystal-rich calc-alkaline dacites (e.g., Harmon et al., 1984; de Silva and Francis, 1989; Soler et al., 2007). In contrast, the ignimbrites of the Southern CVZ (SCVZ, 25–27°S) have few detailed studies and a small number of recognized

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volcanic centres. In the western portion (between 67 and 69°W) they are characterized by small to medium volume (<10 km³), crystal-poor ignimbrites of felsic compositions (e.g., Siebel et al., 2001; Schnurr et al., 2007); whereas to the east, the Cerro Galán caldera (e.g., Sparks et al., 1985) and the recently identified Luingo caldera (Guzmán and Petrinovic, 2008) have larger volumes and crystal-rich dacitic to rhyodacitic compositions as most of the ignimbrites of the APVC.

Luingo is the oldest (13–12 Ma; Guzmán and Petrinovic, 2010) and south-easternmost caldera recognized so far in the SE Puna plateau margin, in the transition with the Pampean Ranges. In the present article we describe the petrography, mineral chemistry and geochemistry of the Luingo volcanic rocks in order to characterize their composition and magmatic evolution.

The location of the Luingo caldera, close to the well known Cerro Galán caldera, provides an excellent opportunity to investigate their possible genetic relationship as suggested earlier by González et al. (1999) and Hongn and Seggiaro (2001). Furthermore, the Luingo caldera becomes a key area to understand possible variations in magma source and composition before and after the major thickening event (ca. 10 Ma: Isacks, 1988) occurred in this sector of the Puna.

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Finally, we compare the Luingo rocks with coeval ignimbrites and lavas located to the west in order to identify the possible middle Miocene arc-back arc configuration of the Southern CVZ.

2. Geological background

The Altiplano–Puna plateau results from the convergence between the South-American and Nazca plates (Isacks, 1988; Allmendinger et al., 1997; Riller and Oncken, 2003). An outstanding feature of this plateau is its large crustal thickness (e.g., Yuan et al., 2000; 2002), which has been attributed to the combined effect of horizontal shortening (Allmendinger, 1986; Isacks, 1988; Allmendinger and Gubbels, 1996; Allmendinger et al., 1997), magmatic addition (Allmendinger, 1986; Marrett and Emerman, 1992; Allmendinger et al., 1997), lithospheric thinning and hydration of the upper mantle (e.g., Allmendinger et al., 1997), underplating of the Brazilian Shield below the Altiplano and Northern Puna during the late Miocene (e.g., Allmendinger et al., 1986; Gubbels et al., 1993; Allmendinger et al., 1997; Sempere et al., 1997; Riller and Oncken, 2003) and weather variations that led to differences in the sedimentation rates (e.g., Riller and Oncken, 2003; Strecker et al., 2007).

In the southern portion of the CVZ, a group of pyroclastic deposits and minor lava flows, located immediately to the SE of the Cerro Galán caldera, were grouped in the Pucarilla–Cerro Tipillas Volcanic Complex (PCTVC; Guzmán and Petrinovic, 2008). The complex includes the middle Miocene Luingo caldera, which is the main subject of the present study. The Luingo caldera is set at the boundary between the Puna and the Eastern Cordillera and Pampean Ranges, where the crustal thickness is about 60 km (e.g., Heit et al., 2007). Luingo caldera marks the eastern limit of the large-volume Cenozoic magmatism of the Altiplano–Puna plateau.

Most of the Luingo caldera deposits are located in the southern extension of the Colomé–Hualfín Valley (Fig. 1) which is bounded by regional faults with opposite vergence. The western fault thrusts basement rocks onto the Payogastilla Group, whereas the eastern fault (Jasimaná Fault: Marrett et al., 1994) reflects a Neogene inversion of a previous normal fault (e.g., Seggiaro et al., 2006; Carrera and Muñoz, 2008) that marks the western limit of the Cretaceous Salta rift (e.g., Hongn and Seggiaro, 2001; Mortimer et al., 2007) and the eastern limit of the PCTVC. Other faults farther to the west have trends similar to the Jasimaná fault also with left-lateral oblique components and westward vergence (Guzmán, 2009). Some of these faults were active before and others after the Luingo eruptive event (Guzmán, 2009).

Units from the PCTVC unconformably cover lower Miocene fluvial sediments of the basal and middle sections of the Angastaco Formation (Marrett et al., 1994; Guzmán, 2009). In other sections, the PCTVC covers the local Neoproterozoic–Palaeozoic basement, including medium to high grade metamorphites, such as the La Paya Formation (Aceñolaza et al., 1976), the Río Blanco Metamorphic Complex (e.g., Hongn and Seggiaro, 2001) and the Loma Corral Formation (Navarro García, 1984), as well as granitoids of the Oire Eruptive Complex (Blasco and Zapettini, 1996) and the Peñas Blancas Granite (Castillo, 1999).

2.1. Volcanic units related to the Luingo caldera

Products from the Luingo caldera are mostly welded and devitrified ignimbrites and minor lavas. Their units, simplified stratigraphic relationships and available ages are summarized in Fig. 2. For a detailed description and discrimination of different units the reader is referred to Guzmán and Petrinovic (2010).

Pyroclastic units from the Luingo caldera comprise the Pucarilla and Alto de Las Lagunas ignimbrites (outflow units) and the Luingo I and Luingo II ignimbrites (intracaldera units); the latter two are separated from each other by a collapse breccia, the Luingo Breccia. The PCTVC lavas are separated in Lavas I (related to the Luingo

caldera) and Lavas II (unrelated to the caldera). The Luingo caldera and related deposits are middle Miocene in age (Guzmán and Petrinovic, 2010; Fig. 2). New Laser-heating $^{40} \rm Ar/^{39} Ar$ analyses of two biotite separates from Lava II have been obtained at the $^{40} \rm Ar/^{39} Ar$ geochronology laboratory in the University of Potsdam, plateau ages yielded 7.59 \pm 0.03 and 7.60 \pm 0.02 Ma (see Appendix A for details). The bulk composition of the ignimbrites is dacitic, whereas Lavas I vary from trachyandesites to less common trachydacites and Lavas II are trachydacites.

Intracaldera facies (Luingo I and Luingo II ignimbrites) always show strong hydrothermal alteration, making it difficult to constrain the original composition and primary mineral association, thus major petrographic considerations were done in the outflow ignimbritic facies (Pucarilla and Alto de Las Lagunas ignimbrites) and lavas.

3. Petrography

The Pucarilla and Alto de Las Lagunas ignimbrites are pumice-rich (up to 40 vol.%), crystal-rich (between ca. 25 and 38 vol.% in pumices) and have low lithic contents (<3 vol.%). Three varieties (Fig. 3) of vesiculated fragments are recognized. The most common type (Fig. 3a, b) is white to black in hand specimen and grey to white in polarized light, the colour variations corresponding to different degrees of welding and devitrification. These vesiculated fragments are flattened to a variable extent as a function of the overall welding and compaction of the units. They are distinguished from the other two types by a generally coarser grain-size and by the common presence of clinopyroxene, both as phenocrysts and microphenocrysts and of titanite phenocrysts. Another pumice variety (<2%), is pink, typically flattened (Fig. 3c, d) and characterized by a large amount of broken and in some cases rounded crystals set in an aphanitic matrix where vesicles are overprinted by alteration and no longer recognizable; this type lacks titanite and only rarely contains clinopyroxene. A third variety of vesiculated clasts comprises grey crystal mush fragments (Fig. 3e, f), which are extremely crystal-rich (>90%), vesicle-poor and almost always sub-rounded, composed dominantly of plagioclase and biotite.

Both outflow ignimbrites (Alto de Las Lagunas and Pucarilla) contain plagioclase, biotite, magnetite, quartz, titanite, apatite and zircon; distinctively, the Pucarilla Ignimbrite also contains clinopyroxene whereas the Alto de Las Lagunas Ignimbrite has amphibole and scarce K-feldspar. All ignimbrites are fine- to medium-grained with an average phenocryst size of 2–3 mm occasionally reaching 6 mm. The Pucarilla Ignimbrite is divided in three units (Hualfin, Arremo and Jasimaná, from base to top) on the basis of variable devitrification and welding, and also of petrographic criteria, as detailed below.

The intracaldera ignimbrites (Luingo I and Luingo II) were strongly modified by post emplacement hydrothermal alteration. These are intensely welded ignimbrites in which flattened pumice fragments (up to 18 vol.%) are often difficult to recognize. They are crystal-rich (~35 vol.%), have low lithic contents (1–3 vol.%), and consist of variably altered plagioclase, clinopyroxene, biotite, quartz, apatite and titanite pseudomorphs with accessory Fe–Ti oxides and zircon and a variety of secondary minerals such as rutile, calcite, kaolinite, chlorite, sericite and epidote.

Lavas I are mostly trachyandesites with minor trachydacites. They are medium to fine grained and show porphyritic texture with a micro to cryptocrystalline intersertal and/or pilotaxic groundmass. Their mineral association consists of plagioclase, K-feldspar, clinopyroxene, with accessory apatite, Fe–Ti oxides \pm biotite and titanite. Lavas II are fine to medium grained porphyritic dacites whose groundmass is either intersertal or pilotaxic and varies from microcrystalline to cryptocrystalline. They are composed of biotite, amphibole, plagioclase, quartz, Fe–Ti oxides \pm clinopyroxene, orthopyroxene and titanite

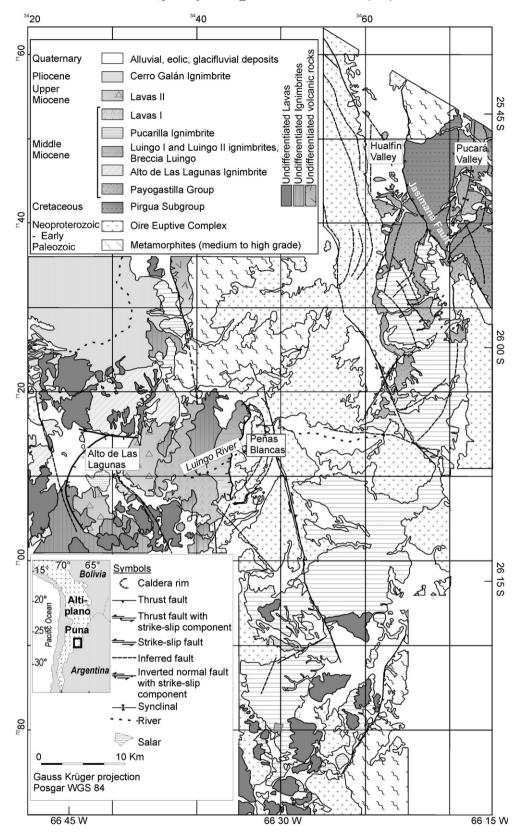


Fig. 1. Geologic map of the Pucarilla-Cerro Tipillas Volcanic Complex and the surrounding area. Modified from Guzmán and Petrinovic, 2010.

4. Mineral chemistry

Chemical analyses of selected minerals were performed by WDS with a Cameca SX-50 electron microprobe at the Institute of Geosciences, University of Brasilia, (see Appendix A for details). In

the particular case of the ignimbrites, analyses were carried out only on grains located within pumice fragments, to minimize the possibility of contamination with xenocrystic material. In order to gain insights into chemical variation trends during crystal growth, analytical profiles were performed in selected individual crystals.

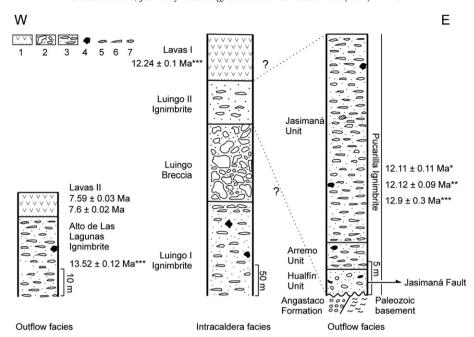


Fig. 2. Generalized sections of volcanic sequences from the PCTVC. 1) lavas, 2) breccias, 3) ignimbrites, 4) lithic fragments, 5) pink fiamme, 6) common fiamme/pumices, 7) grey crystal mush clasts. Ages are from this contribution and from *Marrett et al. (1994), **Folkes et al. (submitted for publication), *** Guzmán and Petrinovic, 2010. Modified from Guzmán and Petrinovic, 2010.

Representative chemical analyses of each mineral phase are shown in Tables 1–5, whereas the whole data set is represented in Figs. 4–8.

4.1. Pyroxene

Clinopyroxenes occur in the Pucarilla Ignimbrite and in the Lavas I and Lavas II units, but orthopyroxene is restricted to Lavas II. Representative analyses are given in Table 1. For classification and other diagrams, the cationic proportions were calculated on the basis of 6 oxygen and Fe_2O_3 was estimated according to the method of Droop (1987).

Clinopyroxene from the Pucarilla Ignimbrite occurs as 0.2 to 1.85 mm-long, subhedral, prismatic, often broken phenocrysts, locally forming glomerophyric aggregates. Many crystals are optically zoned, especially near the rims, which are often coated in Fe–Ti oxides. Common inclusions comprise apatite, with subordinate plagioclase, Fe–Ti oxides, and biotite.

In the Morimoto (1988) quadrilateral classification diagram (Fig. 4) they plot mostly as diopside ($Wo_{48-44} En_{41-34} Fs_{18-14}$). A single analysis (out of 93), belonging to a Jasimaná Unit pumice, plots in the augite field, with a minor calcic augite component. The fields for different ignimbrite and pumice types overlap widely, and there are no significant compositional differences between them. Data from single-crystal microprobe profiles reveal only narrow chemical zoning.

The pyroxene in the Lavas I Unit is calcic augite (Wo₄₄₋₃₉ En₄₄₋₄₀ Fs₁₈₋₁₃). Fig. 4 shows the compositional range, compared with the field of the Pucarilla Ignimbrite and the Lavas II pyroxenes. Analyzed grains from samples near the base of Lavas I Unit show normal zoning, whereas in those samples collected towards the top the zoning is both normal and reverse. Lavas II are the only unit where both clino and orthopyroxene are present (Fig. 4). The clinopyroxene is diopside (Wo_{45.14 - 46.46}, En_{38.53 - 40.94}, Fs_{13.27 - 15.55}), it is almost unzoned, whereas the orthopyroxene shows reverse zoning and has a wide compositional variation (Wo_{1.67-3.27} En_{60.65-79.95} Fs_{17.82-37.67}).

4.2. Amphibole

Amphibole from the Alto de Las Lagunas Ignimbrite occurs as 0.05 to 2 mm-long crystals, they are subhedral to euhedral with tabular

habit and are often coated in Fe–Ti oxides even becoming pseudomorphs. Lavas II amphiboles (0.1 to 1.25 mm) are subhedral to anhedral and frequently form glomerophyric aggregates.

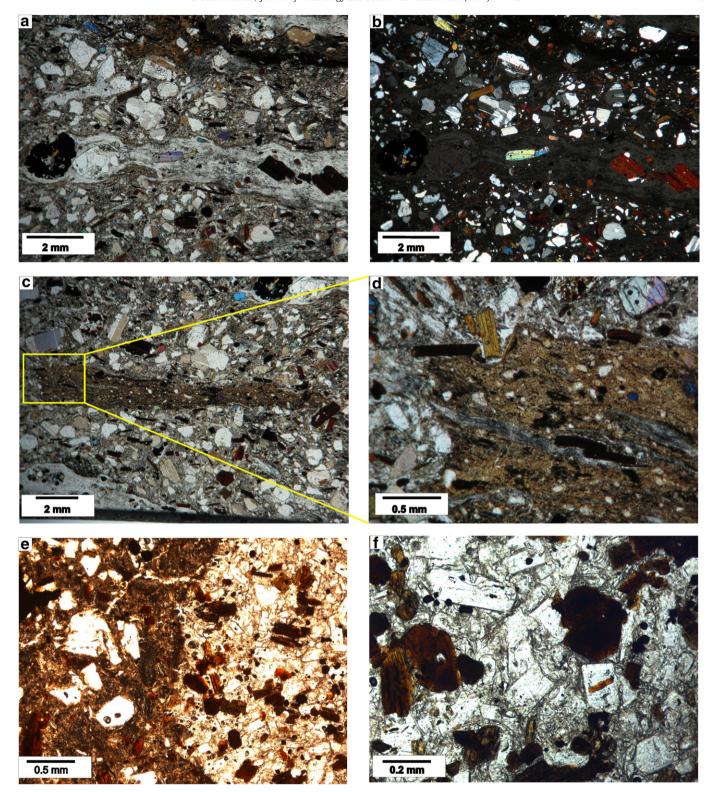
For classification and other diagrams, cations were calculated on the basis of 23 O, and 2 (OH, F, Cl) groups. Fe³⁺ was calculated as the average between minimum and maximum values as suggested by Leake et al. (1997). Since $(Ca+Na)_B \ge 1$ and $Na_B \le 0.5$ for all analyses, they are classified as calcic amphiboles. Two subgroups were recognized on the basis of the parameters (a) $Ca_B \ge 1.5$, $(Na+K)_A \ge 0.5$, $Ti \le 0.5$ and (b) $Ca_B \ge 1.5$, $(Na+K)_A \le 0.5$, $Ca_A \le 0.5$, and classified accordingly in the diagrams of Fig. 5(a) and (b) (Leake et al., 1997). Representative analyses are given in Table 2.

Amphiboles from the Alto de Las Lagunas Ignimbrite vary compositionally between alumino-edenite and alumino-magnesio-hornblende. Crystals from the Lavas II Unit, on the other hand, show a wider composition range, encompassing alumino-edenite, alumino-pargasite, alumino-magnesiohornblende, and alumino-tschermakite.

Microprobe profiles in individual grains indicate that zoning in amphibole from the Alto de Las Lagunas Ignimbrite is very subtle and may be normal or reverse. In contrast, amphiboles from the Lavas II typically show reverse zoning, the rims being more Mg-rich and Aland Ti-poor than the cores.

4.3. Biotite

Biotite (<0.05–2.95 mm) is subhedral with tabular to acicular habit; some crystals are bent and others show domino structure. It contains inclusions of iron–titanium oxides, apatite, plagioclase and zircon. Biotite crystals are often coated or even replaced by opaques due to vapour-phase alteration. Representative biotite analyses are given in Table 3. Chemical classification of the analyzed micas followed the recommendations in Tischendorf et al. (1997) and Yavutz (2001). LiO₂ content was estimated through the equation Li₂O wt.% = 155*(MgO)^{-3.1}, for SiO₂>34 wt.% and MgO>6 wt.%. All analyzed grains classify as Mg-biotite (Fig. 6), although it should be stressed that except for a few analyses, the micas in the Lavas I and Lavas II units are systematically richer in MgO (wt.%) than those in the ignimbrites. It is important to note that the presence of these MgO-rich biotites in the Pucarilla Ignimbrite may represent crystals that



 $\textbf{Fig. 3.} \ \, \textbf{Microphotographs of Luingo caldera's juvenile fragments. a, b) "common" pumice clasts, c, d) pink pumice type, in the detail photograph (d) it is seen that although the main fragment is of pink pumice, it is banded with the "common" type pumice; e, f) grey crystal mush clast to the right (e), and detail of that fragment (f). \\$

belong to a magma batch of a different composition than the main one, as discussed latter in this paper.

Biotite from the pyroclastic sequences has high titanium content (TiO₂ wt.%: 4.12–5.13) similar to many other Puna ignimbrites (e.g., Vilama: Soler, 2005; Atana: Lindsay et al., 2001a) indicating relatively high pre-eruptive temperatures (c.f., Patiño Douce, 1993). TiO₂

content of biotite in the Lavas I Unit is even higher (5.39-5.82 wt.%), whereas biotite in Lavas II has TiO_2 contents in a range similar to the Pucarilla Ignimbrite. Biotites from the Alto de Las Lagunas Ignimbrite show the lowest TiO_2 content (4.12-4.53 wt.%), suggesting that these ignimbrites have the lowest pre-eruptive temperatures within the Luingo caldera ignimbrites. This is also in accordance with a typically

 Table 1

 Representative microprobe analyses of pyroxenes.

Unit	Pucarill	ucarilla Ignimbrite	rite														avas I			П	Lavas II				
	Hualfîn Unit	Unit				Arremo Unit	Unit				Jasimaná Unit	Unit													
Analysis	PAMP- 06-6-1	PAMP- 06-6-2	PAMP- 06-6-3	PAMP- 06-8-2	PAMP- 06-8.3	N02- 1-B5	N02- 1-B6	N02- 3-B1	N02- 3-B2	SN02- 3-B3	D04- 1B-1	D04- I	D04- 1	PAMP- F 01-1-B1 (PAMP- 01-1-B2	PAMP- (01-1-B3	CMS21- (CMS21- C	CMS18- (CMS18- (CMS30- 1-1	CMS30- 1-2	CMS30- 2-B1	CMS30- 2-B2	CMS- 30-2-3B
Position	Rim		_	Core	Rim	Core	Rim	Rim	Core	Rim	Rim	_	Rim	Rim (Core	Rim		_			Sim	Core	Rim	Core	Rim
	Pumice		_			Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	_	Pumice	Pumice	Pumice	Pumice									
Profile	15.1		15.3	16.2	16.3	33.2	33.3	31.1	31.2	31.3	4.1			20.1	20.2	20.3	39.1	. ,	.,		13.1	43.2	17.1	17.2	17.3
SiO ₂	52.73		52.18	53.35	53.88	51.89	52.46	52.94	52.48	52.07	52.69			52.54	52.37	51.95	52.09	Ì	•		52.57	49.19	52.14	52.22	52.55
TiO ₂	0.17		0.21	0.10	0.14	0.13	0.12	0.12	0.16	0.14	0.12			0.14	0.14	0.13	0.26				0.17	0.34	0.22	0.15	0.17
Al_2O_3	1.18	99.0	1.56	0.94	1.01	0.99	0.77	0.82	06.0	0.81	1.07	0.95	0.84	0.94	86.0	0.88	1.60	1.54	3.77	3.15	2.29	7.02	1.25	0.89	1.10
FeO	9.33		9.17	8.38	8.32	8.61	8.24	8.33	8.95	8.07	8.82	_		8.77	9.23	8.15	9.39				12.46	18.53	9.18	8.64	8.92
Fe_2O_3	0.01		0.35	0.16	0.35	0.00	0.00	0.00	0.01	0.00	90.0			0.00	0.00	0.00	0.00			_	0.01	0.01	0.00	0.00	0.00
Cr_2O_3	0.04		0.00	0.04	0.00	0.03	0.02	0.01	00.00	0.02	0.03	_		0.00	0.01	0.00	0.02				0.76	0.16	0.02	0.00	0.02
MnO	0.82		0.91	0.91	06'0	0.84	0.79	0.65	0.84	0.89	86.0			92.0	0.88	0.84	0.63				0.18	0.33	0.64	92.0	0.63
Nio	0.00		0.00	0.03	0.00	0.07	0.01	0.04	0.01	0.04	0.02	_	_	0.00	0.00	60.0	0.00	_			0.03	0.00	0.03	0.00	0.00
MgO	13.86		12.28	13.41	13.63	13.49	14.15	13.84	14.66	14.13	13.47	_		14.16	13.56	14.36	15.39				28.41	22.74	13.85	14.35	14.18
Ca0	22.01		22.09	23.05	22.44	23.06	22.97	22.91	22.36	22.79	22.49	_		22.08	22.18	22.46	20.87				1.19	1.57	22.38	22.46	22.42
Na_2O	0.00		0.14	90.0	0.14	0.04	0.37	0.29	0.45	0.04	0.02			0.39	0.74	0.13	0.13				80.0	0.25	0.00	0.00	0.04
K_2O	0.04		0.03	0.03	0.01	0.00	0.01	0.00	0.01	0.00	0.03			0.02	0.01	0.02	0.01				0.01	0.00	0.02	0.02	0.00
Total	100.17	-	98.92	100.45	100.83	99.14	99.91	99.95	100.83	00.66	08'66	_		82.66	100.10	99.01	100.39		•	-	98.17	100.12	99.72	99.48	100.03

 Table 2

 Representative microprobe analyses of amphiboles.

Unit	Alto de I	Las Laguna	as Ignimbri	te	Lavas II		
Analysis	SCG4- 4-A7	SCG4- 4-A8	CMS-38- 1-F1	CMS-38- 1-F2	CMS28- 1-A6	CMS28- 1-A5	CMS28- 1-A4
	Pumice	Pumice	Pumice	Pumice			
Position	Core	Rim	Rim	Core	Rim	Core	Rim
SiO ₂	46.06	46.09	45.25	45.21	46.04	42.78	47.13
TiO ₂	1.35	1.33	1.43	1.50	1.91	2.87	1.66
Al_2O_3	8.22	8.02	8.59	8.91	8.47	11.39	8.08
Fe_2O_3	2.61	4.55	4.24	2.33	4.14	3.90	5.37
Cr_2O_3	0.04	0.02	0.00	0.02	0.01	0.00	0.06
FeO	12.50	11.25	12.31	14.03	8.93	8.89	7.94
MnO	0.70	0.67	0.76	0.73	0.31	0.15	0.34
ZnO	0.00	0.00	0.00	0.00	0.00	0.00	0.00
MgO	13.08	12.94	12.83	11.97	14.98	14.02	15.54
CaO	11.63	11.51	11.35	11.45	11.34	11.34	11.32
Na_2O	1.78	1.09	1.92	2.06	1.65	1.88	1.48
K_2O	0.96	1.02	1.148	1.084	0.88	1.06	0.78
Total	98.91	98.49	99.83	99.28	98.66	98.29	99.69

higher temperature association (biotite + pyroxene, with no amphibole) in the Pucarilla Ignimbrite than in the Alto de Las Lagunas Ignimbrite (amphibole + K feldspar) (c.f., Wones, 1970). Zoning in biotite is generally very subtle and in most cases it is normal.

4.4. Plagioclase

Plagioclase occurs in all studied units. In the Pucarilla Ignimbrite it often shows sieve texture, whereas embayments and corroded rims are rare. It also contains abundant inclusions, preferably located in zoning limits. Among these, apatite is conspicuous, biotite, zircon, and Fe–Ti oxides are less common and clinopyroxene is rare. Plagioclase from Luingo I and Luingo II ignimbrites is strongly altered to calcite and sericite. Plagioclase from the Lavas I and Lavas II units commonly shows corroded rims or intra-crystal disequilibrium features and sieve textures. Additionally, in Lavas I crystal growth under disequilibrium is apparent from plagioclase step zoning patterns.

Representative analyses are given in Table 4 and their classification in the Ab–Or–An diagram is shown in Fig. 7. Overall plagioclase composition in the outflow pyroclastic units varies from andesine to bytownite. In the intracaldera (Luingo I and Luingo II) ignimbrites plagioclase ranges from albite to oligoclase due to subsolidus alteration. These do not represent primary plagioclase, and will not be discussed here. Chemical zoning is common in plagioclase from all units, varying in intensity from prominent in the Arremo and Hualfin units to subtle in the Alto de Las Lagunas Ignimbrite.

Plagioclase from the Alto de Las Lagunas Ignimbrite has a composition restricted to andesine (An_{45-31}), does not show optical disequilibrium textures but shows zoning, both normal and reverse and locally oscillatory. Plagioclase from the Hualfin Unit varies from An_{81} to An_{33} , is typically zoned towards lower An contents at the rims, with some detailed microprobe profiles suggesting the presence of step zoning. A grey crystal mush found in the Hualfin Unit and interpreted as an accidental lithic (although still related to Luingo caldera), contains two groups of plagioclase microlites: one varies from An_{53} to An_{35} , but the other displays a much wider range (An_{74-43}) even within a single crystal. In the Arremo Unit, plagioclase varies from An_{84} to An_{32} . Zoning patterns vary from normal, reverse, through oscillatory with strong evidence of mafic magma recharging (Fig. 8).

Plagioclase from the Jasimaná Unit ranges from An_{64} to An_{29} . Specifically plagioclase in pink pumices has a more restricted range (An_{46-33}) . Zoning patterns are both reverse and normal. Nevertheless, it is clear that the highest An % contents in the Jasimaná Unit plagioclase are found in the cores of normally-zoned crystals.

Lable 3 Representative microprobe analyses of micas.

Paisimaná Unit Paisimaná Unit Paisimaná Unit	Unit	Pucarilla Ignimbrite	imbrite							Alto de Las	Alto de Las Lagunas Ignimbrite	mbrite	Lavas I			Lavas II	
sist Pony-1-D1 Pony-1-D1 Pony-1-D2 Doy-1-D2		Hualfîn Unit		Arremo Uni	it		Jasimaná Uı	nit									
on Pumice	Analysis	P007-1-D1	P007-1-D2	D07-1-D1	D07-1-D2	D07-1-D3	D04-1-D1	D04-1-D2	D04-1-DE	CG4-4-D1	CG4-4-D2	CG4-4-D3	CMS21-3-D1	CMS21-3-D2	CMS21-3-D3	CMS28-1-D5	CMS-281-D6
on Rim Core Rim R		Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice					
36.79 36.58 37.79 37.43 37.73 36.58 37.45 37.13 36.00 36.99 36.31 36.31 36.00 36.99 36.31 36.31 36.00 440 444 444 464 481 458 453 440 444 444 444 464 481 458 453 440 444 444 444 464 481 458 453 440 444	Position	Rim	Core	Rim	Core	Rim	Rim	Core	Rim	Rim	Core	Rim	Rim	Core	Rim	Core	Rim
478 4.69 4.43 4.62 4.54 4.64 4.81 4.58 4.53 4.40 4.44 13.35 13.31 13.20 13.54 13.49 13.60 13.52 13.47 13.91 13.62 13.86 16.44 16.49 15.96 16.07 15.58 16.39 16.02 16.15 18.22 17.79 17.49 16.44 16.49 15.96 16.07 15.58 16.39 16.02 16.15 18.22 17.79 17.49 0.48 0.78 0.00 0.01 0.01 0.02 0.04 0.00 0	SiO ₂	36.79	36.58	37.79	37.43	37.73	36.58	37.45	37.13	36.00	36.69	36.31	36.88	36.48	37.05	37.28	37.39
13.35 13.31 13.20 13.54 13.60 13.52 13.47 13.91 13.62 13.86 16.44 16.49 15.96 16.07 15.88 16.39 16.02 16.15 18.22 17.79 17.49 0.48 0.78 0.00 0.04 0.10 0.06 0.06 0.06 0.00 0.00 0.32 0.32 0.27 0.27 0.23 0.26 0.41 0.28 0.36 0.00	TiO_2	4.78	4.69	4.43	4.62	4.54	4.64	4.81	4.58	4.53	4.40	4.44	5.48	5.47	5.49	4.75	4.59
16.44 16.49 15.96 16.07 15.58 16.39 16.02 16.15 18.22 17.79 17.49 0.48 0.78 0.00 0.47 0.00 0.10 0.06 0.00 0.26 0.20 0.00 0.32 0.27 0.22 0.44 0.23 0.36 0.41 0.28 0.36 13.55 13.50 13.75 14.70 13.63 13.77 14.03 13.75 12.59 13.24 12.87 0.00	Al_2O_3	13.35	13.31	13.20	13.54	13.49	13.60	13.52	13.47	13.91	13.62	13.86	13.65	13.61	13.72	13.92	13.71
0.48 0.78 0.00 0.47 0.00 0.10 0.06 0.00 0.26 0.20 0.00 0.32 0.32 0.27 0.22 0.44 0.23 0.30 0.26 0.41 0.28 0.30 13.55 13.50 13.75 14.70 13.63 13.77 14.03 13.75 12.59 13.24 12.87 0.00 <	FeO	16.44	16.49	15.96	16.07	15.58	16.39	16.02	16.15	18.22	17.79	17.49	11.88	15.07	13.53	13.74	13.30
0.32 0.23 0.23 0.23 0.26 0.41 0.28 0.36 13.55 13.50 13.75 14.70 13.63 13.77 14.03 13.75 12.59 13.24 12.87 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.10 0.03 0.14 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.46 0.71 0.08 0.84 0.00 0.09 0.21 0.56 0.17 0.02 9.29 9.57 9.12 9.74 8.53 9.72 9.70 9.87 9.59 9.72 9.75 0.20 0.18 0.24 0.17 0.18 0.18 0.18 0.19 0.13 0.13 3.92 3.93 3.90 3.93 3.90 3.93 3.90 3.93 3.90 9.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00	Fe_2O_3	0.48	0.78	0.00	0.47	0.00	0.10	90.0	0.00	0.26	0.20	0.00	1.01	0.94	0.70	0.51	0.36
13.55 13.50 13.75 14.03 13.75 14.03 13.75 12.59 13.24 12.87 0.00	MnO	0.32	0.32	0.27	0.22	0.44	0.23	0.30	0.26	0.41	0.28	0.36	0.12	0.23	0.16	0.09	0.12
0.00 0.00 <th< td=""><td>MgO</td><td>13.55</td><td>13.50</td><td>13.75</td><td>14.70</td><td>13.63</td><td>13.77</td><td>14.03</td><td>13.75</td><td>12.59</td><td>13.24</td><td>12.87</td><td>16.18</td><td>14.38</td><td>15.54</td><td>16.02</td><td>16.22</td></th<>	MgO	13.55	13.50	13.75	14.70	13.63	13.77	14.03	13.75	12.59	13.24	12.87	16.18	14.38	15.54	16.02	16.22
0.10 0.03 0.14 0.00 0.08 0.00 0.01 0.00 <th< td=""><td>BaO</td><td>0.00</td><td>0.00</td><td>0.00</td><td>0.00</td><td>0.00</td><td>0.00</td><td>0.05</td><td>0.00</td><td>0.03</td><td>0.00</td><td>0.00</td><td>0.15</td><td>0.38</td><td>0.00</td><td>0.09</td><td>0.18</td></th<>	BaO	0.00	0.00	0.00	0.00	0.00	0.00	0.05	0.00	0.03	0.00	0.00	0.15	0.38	0.00	0.09	0.18
0.46 0.71 0.08 0.08 0.84 0.00 0.09 0.21 0.56 0.17 0.00 9.29 9.57 9.12 9.54 8.53 9.72 9.70 9.87 9.59 9.72 9.75 0.20 0.18 0.24 0.17 0.21 0.18 0.18 0.00 0.17 0.15 0.13 3.92 3.93 3.91 3.93 3.90 3.93 3.90 3.93 3.90 0.00	Ca0	0.10	0.03	0.14	0.00	0.08	0.00	0.00	0.00	0.00	0.00	0.02	0.01	0.00	0.00	0.00	0.01
9.29 9.57 9.12 9.54 8.53 9.72 9.70 9.87 9.59 9.72 9.75 0.20 0.18 0.24 0.17 0.21 0.18 0.18 0.20 0.17 0.15 0.13 3.92 3.93 3.91 3.93 3.90 3.96 3.92 3.90 3.93 3.90 0.00 <td>Na_2O</td> <td>0.46</td> <td>0.71</td> <td>80.0</td> <td>80.0</td> <td>0.84</td> <td>0.00</td> <td>0.09</td> <td>0.21</td> <td>0.56</td> <td>0.17</td> <td>0.00</td> <td>09.0</td> <td>0.18</td> <td>0.09</td> <td>0.80</td> <td>0.59</td>	Na_2O	0.46	0.71	80.0	80.0	0.84	0.00	0.09	0.21	0.56	0.17	0.00	09.0	0.18	0.09	0.80	0.59
0.20 0.18 0.24 0.17 0.21 0.18 0.18 0.20 0.17 0.15 0.13 3.92 3.93 3.94 3.93 3.90 3.96 3.96 3.93 3.90 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.66 8 100.84 98.88 100.84 98.99 94.11 10.016 99.56 100.15 100.19 99.17 99.18 99.17	K_2O	9.29	9.57	9.12	9.54	8.53	9.72	9.70	9.87	9.59	9.72	9.75	9.46	9.18	9.33	8.99	9.02
3.92 3.93 3.91 3.99 3.93 3.90 3.96 3.92 3.90 3.93 3.90 0.00 0.00 0.00 0.00 0.00	C	0.20	0.18	0.24	0.17	0.21	0.18	0.18	0.20	0.17	0.15	0.13	0.34	0.33	0.41	0.23	0.25
0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.0	H_2O	3.92	3.93	3.91	3.99	3.93	3.90	3.96	3.92	3.90	3.93	3.90	3.95	3.92	3.93	4.00	3.97
99 68 100 18 98 88 100 84 98 99 9911 100 16 99 56 100 16 100 19 99 12	SrO	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
21.00 01.001 01.001 01.001 11.00 01.001 00.001 00.001	Total	89.66	100.08	98.88	100.84	66.86	99.11	100.16	99.56	100.16	100.19	99.12	99.72	100.17	96.66	100.40	99.71

Overall, An % content is distinctively higher in plagioclases of the Hualfin and Arremo units, in comparison with the Jasimaná Unit and the Alto de Las Lagunas Ignimbrite. It is noteworthy that some of the anorthite contents (up to 84 mol%) are significantly high for an overall dacitic system. It is possible that those Ca-rich plagioclases are related to recharge of the chamber with higher-temperature magmas and/or variations in pressure and volatile content (e.g., Streck, 2008).

In both outflow units of the Luingo caldera, there is a tendency for plagioclase cores from the pink pumices to have lower and more restricted (andesine) An contents than those in the grey crystal mush fragments or in the more abundant, common-type pumices.

Plagioclase from Lavas I varies from andesine to labradorite (An_{64-26}), exceptionally bytownite (An_{74}). Zoning is mostly toward An decrease from core to rim, although one analyzed crystal has An_{39} core then jumps to An_{76} and finally yields An_{38} to the rims; this could reflect corrosion in some part of the crystal growth history due to magma chamber recharge (Fig. 8). Plagioclase from Lavas II ranges from andesine to labradorite (An_{62-33}) with both direct and reverse zoning.

4.5. Fe-Ti oxides

Spinel-group minerals are abundant both as inclusions in clinopyroxene, plagioclase, titanite and biotite, and as a matrix constituent. Representative analyses are given in Table 5. Ilmenite was not found during petrography and microprobe work.

The Fe–Ti oxides analyzed belong to the magnetite-ulvospinel series, with small amounts of MgO (up to 1.5 wt.%, rarely 2.43 wt.%), Al_2O_3 (up to 3.1 wt.%, exceptionally 6.53 wt.%), V_2O_3 (up to 0.64 wt.%), and very low Cr_2O_3 (<0.1 wt.%). TiO_2 , on the other hand, varies widely (generally up to 13.11 wt.%, and exceptionally up to 25.6 wt.%). In most cases the highest TiO_2 (wt.%) content in spinels from a single sample or unit are from spinel inclusions in mafic minerals. The large majority of the analyzed crystals contain over 3 wt.% TiO_2 , and are therefore classified as Ti-magnetite. Increasing TiO_2 (wt.%) contents in titanomagnetite are generally correlated with increasing temperature (e.g., Devine et al., 2003), whereas other oxides, such as MgO and Al_2O_3 may be strongly influenced by co-precipitating phases, such as mafic minerals and plagioclase (e.g., Tomiya and Takahashi, 2005).

The ranges of spinel composition from the various units overlap, but those from Arremo (Usp₃₄₋₂) and Lavas I (Usp₇₆₋₁) are much wider than those from Hualfin (Usp₁₉₋₁₄) and Jasimaná (Usp₂₁₋₃). A single grain analyzed from Alto de Las Lagunas yielded Usp₈, and one from the Lavas II Unit yielded Usp₁₀. The wider range observed in Arremo and Lavas I in individual magma batches is consistent with magma mixing or thermal/compositional zonation in the chamber (c.f., Tomiya and Takahashi, 2005).

5. Whole-rock chemistry

Major and minor elements were determined by XRF at the Universidad Nacional de Salta, rare earth elements and other minor and trace elements were analyzed at the ACME laboratories (Canada) by ICP-MS. Isotopic ratios were determined at the Geochronology Laboratory of the Universidade de Brasília. Analytical methodology is explained in Appendix A.

Absolute values of major, minor and rare earth elements and isotopic ratios are given in Table 6. The ignimbrites composition (on a volatile free basis) is restricted to calc-alkaline dacites and trachydacites (Fig. 9) whereas Lavas I are trachyandesites and minor trachydacites and Lavas II dacites (see Fig. 9). The ignimbrites classify as metaluminous to mildly peraluminous (A/CNK>1), with values ranging from 0.92 to 1.1, their alkalis content is high ($K_2O+Na_2O>6\%$) as it is seen in most of the ignimbrites from the CVZ; both groups of lavas are metaluminous. Correlation diagrams (Fig. 10) for some major and minor elements show tendencies compatible with fractionation of different mineral phases as evidenced by CaO vs Al_2O_3

Table 4Representative microprobe analyses of feldspars.

Unit	Alto de Las La	agunas Ignimbri	ite		Pucarilla Ign	imbrite											
					Accidental li in Hualfin U		Hualfín Un	it				Arremo U	nit				
Analysis	CMS-38-4-1	CMS-38-4-2	CG4-1-2	CG4-1-3	H006B-1-2	H006B-1-3	H006B-3-1	H006B-3-3	P007-1-1	P007-1-2	P007-1-3	N02-4-1	N02-4-2	N02-4-2.5	N02-4-3.5	N02-4-4	N02-4-5
Profile	66.1	66.2	60.2	60.3	92.1	92.2	94.1	94.3	89.1	89.2	89.3	46.1	46.2	46.3	46.4	46.5	46.6
	Pumice	Pumice	Gcm ¹	Gcm ¹	Gcm ¹	Gcm ¹	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice	Pumice
Position	Rim	Core	Core	Rim	Core	Rim	Rim	Core	Rim	Core	Rim	Rim					Rim
SiO ₂	57.76	59.70	57.90	57.31	57.12	53.97	57.42	57.14	58.14	58.56	57.79	58.02	47.16	59.36	58.23	51.78	57.78
Al_2O_3	26.20	25.01	25.64	25.84	26.31	28.60	25.76	25.94	25.45	25.85	25.08	25.95	32.90	25.51	26.34	29.38	26.28
FeO	0.20	0.27	0.23	0.23	0.46	0.37	0.22	0.22	0.24	0.21	0.25	0.26	0.36	0.24	0.23	0.30	0.31
CaO	8.43	7.27	7.83	8.40	8.79	11.31	8.23	8.60	8.01	8.06	7.52	8.26	16.37	7.76	8.50	12.83	8.50
Na_2O	6.78	8.41	7.40	7.18	6.16	5.38	6.62	6.49	6.69	7.01	7.91	7.26	2.77	6.21	6.27	4.24	7.25
K_2O	0.52	0.73	0.68	0.67	0.65	0.36	0.72	0.62	0.82	0.83	0.83	0.68	0.13	0.79	0.69	0.28	0.67
BaO	0.06	0.04	0.00	0.00	0.00	0.09	0.00	0.09	0.00	0.10	0.03	0.01	0.00	0.00	0.00	0.15	0.09
SrO	0.00	0.00	0.00	0.02	0.16	0.00	0.00	0.02	0.05	0.10	0.07	0.01	0.25	0.03	0.00	0.00	0.13
Total	99.93	101.42	99.68	99.64	99.65	100.08	98.96	99.13	99.38	100.72	99.48	100.44	99.92	99.90	100.26	98.96	101.00
Unit	Jasimaná Ur	nit–Pucarilla Igr	imbrite				Lavas I					Lavas I	I				
Analysis	PAMP01- 1-1	PAMP01- 1-2	PAMP01- 1-3	P002B- 1-7	P002B- 1-8	P002B- 1-9		CMS- CM 8-2-2 18-				CMS- 30-5-4	CMS- 30-5-5	CMS- 30-5-6	CMS- 30-5-1	CMS- 30-5-2	CMS- 30-5-4A
Profile	56.1	56.2	56.3	54.1	54.2	54.3	68.1	68.2	68.4	68.5	68.6	75.1	75.2	75.3	74.1	74.2	74.4
	Pumice	Pumice	Pumice	Pinkp ²	Pinkp ²	Pinkp ²											
Position	Rim	Core	Rim	Rim	Core	Rim	Rim		Core	_	Rim	Rim	Core	Rim	Rim	Core	Rim
SiO ₂	57.76	58.73	58.61	57.85	57.77	58.29	57.16	55.46 55.	46 58.	66 53.83	56.71	58.98	55.25	58.32	52.39	57.77	52.14
Al_2O_3	25.72	24.98	25.83	26.38	27.15	26.47		27.53 26.				25.68		26.09	29.65	26.24	29.63
FeO	0.27	0.25	0.24	0.28	0.25	0.24			46 0.4			0.30		0.30	0.85	0.29	0.62
CaO	8.03	6.99	7.87	8.54	9.06	8.42		0.41 9.				7.85		8.30	12.79	8.41	13.08
Na ₂ O	7.55	8.33	6.83	6.69	6.09	6.99		5.24 4.			5.38	7.08		6.88	4.48	7.09	4.26
K_2O	0.61	0.87	0.74	0.39	0.61	0.68		0.80 0.				0.87		0.85	0.28	0.64	0.32
BaO	0.07	0.13	0.00	0.00	0.00	0.00		0.13 0.				0.04		0.08	0.00	0.15	0.13
SrO	0.00	0.00	0.19	0.11	0.03	0.16	0.15	0.01 0.	16 0.	12 0.12	0.00	0.03	0.02	0.09	0.00	0.00	0.09
Total	100.00	100.26	100.30	100.22	100.97	101.25	99.52	99.96 98.	14 100.	19 98.69	99.59	100.82	98.35	100.91	100.45	100.59	100.27

¹Grey crystal mush. ²pink pumice.

Table 5Representative microprobe analyses of spinels.

Unit	Pucarilla Ig	gnimbrite									Lavas I		Lavas II
	Hualfin Un	it		Arremo Uni	t			Jasimaná	Unit				
Analysis	D-02-10- A1	D-02- 7B-2	D-02- 4B-4	PAMP-02- 3B-1	PAMP-02- 3B-2	PAMP-02- 3A-2	PAMP-02- 3B-1	P002B- 3C-1	D-04- 2A-1	D-04- 2A-3	SCMS-21-1- C2A	SCMS-21- 3C1	SCMS-30- 2-C1
	Pumice	Pumice		Pumice			Pumice	Pumice	Pumice				
Position	Core	Core	Core	Core	Core	Rim	Core	Core	Core		Rim	Core	Core
SiO ₂	0.06	0.06	0.10	0.00	0.09	0.05	0.00	0.00	0.03	0.00	0.11	0.63	0.11
Al_2O_3	1.16	2.48	2.16	1.34	1.26	1.49	1.34	1.23	2.07	1.02	2.14	2.19	1.31
TiO ₂	4.64	5.11	2.91	2.30	2.59	2.65	2.30	0.92	2.57	3.56	7.73	7.28	3.32
FeO	33.31	33.40	31.61	31.44	32.19	31.47	31.44	27.25	31.26	32.23	37.03	36.02	32.00
MnO	1.23	1.14	1.07	0.57	0.54	0.76	0.57	0.84	0.71	0.84	0.40	0.61	0.72
MgO	0.54	0.89	0.69	0.70	0.56	0.73	0.70	2.43	0.98	0.67	0.59	1.07	1.04
CaO	0.01	0.03	0.14	0.02	0.00	0.01	0.02	0.00	0.02	0.14	0.09	0.05	0.13
Cr_2O_3	0.07	0.06	0.05	0.04	0.10	0.06	0.04	0.09	0.07	0.01	0.09	0.03	0.08
Fe_2O_3	57.99	56.02	60.26	62.56	62.05	61.38	62.56	66.12	61.16	60.60	50.43	49.73	61.10
ZnO	0.03	0.32	0.21	0.16	0.12	0.15	0.16	0.10	0.06	0.08	0.01	0.14	0.12
V_2O_3	0.26	0.32	0.32	0.39	0.24	0.26	0.39	0.43	0.22	0.26	0.41	0.40	0.30
NiO	0.00	0.00	0.00	0.00	0.00	0.08	0.00	0.10	0.00	0.00	0.10	0.00	0.04
Total	99.29	99.83	99.52	99.52	99.74	99.08	99.52	99.51	99.15	99.39	99.13	98.15	100.26

(plagioclase), MgO vs CaO (pyroxene), P₂O₅ vs SiO₂ (apatite), SiO₂ vs MgO (pyroxene), SiO₂ vs Sr (plagioclase) and also versus other trace elements (diagrams not shown) as Ba (biotite) and Y (amphibole).

Absolute SiO₂ (wt.%) is between 60.9 and 66.6, the most silicic units being the Alto de Las Lagunas and Luingo I and Luingo II Ignimbrites. Mg# is between 0.32 and 0.5 for the pyroclastic facies,

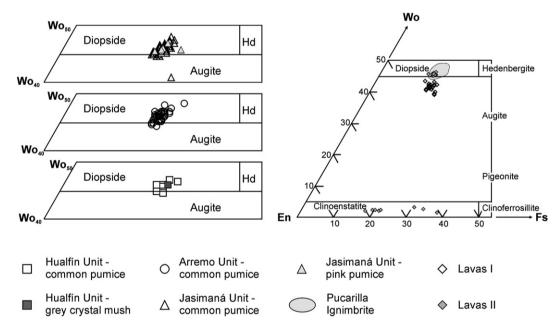


Fig. 4. Pyroxenes from Luingo caldera pumices plotted in the Morimoto (1988) classification diagram.

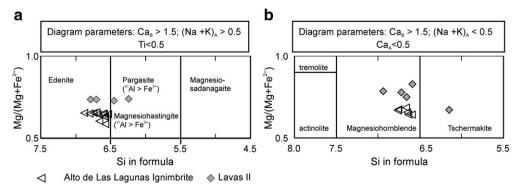


Fig. 5. Calcic amphiboles from the PCTVC classified in the Leake et al. (1997) diagrams.

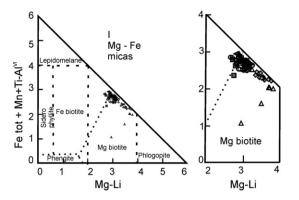


Fig. 6. Micas from PCTVC plotted in the Tischendorf (1997) classification diagram. Symbols as in Figs. 4 and 5.

and is highest in the Hualfin Unit, intermediate in the Arremo Unit and the lowest in the Jasimaná Unit and Alto de Las Lagunas Ignimbrite. In some cases there is a marked variation in the absolute content of major elements even within samples of a given unit. This is particularly evident in the Jasimaná Unit where SiO₂ (wt.%) can vary by ~3–4%; variations are also recorded in minor elements both mobile and immobile (e.g., Ba, Rb, Sr, Zr, and Y) but is sharper in the first ones. In the Pucarilla Ignimbrite – where more samples were collected and analyzed - compositional variations within units are evident from vertical profiles. These are especially obvious for SiO₂, K₂O and Rb, which are higher in the central parts of each unit (Fig. 11). The pattern displayed by K₂O (wt.%) fits well with classical devitrification patterns in ignimbrites (Smith, 1960), where the more devitrified zones (enrichment in K2O: e.g., Lipman, 1965; Kochhar, 1977) are in the middle sectors. We interpret that, at least in the case of SiO2 and of mobile elements, these variations might be related to post-depositional processes like vapour-phase alteration. In the case of immobile elements such as Zr and Y, this variation within each ignimbrite could be better

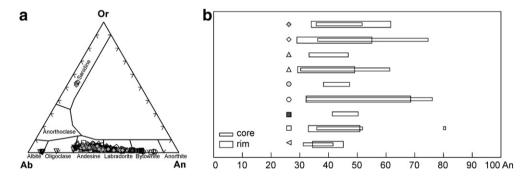


Fig. 7. Feldspars from PCTVC a) An-Or-Ab classification diagram, b) anorthite range content for plagioclases of the different units of PCTVC. Down-pointing triangles belong to Luingo II Ignimbrite, light-grey circles are from Arremo Unit pink pumices and other symbols as in the figures above.

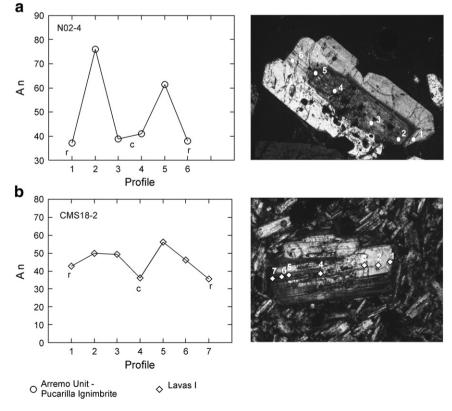


Fig. 8. Selected plagioclase profiles a) Arremo Unit-Pucarilla Ignimbrite; b) Lavas I.

 Table 6

 Geochemical analyses showing absolute values of major, minor and rare earth elements and isotopic ratios.

Unit	Luingo I Ig	nimbrite	Luingo II	Ignimbri	te	Alto de Las	Lagunas	Ignimbrite	Hualfín Ur	nit–Pucari	lla Ignimb	orite			
Sample	CMS-05	CMS- 07	PAMP04	CMS- 13	CMS- 15	CMS-38	CMS- 38PR	CG-04	D-02	H-006	H-008		P-007	PAMPO	06
Location	26 07 39.1 S 66 34 53.6 W	26 07 38 S 66 34 55.1 W	26 05 44.2 S 66 33 0.1 W	26 07 38 S 66 34 55.1 W	26 08 16.4 S 66 36 08.3 W	26 05 13.7 S 66 41 15.7 W	26 05 13.7 S 66 41 15.7 W	25 59 50 \$ 66 45 24.9 W	25 56 20.7 S 66 19 27.1 W	25 48 59.6 S 66 21 50.1 W	25 48 59.6 S 6 21 50.1		25 56 9.7 S 66 17 47.6 W	25 56 09.3 S 17 50.3	
SiO ₂	66.44	65.48	63.87	61.26	62.77	65.03	66.61	65.61	60.91	61.96	62.07		65.14	65.13	
Al_2O_3	15.88	15	15.84	15.67	14.62	15.7	15.68	15.5	16.45	15.78	16.01		14.35	15.05	
Fe_2O_3	2.93	3.44	4.66	4.94	4.07	4.96	3.7	3.65	4.99	4.58	4.65		4.36	4.1	
MnO	0.05	0.09	0.1	0.1	0.1	0.13	0.03	0.05	0.11	0.11	0.1		0.09	0.09	
MgO	1.2	1.66	2.02	2.18	1.61	1.23	1.14	1.16	2.28	2.38	2.4		1.8	1.68	
CaO	2.54	3.28	3.93	4.47	4.49	3.4	3	3.47	4.29	3.37	3.4		3	3.17	
Na ₂ O	2.37	2.7	3.06	2.64	2.14	2.77	2.58	3.13	2.93	2.73	2.61		2.76	2.93	
K ₂ O	5.8 0.23	4.41 0.24	4.18 0.26	4.24 0.32	4.35 0.26	4.19 0.26	4.35 0.22	4.27 0.24	3.65 0.28	3.73 0.27	3.62 0.28		4.36 0.27	4.15 0.26	
P_2O_5 Ti O_2	0.23	0.24	0.20	0.52	0.20	0.62	0.22	0.24	0.28	0.27	0.28		0.63	0.58	
LOI	2	3.02	1.97	3.73	5.54	1.71	2.35	1.29	3.22	3.77	3.94		2.8	2.98	
Total	99.88	99.79	100.52	100.22	100.49	100	100.16	99.05	99.78	99.34	99.71		99.56	100.12	
Cr	7	4	7	12	2	5	4	55.65	5	8	8		7	7	
Ba	588	563	602	593	619	402	397		553	491	491		403	491	
Be	2	3	3	1	1	2	3		3	3	2		3	3	
Cs	6.3	9	5.9	14.2	5.9	7.1	8.3		6.3	8.8	8.4		11.5	11.5	
Ga	17.9	17	18.6	19	17.2	17.2	17.5		19.1	19.6	19.6		16.8	19.5	
Hf	4.7	4.9	5.7	5	4.7	4.7	5.3		6	6.6	6		5.3	5.1	
Nb	23.2	21.4	23.3	25.2	22.4	22.3	21.7		23.2	26.7	26.4		26.8	25	
Rb	237 2	180	169 2	179 2	184 1	177 2	186 2		145 2	168	168 3		200	195 2	
Sn Sr	409	2 469	492	485	410	379	366		508	2 428	433		2 355	409	
Ta	1.6	1.4	1.7	1.7	1.5	1.6	1.6		1.6	2	1.9		2	1.9	
Th	21.6	19.4	17.5	14.6	17.3	16.5	19.5		15.9	24.3	23.2		24.7	22.6	
U	5.3	4.9	4.4	3.9	3.9	3.8	4		4	4.5	4.2		5.3	4.9	
V	63	66	95	99	70	103	88		98	91	91		86	89	
Zr	177	189	190	195	184	173	202		229	230	228		180	180	
Y	21.6	21.2	24.3	25.5	24.5	36.2	27.1		23.1	25.4	23.9		26.1	23.8	
La	41.6	37.4	40.6	39.5	37.8	33.2	35.2		38.5	45.5	44.3		43.5	43.2	
Ce	81.3	72.8	87.5	83.9	78.1	73.3	70.1		82.3	94.4	91.7		96.6	88	
Pr	8.1	7.33	9.01	9.15	8.17	8.53	8.24		9.19	9.5	9.55		9.6	8.76	i
Nd	27.6	25.5	30.7	32	29.4	31.6	29.5		30.9	36.7	31.6		34.9	31.9	
Sm	4.91	4.42	6.4	6.15	5.51	6.46	5.79		6.4	6.1	6		7.5	6	
Eu Gd	1.1 3.63	0.96 3.49	1.04 4.1	1.32 4.77	1.13 4.25	1.24 5.61	1.12 4.72		1.09 4.4	1.18 3.85	1.11 3.99		1.07 5.08	1.07 4.81	
Tb	0.61	0.58	0.65	0.78	0.72	0.98	0.82		0.77	0.9	0.73		0.94	0.67	
Dy	2.88	2.85	3.54	3.63	3.36	4.95	3.88		3.88	4.09	3.34		3.73	3.9	
Ho	0.56	0.54	0.76	0.72	0.68	0.93	0.73		0.72	0.82	0.71		0.86	0.82	
Er	1.81	1.81	2.21	2.18	2.15	3.11	2.32		2.15	2.22	2.27		2.42	2.03	
Tm	0.27	0.25	0.35	0.32	0.32	0.44	0.35		0.35	0.37	0.39		0.39	0.35	
Yb	2	1.88	2.44	2.17	2.19	3.09	2.43		1.78	2.37	2.1		2.44	2.29)
Lu	0.31	0.3	0.29	0.31	0.3	0.46	0.37		0.4	0.42	0.43		0.36	0.36	i
143Nd/144	\Ndm					0.51232		$0.512308\pm$				375 ± 7	0.512411		2384 ± 16
143Nd/144 147Sm/144	Ndi					0.51231	1	0.512298	0.51243	2	0.512		0.512402		2375
,	'Nd					0.1156		0.1131	0.11		0.106	15	0.1091	0.11	
εNd (t) TDM						-6.05 1.12		-6.30 1.11	-3.71 0.90		-4.99 0.95		-4.29 0.10	-4.82 0.97	
⁸⁷ Sr/ ⁸⁶ Srn	n					0.70877	⊥ 2	0.70871 ± 2	0.90	⊥ 3		37±33	0.10 0.70778±		928±3
87Sr/86Sri						0.70877	± 2	0.70871±2	0.70795		0.703		0.70777	0.70	
Unit		mo Unit-	Pucarilla l	gnimbrite	:	0170002		Jasimaná ¹	Unit-Pucarilla				0.70777	0170	
Sample	D-03	B PA	MP02	D-07	N-02	P-008	PAMP0	9 D-04	D-08 CN	/IS-31	H-009	P-002	P-006	PAMP01	CG-01
Location	25 5	6 26	09	25 58	25 56	25 56	25 56	25 56	25 58 26	09	25 48	25 54	25 55	26 09	25 56
Location	20.7			11.9 S	9.4 S	9.7 S	09.3 S	20.7 S			59.6 S	31.1 S	18.7 S	30.6 S	27 S
	66 1			66 18	66 17	66 17	66 17	66 19			66 21	66 19	66 19	66 27	66 38
	27.1			41.8 W	50.9 W	47.6 W	50.7 W				50.1 W	29.4 W		47.7 W	10.9 W
SiO ₂	61.	73 6	2.51	63.21	65.24	64.86	64.64	62.81	62.39 6	3.63	65.54	64	63.97	63.65	64.87
Al_2O_3	16.		6.52	15.1	14.72	14.87	14.56	16.67		6.4	16.13	15.93		16.45	15.85
Fe ₂ O ₃	4.		4.95	4.59	4.25	4.41	4.08	5.18		4.83	4.28	4.66		4.97	4.25
MnO	0.		0.11	0.1	0.1	0.1	0.09	0.1		0.1	0.1	0.1	0.11	0.11	0.1
MgO	2.		2.03	2.01	1.8	1.91	1.76	1.98		1.47	1.1	1.7	1.57	1.84	1.6
	4.		4.64	3.71	3.28	3.56	3.48	4.52		4.01	3.69	4.04		4.52	3.82
			224	2.93	2.87	2.96	2.96	3.32	3.41	3.3	3.19	3.3	3.09	3.26	3.34
Na ₂ O	3.		3.24										4 0 0		
Na ₂ O K ₂ O	3.	3	3.9	4.08	4.43	4.38	4.44	3.92	3.91	4.31	4.46	4.19		4.06	4.3
CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂	3. 0.	3 28							3.91 0.28			4.19 0.29 0.64	0.27		

(continued on next page)

Table 6 (continued)

Unit	Arremo l	Jnit–Pucarill	a Ignimbrit	e			Jasimaná	Unit-Puca	rilla Ignim	brite				
Sample	D-03	PAMP02	D-07	N-02	P-008	PAMP09	D-04	D-08	CMS-31	H-009	P-002	P-006	PAMP01	CG-01
Location	25 56 20.7 S 66 19 27.1 W	26 09 43.7 S 66 27 50.9 W	25 58 11.9 S 66 18 41.8 W	25 56 9.4 S 66 17 50.9 W	25 56 9.7 S 66 17 47.6 W	25 56 09.3 S 66 17 50.7 W	25 56 20.7 S 66 19 27.1 W	25 58 11.9 S 66 18 41.8 W	26 09 22.6 S 66 44 51.8 W	25 48 59.6 S 66 21 50.1 W	25 54 31.1 S 66 19 29.4 W	25 55 18.7 S 66 19 15 W	26 09 30.6 S 66 27 47.7 W	25 56 27 S 66 38 10.9 W
101														
LOI Total	2.33 99.36	1.38 100.21	3.22 99.85	2.52 100.06	2.33 100.26	2.48 99.18	0.63 100.13	0.77 99.19	0.82 99.82	1.18 100.52	0.55 99.4	1.25 99.74	0.7 100.5	0.38 99.36
Cr	19	8	8	5	6	11	8	10	7	6	11	5	7	
Ba	540	549	476	470	504	496	560	569	654	627	569	556	555	
Be Cs	3 7.9	3 9.3	3 13.6	3 13	3 11.1	3 11.6	3 6.5	3 5.6	2 6.6	2 6.2	3 8.3	3 7.4	3 6.1	
Ga	20.2	20.8	20.2	17.9	16.1	17.25	22.2	20.7	19.4	19.3	20.5	18.2	20.6	
Hf	5.2	5.9	5.7	5.9	5.3	5.7	4.7	5.3	5.4	5.5	6.2	5.5	6.1	
Nb Rb	24.9 161	25.9	25.7 180	24.5 191	24.9 194	24.7 193	25.7	25.4 145	25.3 175	23 182	24.8 178	24.6 179	23.7 165	
Sn	2	167 2	2	2	2	2	157 2	2	175	2	2	2	2	
Sr	527	562	407	387	418	402	540	525	550	465	496	500	527	
Ta	1.7	1.8	1.9	1.7	1.9	1.95	1.7	1.7	1.7	1.8	1.9	1.9	1.9	
Th U	17.5 4.1	20.6 4.2	19.1 4.5	18.7 5.2	21.7 5.3	21.3 5.35	16.2 4	15 3.8	19 3.6	19.6 4.3	19 4.1	18.7 4	16.8 3.8	
V	99	112	88	90	97	88.5	108	102	85	84	84	104	99	
Zr	196	198	194	197	182	173	196	196	205	204	192	184	203	
Y	25.8	27.8	25.5	22.5	25.4	24	27.1	24.7	26.3	22.2	25.3	24.2	26.4	
La Ce	43.2 91.3	43.4 94.8	43.5 89.5	41.3 86.6	42.9 93.6	42.75 90.1	43.3 92.3	42.2 89.1	41.5 89.2	42.3 82.5	42.1 90.2	42 86.8	41.5 88.1	
Pr	9.47	9.64	9.27	8.66	9.15	9.17	9.55	9.5	9.7	8.68	9.01	9.13	9.66	
Nd	38	36.1	34.3	28.7	32.3	32.25	34.8	37.5	35.5	30.2	32.1	33.4	37.7	
Sm	6.3	6.9	6.3	5.4	6.3	6.55	6.6	6.8	6.38	6	6.4	6.1	7.2	
Eu Gd	1.29 4.88	1.36 4.94	1.1 4.36	1.04 4.23	1.06 3.91	1.03 4.31	1.48 4.86	1.3 3.88	1.43 5.06	1.08 3.17	1.25 4.53	1.26 4.4	1.22 4.89	
Tb	0.8	0.8	0.81	0.78	0.89	0.75	0.88	0.84	0.86	0.74	0.82	0.73	0.87	
Dy	4.14	4.15	4.11	3.87	3.92	3.65	4.57	4.23	4	3.45	4.05	3.43	3.95	
Ho	0.81	0.99	0.76	0.75	0.87	0.76	0.82	0.72	0.74	0.68	0.83	0.77	0.82	
Er Tm	2.15 0.35	2.56 0.38	2.24 0.33	2.21 0.37	2.2 0.4	2.26 0.34	2.44 0.35	2.05 0.35	2.33 0.33	1.96 0.32	1.98 0.36	2.14 0.3	2.26 0.32	
Yb	2.01	2.51	2.15	2.45	2.35	2.28	2.01	1.77	2.28	1.92	2.41	2.6	2.4	
Lu 143 Nd/144 Ndm 143 Nd/144 Ndi 147 Sm/144 Nd ENd (t) TDM 87 Sr/86 Srm	0.41 0.5123 0.5123 0.11 -5.39 1.02 0.7086		0.4	0.32 0.5124 0.5124 0.09 -3.28 0.75 0.7089		0.36	0.44	0.39 0.5123 0.5123 0.1115 -4.96 0.29 0.7076	į	0.34	0.5124 0.19 -3.27	76±2	0.4 0.51239 0.51239 0.11 -4.54 0.91 0.70793	90
⁸⁷ Sr/ ⁸⁶ Sri	0.7085	4		0.7082	26			0.7075	0		0.7075	58	0.70778	3
Unit	Lavas		CN CC D4		7.40.044	CNIC	40	CN 1C OO		Lavas II		CN 46 00	CD #	2.20
Sample	CMS-	18 21.3 S	CMS-21 26 08 35.0		MS-24A	CMS-		CMS-03	. 1 C	CMS-29A		CMS-28		S-30 09 19.1 S
Location		13.8 W	66 36 24.		6 07 31.7 S 6 37 23.6 W		44.5 S 40.6 W	26 07 37 66 34 38		26 06 47 S 66 40 25.7 V		27 06 47 S 66 40 25.7		42 51.5 W
SiO ₂	59.6		60.87		15.01	58.		65.81		65.43		66.39		.16
Al_2O_3	16.89 5.33		16.5 5.77		15.61 6.89	15. 6	66 25	15.68 3.27		15.75 3.85		15.44 3.6		.52 .59
Fe ₂ O ₂	5.5.		0.07		0.09		07	0.08		0.06		0.06		.07
Fe ₂ O ₃ MnO	0.1				3.66	າ	35	1.31		1.86		1.68		.65
MnO MgO	2.13	8	2.36											
MnO MgO CaO	2.13 4.9	8 5	4.48		6.26	5.	43	2.03		3.49		3.34		.11
MnO MgO CaO Na ₂ O	2.13 4.93 3.19	8 5 9	4.48 2.66		6.26 2.82	5. 2.	43 92	2.03 2.83		3.49 3.25		3.38	3	.03
MnO MgO CaO	2.13 4.9	8 5 9 2	4.48		6.26	5. 2. 4.	43	2.03		3.49			3	
$\begin{array}{l} MnO \\ MgO \\ CaO \\ Na_2O \\ K_2O \\ P_2O_5 \\ TiO_2 \end{array}$	2.13 4.93 3.13 4.44 0.42 0.7	8 5 9 2 2	4.48 2.66 4.56 0.42 0.69		6.26 2.82 4.18 0.53 0.87	5. 2. 4. 0. 0.	43 92 85 53 81	2.03 2.83 5.13 0.28 0.57		3.49 3.25 3.91 0.22 0.54		3.38 4.13 0.22 0.52	3 3 0 0	.03 .69 .27
$\begin{array}{l} \text{MnO} \\ \text{MgO} \\ \text{CaO} \\ \text{Na}_2\text{O} \\ \text{K}_2\text{O} \\ \text{P}_2\text{O}_5 \\ \text{TiO}_2 \\ \text{LOI} \end{array}$	2.13 4.93 3.19 4.44 0.42 0.7 1.33	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08		6.26 2.82 4.18 0.53 0.87 1.79	5. 2. 4. 0. 0. 2.	43 92 85 53 81 04	2.03 2.83 5.13 0.28 0.57 2.09		3.49 3.25 3.91 0.22 0.54 1.89		3.38 4.13 0.22 0.52 1.57	3 3 0 0 1	.03 .69 .27 .62
$\begin{array}{c} MnO \\ MgO \\ CaO \\ Na_2O \\ K_2O \\ P_2O_5 \\ TiO_2 \end{array}$	2.13 4.93 3.13 4.44 0.42 0.7	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69		6.26 2.82 4.18 0.53 0.87	5. 2. 4. 0. 0.	43 92 85 53 81 04	2.03 2.83 5.13 0.28 0.57		3.49 3.25 3.91 0.22 0.54		3.38 4.13 0.22 0.52	3 3 0 0 1	.03 .69 .27 .62 .67
$\begin{array}{l} \text{MnO} \\ \text{MgO} \\ \text{CaO} \\ \text{Na}_2\text{O} \\ \text{K}_2\text{O} \\ \text{P}_2\text{O}_5 \\ \text{TiO}_2 \\ \text{LOI} \\ \text{Total} \\ \text{Cr} \\ \text{Ba} \end{array}$	2.1: 4.9: 3.1: 4.4: 0.4: 0.7 1.3: 99.2	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46		6.26 2.82 4.18 0.53 0.87 1.79 42.7 17	5. 2. 4. 0. 2. 100. 35	43 92 85 53 81 04	2.03 2.83 5.13 0.28 0.57 2.09 99.06		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10		3.38 4.13 0.22 0.52 1.57 100.33 12 555	3 3 0 0 1 99 56 644	.03 .69 .27 .62 .67
$\begin{array}{l} \text{MnO} \\ \text{MgO} \\ \text{CaO} \\ \text{Oa}_{\text{O}} \\ \text{Na}_{\text{2}}\text{O} \\ \text{K}_{\text{2}}\text{O} \\ \text{P}_{\text{2}}\text{O}_{\text{5}} \\ \text{TiO}_{\text{2}} \\ \text{LOI} \\ \text{Total} \\ \text{Cr} \\ \text{Ba} \\ \text{Be} \end{array}$	2.1: 4.9: 3.1: 4.4: 0.4: 0.7 1.3: 99.2: 886 3	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46		6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70	5. 2. 4. 0. 2. 100. 35 1127	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2	3 3 0 0 1 99 56 644 2	.03 .69 .27 .62 .67
MnO MgO CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂ LOI Total Cr Ba Be Cs	2.13 4.99 3.11 4.44 0.4 0.7 1.33 99.2 886 3 5.1	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6	5. 2. 4. 0. 2. 100. 35 1127 3	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4	3 3 0 0 1 99 56 644 2 39	.03 .69 .27 .62 .67
$\begin{array}{l} \text{MnO} \\ \text{MgO} \\ \text{CaO} \\ \text{Oa}_{\text{O}} \\ \text{Na}_{\text{2}}\text{O} \\ \text{K}_{\text{2}}\text{O} \\ \text{P}_{\text{2}}\text{O}_{\text{5}} \\ \text{TiO}_{\text{2}} \\ \text{LOI} \\ \text{Total} \\ \text{Cr} \\ \text{Ba} \\ \text{Be} \end{array}$	2.13 4.99 3.11 4.44 0.4 0.7 1.33 99.2 886 3 5.1 19.7	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7 19.2	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9	5. 2. 4. 0. 0. 2. 100. 35 1127 3 3.	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1 22.4		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4 21.5	3 3 0 0 1 99 56 644 2 39	.03 .69 .27 .62 .67 .38
MnO MgO CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂ LOI Total Cr Ba Be Cs Ga	2.13 4.99 3.11 4.44 0.4 0.7 1.33 99.2 886 3 5.1	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9 6.3 25.7	5. 2. 4. 0. 2. 100. 35 1127 3	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1 5.1 24.2		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4	3 3 0 0 1 99 56 644 2 39	.03 .69 .27 .62 .67 .38
$\begin{array}{l} \text{MnO} \\ \text{MgO} \\ \text{CaO} \\ \text{Na}_2\text{O} \\ \text{K}_2\text{O} \\ \text{P}_2\text{O}_5 \\ \text{TiO}_2 \\ \text{LOI} \\ \text{Total} \\ \text{Cr} \\ \text{Ba} \\ \text{Be} \\ \text{Cs} \\ \text{Ga} \\ \text{Hf} \\ \text{Nb} \\ \text{Rb} \\ \end{array}$	2.14 4.9 3.14 4.4 0.7 1.3 99.2 886 3 5.1 19.7 6.1 25.7	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7 19.2 7.1 25.6 204	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9 6.3 25.7	5. 2. 4. 0. 2. 100. 35 1127 3 3. 19 6. 26.	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1 5.1 24.2 223.7		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1 22.4 5.2 15.1		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4 21.5 5 14.6 208	3 3 0 0 1 99 56 644 2 39 20 5 14	.03 .69 .27 .62 .67 .38
MnO MgO CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂ LOI Total Cr Ba Be Cs Ga Hf Nb Rb Sn	2.14 4.9 3.11 4.4.4 0.7 1.33 99.2 886 3 5.1 19.7 6.1 25.7 162 2	8 5 9 2 2 2	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7 19.2 7.1 25.6 204 2	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9 6.3 25.7	5. 2. 4. 0. 2. 100. 35 1127 3 3. 19 6. 26, 203	43 92 85 53 81 04 25	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1 5.1 24.2 223.7 2		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1 22.4 5.2 15.1		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4 21.5 5 14.6 208 2	3 3 0 0 1 99 56 644 2 39 20 5 14	.03 .69 .27 .62 .67 .38
MnO MgO CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂ LOI Total Cr Ba Be Cs Ga Hf Nb	2.14 4.9 3.14 4.4 0.7 1.33 99.2 886 3 5.1 19.7 6.1 25.7 162 2	8 5 9 9 2 2 2 8 7	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7 19.2 7.1 25.6 204 2 554	8	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9 6.3 25.7 69 2	5. 2. 4. 0. 2. 100. 35 1127 3 3. 19 6. 26.	43 92 85 53 81 04 225	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1 5.1 24.2 223.7 2		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1 22.4 5.2 15.1		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4 21.5 5 14.6 208 2 361	3 3 0 0 1 99 56 644 2 39 20 5 14 171 2 438	.03 .69 .27 .62 .67 .38
MnO MgO CaO Na ₂ O K ₂ O P ₂ O ₅ TiO ₂ LOI Total Cr Ba Be Cs Ga Hf Nb Rb Sn Sr	2.14 4.9 3.11 4.4.4 0.7 1.33 99.2 886 3 5.1 19.7 6.1 25.7 162 2	8 5 9 2 2 2 2 8 7	4.48 2.66 4.56 0.42 0.69 2.08 100.46 919 3 2.7 19.2 7.1 25.6 204 2	1 7	6.26 2.82 4.18 0.53 0.87 1.79 42.7 17 70 2 3.6 19.9 6.3 25.7	5. 2. 4. 0. 2. 100. 35 1127 3 3. 19 6. 26. 203 2 734	43 92 85 53 81 04 225	2.03 2.83 5.13 0.28 0.57 2.09 99.06 586 2 7.8 18.1 5.1 24.2 223.7 2		3.49 3.25 3.91 0.22 0.54 1.89 100.25 10 580 2 14.1 22.4 5.2 15.1 191 2		3.38 4.13 0.22 0.52 1.57 100.33 12 555 2 11.4 21.5 5 14.6 208 2	3 3 0 0 1 99 56 644 2 39 20 5 14 171 2 438 1	.03 .69 .27 .62 .67 .38 .8 .8 .8

Table 6 (continued)

Unit	Lavas I					Lavas II		
Sample	CMS-18	CMS-21	CMS-24A	CMS-42	CMS-03	CMS-29A	CMS-28	CMS-30
Location	26 08 21.3 S 66 36 13.8 W	26 08 35.6 S 66 36 24.1 W	26 07 31.7 S 66 37 23.6 W	26 06 44.5 S 66 37 40.6 W	26 07 37.1 S 66 34 38.3 W	26 06 47 S 66 40 25.7 W	27 06 47 S 66 40 25.7 W	26 09 19.1 S 66 42 51.5 W
Zr	263	262	241	246	194.5	195	195	219
Y	28.5	32.4	30.2	27.8	22.8	18.6	17	24
La	44.7	46.9	45.1	47	42.5	45.4	43.8	45.2
Ce	98.7	101	100.2	104.8	80.6	98.3	92.9	91
Pr	11.03	11.45	11.44	12.01	8.6	10.41	9.97	10.41
Nd	39.7	41.7	41.9	43.5	31.8	36.6	34.8	38.4
Sm	7.61	8.06	8.15	8.22	5.36	6.45	6.21	6.8
Eu	1.63	1.67	1.76	1.71	1.17	1.17	1.07	1.28
Gd	5.65	6.04	6.36	6.04	4.29	4.39	3.98	4.99
Tb	0.92	1.02	0.98	0.94	0.7	0.67	0.61	0.77
Dy	4.5	4.68	4.66	4.4	3.36	2.92	2.71	3.33
Но	0.78	0.86	0.86	0.81	0.68	0.49	0.47	0.62
Er	2.41	2.78	2.54	2.4	2.15	1.47	1.31	1.78
Tm	0.33	0.39	0.36	0.34	0.34	0.21	0.17	0.25
Yb	2.34	2.76	2.39	2.33	2.23	1.43	1.31	1.71
Lu	0.34	0.4	0.34	0.33	0.33	0.2	0.18	0.25
¹⁴³ Nd/ ¹⁴⁴ Ndm	0.512333 ± 15	5				0.512305 ± 12		
¹⁴³ Nd/ ¹⁴⁴ Ndi	0.512325					0.512300		
¹⁴⁷ Sm/ ¹⁴⁴ Nd	0.104711					0.110685		
εNd (t)	-5.81					-6.41		
TDM	1.00					1.02		
⁸⁷ Sr/ ⁸⁶ Srm	0.70712 ± 3					0.70994 ± 3		
⁸⁷ Sr/ ⁸⁶ Sri	0.70701					0.70979		

explained by mechanical fractionation of accessory minerals produced during transport/depositional processes.

Chondrite (Mc Donough and Sun, 1995)-normalized rare earth elements (Fig. 12a, b) yield patterns with a moderate slope, corresponding to (La/Yb)_N between 17 and 10, exceptionally 8. (La/Sm)_N ranges from 5.5 to 3.5 and the HREE define a slope of 2 to 1.3 $(Gd/Yb)_{N}$; the patterns show no garnet as a residual phase. Eu/Eu* average is 0.69 for the ignimbrites indicating plagioclase fractionation, probably at shallow levels. Some slight differences in the MREE and HREE enrichment found within the Luingo ignimbrites could be related to variations in the amounts of mineral phases – such as hornblende and/or zircon – involved in the fractionation processes. The Lavas I show REE patterns very similar to those of the ignimbrites and the subtle differences could be related to different proportions of fractionation of mineral phases. For instance, a small variation in the amount of titanite can greatly modify these patterns. On the other hand, much more marked differences are evident when comparing with the samples from Lavas II, where the presence of garnet in the source is evident.

Chondrite (Mc Donough and Sun, 1995)-normalized multielement diagrams (Fig. 12c, d) show a common behaviour for the ignimbrites of the Luingo caldera; a slight difference is marked by the higher concentration of Y and Yb in the Alto de Las Lagunas Ignimbrite. The lava flows of different age (Lavas I and Lavas II) are clearly separated also by their geochemical features; the middle Miocene Lavas I have patterns that resemble those of the ignimbrites, whereas the upper Miocene Lavas II show contrasting patterns. Lavas I have greater contents in Ba, Sr, P and Zr than the ignimbrites. These differences can be ascribed to different proportions of a subcrustal mafic end-member in the hybrid magma (Ba, Sr) and/or to different magma temperatures (higher P and Zr in Lavas I is consistent with higher T in these); or may also be related to different fractionation degrees of phases such as plagioclase and biotite. The Lavas II group has more prominent Nb-Ta anomalies (i.e., a more arc-like signature or more crustal component) and less content in Yb and Y (probably related to residual phases such as garnet and/or amphibole).

The PCTVC isotopic $^{87}\text{Sr}/^{86}\text{Sr}_i$ values are in the range of 0.70750 to 0.70918 and $^{143}\text{Nd}/^{144}\text{Nd}_i$ varies between 0.511230 and 0.512455. $^{87}\text{Sr}/^{86}\text{Sr}$ has positive correlation with SiO₂ and Rb/Sr thus indicating that not only fractional crystallization but also assimilation processes occurred. Nevertheless, the wide range of isotopic ratios and T_{DM} values points to complex magmatic processes and is sufficient to discard closed-system magmatic evolution of the PCTVC.

From the geochemical analysis as a whole, some processes are evident: a) fractional crystallization, b) assimilation of crustal material (e.g., on the base of isotopic ratios), c) the presence of a mafic source during recharge that led to the high An % contents (more evident in the Arremo Unit). The latter is also consistent with other evidence, such as different compositional groups of the same mineral phases (e.g., biotite, spinels) within a given unit, and petrographic disequilibrium features.

Although the magmatic evolution of PCTVC must have included complex magmatic processes, probably with deep or intermediate magma chambers feeding the shallowest and ultimate magma chamber within the crust, as suggested for many other calderas in the Puna (e.g., de Silva et al., 2006), we performed a simplified model of the evolutive process in the last magma chamber only.

Together, the geochemical and petrographic features point to assimilation and fractional crystallization with recharge (AFC-r) as the most feasible process.

Before moving to AFC-r, we also made attempts (not shown) to model the Luingo samples by a mixing-only model, but, as expected, no successful results were found. For the AFC-r model we followed the formulation proposed by DePaolo (1981) and extended by Aitcheson and Forrest (1994). We first needed to decide which end members (i.e. initial magma composition and assimilant) and recharge magma to use. Because of the limited composition range of the PCTVC exposed rocks, adequate extreme compositions do not occur within the set of volcanic samples analyzed in this work. We therefore selected well known samples from both the local basement and basic rocks within the SCVZ.

For the original magma composition we chose to use the Las Maquinas Basalt (Kay et al., 1999), as it is a ca. 23 Ma back arc basalt

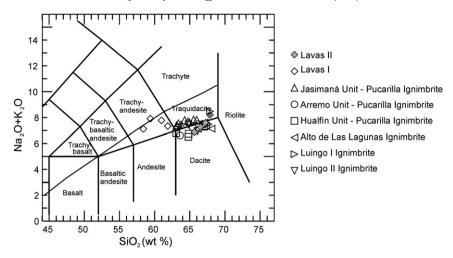


Fig. 9. Total Alcalis Silica (TAS) diagram (Le Maitre et al., 1989) showing the classification of rocks from the PCTVC.

that could represent the basic compositions previous to the major thickening event in this region. Among the results obtained by Caffe et al. (2002) using the Melts software (Ghiorso and Sack, 1995) we selected the theoretical paragenesis of this basalt at 5 kbar and

 $F\!=\!0.5$ (fraction of magma remaining); i.e., CPx: 67%, Pl: 33% thus giving a D_{Sr} of 1.46 and $D_{Nd}\!=\!0.61$ (Table B1, Appendix B in Caffe et al. 2002). As the contaminant we use the average of selected Ordovician granitoids from Antofalla Salar (samples M18/1ª and

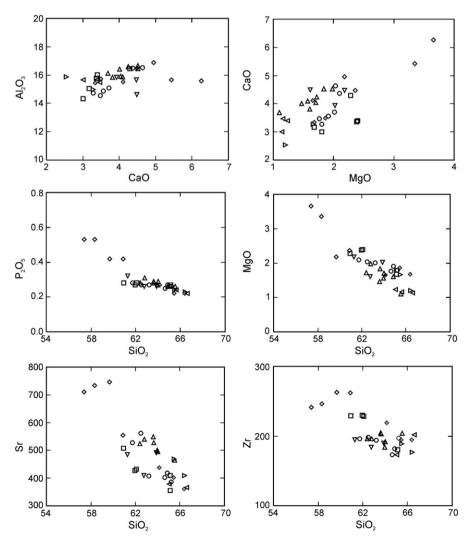


Fig. 10. Selected correlation diagrams of major and minor elements of rocks from the PCTVC. Symbols as in Fig. 10.

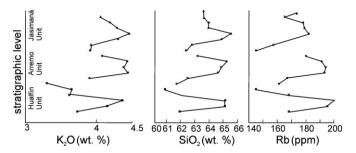


Fig. 11. Stratigraphic level variations of major and minor elements $(K_2O, SiO_2, and\ Rb)$ within Pucarilla Ignimbrite.

M18/5ª from Lucassen et al., 2001). Using the same simplification proposed by Aitcheson and Forrest (1994), we consider for the recharge magma the same composition as the original magma. The graphical method based on Equations (9) and (10) proposed by Aitcheson and Forrest (1994) leads to a family of triple solutions (β , r, ρ) where β = (rate of recharge)/(rate of assimilation), r = (rate of assimilation of crust)/(rate of fractional crystallization) and ρ = (crust)/(magma). However, as suggested by Aitcheson and Forrest (1994), using zero (i.e. no recharge, see Model 1 in Table 7) or low values for β in the presence of high values of bulk D (i.e. highly compatible elements), besides being in contrast with the geochemical evidence discussed above, can lead to an overestimation of ρ . Therefore, we consider that models with higher values of β are more plausible (see Models 2 and 3 in Table 7). Considering all the range

of possible values of β , we always obtain percentages of crustal melts between 17 and 23%.

6. The Luingo caldera and its significance in the southern CVZ

The Luingo caldera is the easternmost large-volume magmatic complex of the CVZ. Its position and age allows constraining possible geochemical changes in the eastern Puna border with time and also evaluating the existence of geochemical variations from W to E during the middle Miocene.

6.1. Crustal thickening in time

During the last decades it has been proposed that the major thickening event within the Puna was about 10 m.y. ago (e.g., James, 1971; Isacks 1988). As the Luingo caldera (middle Miocene) is located close to the Cerro Galán Complex (ca. 6.5 to 2 Ma: e.g., Sparks et al., 1985; Folkes et al., submitted for publication) and is older than the generally accepted time of main thickening event in this area, we first compared our data with those of the younger Cerro Galán Complex (CGC). Some features that come to light from this comparison are a) variation diagrams show correlatable trends, CGC rocks being more evolved, b) REE diagrams show higher LREE and lower HREE for the CGC, i.e., the Luingo caldera magmas show no evidence for garnet as a residual phase, as opposed to what occurs in the Cerro Galán caldera (e.g., Kay et al., 2008), c) N-MORB-normalized multielement patterns of the Luingo caldera are similar to those of the CGC rocks, although the latter evidently contain lesser amounts of P, Ti and Sr, suggesting a

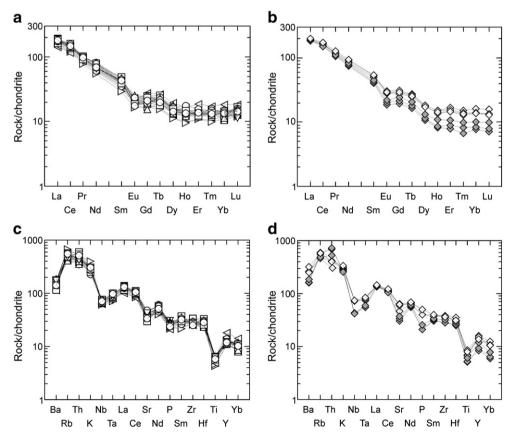


Fig. 12. Chondrite normalized REE (a, b) and multielement diagrams (c, d) diagrams showing (a, c) pyroclastic rocks from the Luingo caldera, (b, d) effusive units from the PCTVC. Normalization chondrite values for multielement diagrams from Thompson (1982), with alternative value for Ba (Hawkesworth et al., 1984); chondrite values for rare earth elements diagrams from (McDonough and Sun, 1995).

Table 7

AFC with recharge model. Original and recharge magma: Las Máquinas basalt. Contaminant: Average of selected Ordovician granitoids (Lucassen et al., 2001); see text. Bulk partition coefficient values as in Caffe et al. (2002) from clinopyroxene and plagioclase partition coefficients given in Bacon and Druitt (1988). The AFC-r models were calculated using DePaolo (1981) equations modified by Aitcheson and Forrest (1994). D (bulk partition coefficient); Ca (element concentration in wall-rock melt); $\rm C^{\infty}m$ (element concentration in the original magma); $\rm E^{\infty}m$ (isotopic ratio in the original magma); $\rm E^{\infty}m$ (isotopic ratio in the contaminated magma), $\rm B^{\infty}n$ (isotopic ratio in the contaminated magma), $\rm B^{\infty}n$ (isotopic ratio in the contaminated magma).

Parameter	Sr	Nd	
D	1.46	0.61	
Ca	71 (67–75)	44 (41-47)	
C°m	520	16	
εа	0.7561705	-13.295	
	(0.751292-0.761049)	(-13.38-13.21)	
ε°m	0.70400	4	
εm	0.70834	-5.0985	
	(0.70750-0.70918)	(-6.63 a-3.567)	
Results	Model 1	Model 2	Model 3
β	0.00	0.75	1.45
r	0.37	0.30	0.17
ρ	0.30	0.28	0.20
% assimilated crust	23%	22%	17%

greater degree of fractionation of certain mineral phases, d) Mg# are similar, indicating similar bulk crystal fractionation in both complexes.

On the other hand, Sr and Nd isotopic ratios (Fig. 13) show a progressive increase in ⁸⁷Sr/⁸⁶Sr and decrease in ¹⁴³Nd/¹⁴⁴Nd from the Luingo caldera rocks (middle Miocene), through Lavas II (upper Miocene), to CGC (upper Miocene to Pliocene), thus indicating an increase in the amount of assimilated component; different proportions and/or compositions of mantle — crustal melts with time (e.g. Kay et al. 2010). This is also confirmed by the AFC and AFC-r models, where the CGC is found to represent a mixture of 50:50 crustal/mantle melts (Kay et al., 2008) whereas Luingo caldera represents ca. 20:80 (this work).

We also selected representative samples from the Cerro Galán Ignimbrite (Pliocene: Folkes et al., submitted for publication), Blanco Ignimbrite (upper Miocene: Folkes et al., submitted for publication) and Luingo caldera ignimbrites (middle Miocene: this work) and plotted their chondrite-normalized REE (see Fig. 14a). Considering the presence of garnet in the source as indicative of crustal thickening (e.g., Coira et al., 1993), different REE trends and slopes are obvious from middle to upper Miocene, but less evident from upper Miocene to Pliocene. So, if we compare Blanco Ignimbrite (upper Miocene) with Galán Ignimbrite (Pliocene), their REE slopes are equal and the slight differences in their absolute content can be easily addressed by FC (e.g., Folkes et al., submitted for publication). This is reinforced if we compare Lavas II (upper Miocene) data (see Table 6) with the Cerro Galán Ignimbrite.

Overall, a higher proportion of crustal material is evident in the CGC (e.g., Kay et al., 2008), in comparison with the middle Miocene Luingo volcanic rocks. Furthermore, CGC magmas exhibit the presence of garnet as a residual phase in the source, which could be interpreted as the result of a thickened crust at this time (e.g., Coira et al., 2003; Kay et al., 2008). The same is true for the upper Miocene Lavas II from the PCTVC.

6.2. Geochemical assessment of arc configuration changes

If we consider a Miocene arc developed close to the present one, 230 km westward from both the Luingo and the Cerro Galán calderas, they are both set in a "back arc" position. This raises the opportunity to try and use geochemical tools to study changes in arc–back arc configuration, as started by Coira et al. (1993) and Kay et al. (1999). To highlight this, we compared the geochemical data from the Luingo

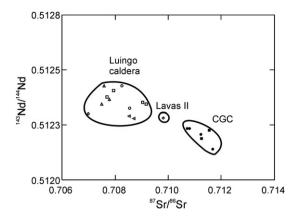


Fig. 13. ⁸⁷Sr/⁸⁶Sr vs ¹⁴³Nd/¹⁴⁴Nd plot of PCTVC and Cerro Galán Complex. PCTVC symbols as in precedent figures; CGC isotopic data (Francis et al., 1989): black squares represent Cerro Galán Ignimbrite and black circles represent upper Miocene ignimbrites of the Cerro Galán Complex.

caldera with those from other middle Miocene volcanic rocks of the Southern CVZ (25–27 S), set between 67 and 69°W. The aim of this comparison is to assess whether geochemical evidence is enough to accurately constrain the arc-back arc position and/or arc width/configuration for middle Miocene times.

It is known that continental arc magmas generally show an enrichment in LIL elements in relation with the concentrations of HFSE (e.g., McDowell and Clabaugh, 1979; Pearce, 1983); this behaviour is basically related to the affinities of these elements with the fluid phases. Through a comparison with the available (Baker et al., 1987; Mpodozis et al., 1995; Kraemer et al., 1999; Trumbull et al., 1999; Siebel et al., 2001; Schnurr et al., 2007) geochemical data over the 25-27° segment of the CVZ it comes out that N-MORB-normalized patterns vary from west to east (this is just a tendency and we need to emphasize that all these element concentrations overlap in the samples from the west and east), showing differences in Th, Ba, Nb and Ta that may mostly reflect the participation of fluid phases in the astenospheric wedge. However, mineral phases as rutile, titanite, feldspars and ilmenite can also affect the concentrations of these elements. Differences in Zr, P, Sr, Ti and Y are also found and can reflect variable fractionation degrees of mineral phases (e.g. amphibole, ilmenite) and/or different proportions of mafic end-members.

Our results show that as was previously pointed by other authors (e.g., Saunders and Tarney, 1984; Davidson and de Silva, 1995; Siebel et al., 2001; Caffe et al., 2002; Schnurr et al., 2007) there is a tendency in the CVZ for Nb concentration to increase, and for Zr/Nb and Ba/Nb to decrease eastward. These variations have been interpreted as a result of variable basement composition from west to east (Schnurr et al., 2007) and/or a decrease in subduction-related signature (Siebel et al., 2001; Caffe et al., 2002); and as variable sources and depth in the Benioff zone (Davidson and de Silva, 1995). It is reasonable to argue that these differences can only reflect different settings if rocks of similar ages are considered (see. e.g., Fig. 13). For instance, it is true that Nb (and Ta) increase eastwards if we compare rocks of the 67-69°W SCVZ with the ones of the Luingo caldera but this is not the case for the Cerro Galán Complex (where they overlap). If this eastward Nb increase and Ba/Nb decrease reflect a progressively increasing distance of the magmas from the arc, it is not possible to explain the differences between Cerro Galán caldera and Luingo caldera which are spatially close, unless a significative time span is considered. This same time-dependence is found if strontium isotopic ratios (e.g., Schnurr et al., 2007) are used as a tool for discrimination between arc and back arc settings. It can also be argued that these ratios and concentrations depend greatly on the petrologic evolution of each volcanic centre, the source of their parental magmas (e.g., Kay et al. 2010, invoke the presence of mineral phases that retain Nb in the source of Galan's ignimbrites) and/or the degree in which crustal

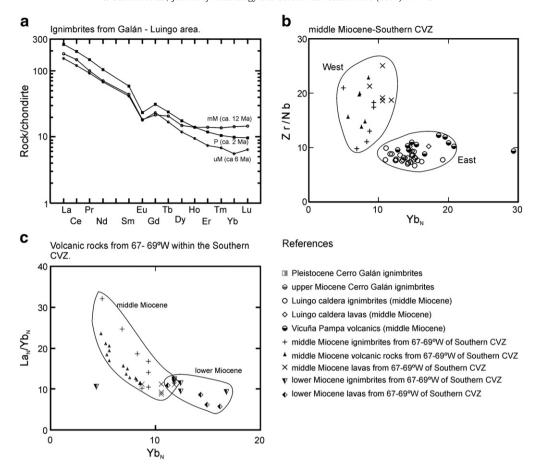


Fig. 14. (a) REE diagram showing volcanic rocks from CGC and PCTVC; (b) Zr/Nb vs. Yb_N diagram displaying middle Miocene volcanic rocks from the Southern CVZ (25–27°S); (c) La_N/Yb_N vs. Yb_N showing middle and upper Miocene volcanic rocks of the west portion of the Southern CVZ (67–69°). Geochemical data from Baker et al., 1987; Mpodozis et al., 1995; Kraemer et al., 1999; Trumbull et al., 1999; Siebel et al., 2001; Schnurr et al., 2007; Folkes et al., submitted for publication; this work.

melts were involved (e.g., Davidson and de Silva, 1995 and references therein).

In order to assess geochemical differences between middle Miocene volcanic centres from west to east, we compare the felsic ignimbrites and andesitic lavas of Southern CVZ (Baker et al., 1987; Mpodozis et al., 1995; Kraemer et al., 1999; Trumbull et al., 1999; Siebel et al., 2001; Schnurr et al., 2007) with the Luingo caldera and the mafic volcanic rocks of Vicuña Pampa (unpublished data) to the east. Differences are best visible when plotting Yb_N vs. Zr/Nb (Fig. 14b), thus indicating that these variables are efficient tools to constrain E-W geochemical differences for the middle Miocene in the Southern CVZ. Furthermore, the western samples have more pronounced HREE slopes (generally Gd_N/Yb_N = 2.64-4.6; exceptionally $Gd_N/Yb_N = 1.76$) than the eastern ones $(Gd_N/Yb_N = 1.31-1.78)$. Yb_N variations can be attributed to different crustal thickness and/or to different depths of magma sources within the mantle, i.e., garnetlherzolite stability field for the western samples and spinel-lherzolite stability field for the eastern ones. Moreover, the differences in the Zr/ Nb ratio may reflect differences in subduction-related signature as suggested by Siebel et al. (2001) and Caffe et al. (2002).

As we found differences in volcanic rocks at the same longitude for different ages and at different longitudes for the same age, we must now search for possible differences between rocks of different ages in the western portion of the Southern CVZ. The results show that the same HREE depletion that we found in the upper Miocene–Pliocene with respect to the middle Miocene in the eastern zone, are found between middle Miocene (11–16 Ma) with respect to lower Miocene (>16 Ma) in the west (samples from Siebel et al., 2001 and Schnurr et al., 2007, see Yb_N vs. La/Yb_N plot in Fig. 14c), with the exception of one isolated sample.

7. Discussion and concluding remarks

The middle Miocene Luingo volcanic rocks can be modelled by a mixture of ca. 80% mafic magmas and 20% of crustal melts by AFC-r process. On the other hand rocks from the nearby Cerro Galán caldera were modelled by mixtures of 50:50 mantle/crust melts (Kay et al., 2008). These differences in the crustal/mafic ratios of Cerro Galán ignimbrites with respect to the one of Luingo caldera, are better explained by the higher crustal component in those rocks (e.g., Kay et al., 2008), which have more evolved magmas and more radiogenic Sr isotopes. Moreover, a significant difference between Luingo and Cerro Galán rocks is the mantelic signature of the mafic component as shown in REE patterns. In the case of Cerro Galán caldera REE patterns indicate the presence of garnet as a residual phase in the source (e.g., Kay et al., 2008), whereas the mafic magmas that were involved in the Luingo caldera's magma evolution do not suggest the presence of garnet in the source of those magmas (see Fig. 14a). Steep REE patterns in the Puna-Altiplano volcanic rocks were interpreted as a result of the presence of garnet in the source but a more striking interpretation is that this points to a thickened crust (e.g., Coira and Kay, 1993). If this is true, then this thickened crust should have been present at least since the Lavas II event (upper Miocene), but not at the Luingo caldera event (13–12 Ma).

Although it is highly accepted that the processes leading to the crustal thickening and uplift of the Altiplano–Puna Plateau have been active since at least the Eocene–Oligocene along both western and eastern margins (e.g., Mpodozis et al., 2005; Carrapa et al., 2005; Arriagada et al., 2006; Hongn et al. 2007); palaeoclimatic and palaeoaltimetric data in the northern Altiplano (Gregory-Wodzicki, 2002; Garzione et al., 2006; Molnar and Garzione, 2007; Hoke and

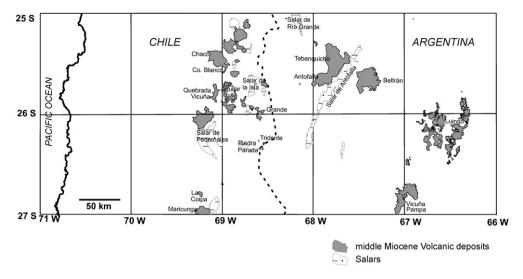


Fig. 15. Map showing areal distribution of middle Miocene (11–16 Ma) volcanic products within the Southern CVZ.

Garzione, 2008) point to a rapid uplift during the upper Miocene, as it was earlier suggested by James (1971) and Isacks (1988).

The comparison of geochemical data of middle Miocene volcanic rocks from west to east suggests that at this time the crust was thicker in the western than in the eastern portion of the Southern CVZ. It is tempting to interpret that the process of crustal thickening was accelerated in the western portion during the middle Miocene and then in the eastern portion of this part of the Puna during the upper Miocene (ca. 10 Ma).

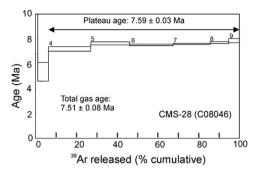
Besides the geochemical evidence, it is important to recognize the areal distribution of volcanic rocks in the Southern CVZ to compare the variations in the magnitude of volcanism from west to east. With the available data from the literature we constructed a map (Fig. 15) with the volcanic rocks of 11-16 Ma. From this figure it is evident that the volcanic deposits, but not necessarily the volcanic centres, have a NNE-SSW trending, which can be related with dominant ranges that were aligned in that direction during the middle Miocene and/or by post middle Miocene NNE-SSW trending faults (e.g., Mon and Salfity, 1995; Adelmann and Görler, 1998; Riller and Oncken, 2003) that would have produced this same configuration. Interestingly, volcanic occurrences appear to be restricted to the west and east Puna margins, and in a middle position following the trend of the structurallycontrolled Antofalla basin (Adelmann and Görler, 1998). Thus, we argue that it is not appropriate to refer to a main arc for the deposits in the western sector. Instead it is more prompting to envisage a broad arc, structurally controlled in both proto-Puna/Puna margins, in which the geochemical differences are related with variations in the crustal thicknesses and heterogeneous mantle magmatic sources from west to east, perhaps also related with these crustal thickness variations.

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Appendix A. Analytical techniques

Age determinations by ⁴⁰Ar/³⁹Ar method were carried out in two fresh biotite samples from Lavas II Unit at the geochronology laboratory in the University of Potsdam. Biotite separates were obtained at the Universidad Nacional de Salta laboratory in Argentina following standard techniques (i.e., crushing, milling, sieving, and mineral separation by a Frantz Magnetic Separator, followed by separation with paper shaking, cleaning in an ultrasonic bath, drying at 80°-100 °C over 24 h, and handpicking under a binocular microscope). Neutron activation of the samples was performed at the GKSS FRG-1 research reactor, Geesthacht Neutron Facility (GeNF), Germany, where the samples were irradiated over 96 h (4 days) with a fast neutron flux of 1×10^{12} n/cm²/s, with the aim of obtaining enough ³⁹Ar from the ³⁹K (n, p) ³⁹Ar nuclear reaction. As an age standard sample, sanidine from the Fish Canyon Tuff was irradiated together with our samples in order to obtain I values, that reflect the degree of neutron activation. The sanidine had been dated at the Geological Survey of Japan, providing an age of 27.5 Ma (Uto et al., 1997; Ishizuka et al., 2002). After one month, the samples had been cooled down at the GeNF and were ready for Ar isotopic analysis at the University of Potsdam. The laboratory's Ar analytical system integrates a New Wave Gantry Dual Wave laser ablation system with a 50 W CO₂ laser (wavelength: 10.6 µm) for heating samples and extracting gas. In addition, the system contains an ultra-high vacuum purification line with Zr-Al alloy SAES getters and a cold trap maintained at the freezing temperature of ethanol, as well as a Micromass 5400 sector-type noble gas mass spectrometer, which has a high sensitivity and an ultra-low background. Samples were analyzed by stepwise heating. The biotite grains were heated by a CO₂ laser beam for periods of between 1 and 1.5 min per step. The Ar isotopic analysis during stepwise heating was repeated until total fusion of the sample had occurred. The age was then calculated following the methodology of Uto et al. (1997). For each irradiation two biotite K-Ar age standards, SORI93 biotite (92.6 \pm 0.6 Ma; Sudo et al., 1998) and HD-B1 biotite (24.21 ± 0.32 Ma; Hess and Lippolt, 1994) were irradiated and the ⁴⁰Ar/³⁹Ar ages measured, in order to confirm the accuracy of the dating system in each irradiation. Results were entered in an Excel spreadsheet, where several corrections were performed including system blank, mass discrimination, postirradiation decay of ³⁷Ar and ³⁹Ar, and neutron-induced interferences from Ca and K. We thus obtained plateau, isochron, and total-gas ages. Steps for plateau determination were performed following McDougall and Harrison (1999) recommendations. Used criterion to determine



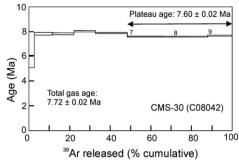


Fig. 16. Biotite ⁴⁰Ar/³⁹Ar age spectra obtained by stepwise heating experiments. Numbers refer to steps used for the calculation of the plateau ages. Uncertainties are given in 1 sigma without J value error.

the plateau steps are involved in McDougall and Harrison (1999) These are, (1) the ages of two consecutive steps should agree to within ± 2 sigma, excluding the J value error, (2) the total plateau length should include more than 50% of the total ^{39}Ar released, and (3) each degassing step making up the plateau should comprise at least 3% of the total ^{39}Ar released. $^{40}\text{Ar}/^{39}\text{Ar}$ results are shown in Fig. 16 and listed (± 1 sigma) in Table 8.

Chemical analyses of selected minerals were performed by WDS with a Cameca SX-50 electron microprobe equipped with 4 WDS spectrometers, operating at 25 nA and 16 KeV, in most cases with a beam size of 1 μ m, at the Instituto de Geociências, Universidade de Brasilia. Equipment was calibrated with a combination of natural and synthetic standards. The analytical errors are below 2% of the concentration for major elements and below 5% of the concentration for minor elements.

Selected samples were processed at the Universidad Nacional de Salta Geochemical laboratory, where they were crushed and milled. All analyses were performed on fresh samples; pumice fragments from ignimbrites were separated when possible. Major and some minor elements were determined by X-ray fluorescence (FRX) at the LANAIS laboratory (Universidad Nacional de Salta, Argentina) on a RIGAKU 2000 with a dispersive wavelength and Rb tube. PC gas flux. SC detector and monochromatic crystals: LIF PET, TAP and GE, Ground and homogenized samples were fused with lithium tetraborate as a flux for major element analyses. Trace elements were determined in rock powder pellets mixed with methacrylate and pressed at 1400 kg/ cm². Major and trace elements were analyzed by standard methods, using standards from Geological Survey of Japan and United States Geological Survey. Loss on ignition was determined by weight difference after heating at 1000 °C. Rare earth elements (REE) and other trace elements (ppm) were quantitatively and qualitatively determined at the ACME (Canada) laboratory by ICP-MS (Inductively Coupled Plasma-Mass Spectrometer) and others by ICP-AES (Inductively Coupled Plasma-Atomic Emission Spectrometer), samples were

Table 8 ⁴⁰Ar/³⁹Ar results on biotite samples Lavas II.

CMS-28	(C08046)																	
J = 0.001	1999	⁴⁰ Ar/ ³⁹	Ar	³⁷ Ar/ ³⁹ Ar		³⁶ Ar/ ³⁹ / ₄ (×10 ⁻³			K/Ca	40 _f (%)		³⁹ Ar _K f	raction	⁴⁰ Ar	*/ ³⁹ Ar _K		Age(±s) Ma)	
Steps	Laser outpu	t				(×10)			(//)	(%)				(ivia)	
1	1.6%	11,606.	313 ± 985.514	2.888 ± 1.5	72	38,480.	538 ± 32	269.494	0.20	2.	.03	0.01		236.	284 ± 43.12	24 6	697.90±	105.65
2	2.2%	322.	348 ± 1.698	0.006 ± 0.0	09	1080.9	952 ± 6 .	522	95.35	0.	.91	1.11		2.	928 ± 1.040	6	$10.53\pm$	3.75
3	2.8%	68.	034 ± 0.068	0.005 ± 0.0	01	225.2	214 ± 0.0	730	118.57	2.	.18	5.28		1.4	484 ± 0.209	9	$5.34\pm$	0.75
4	3.2%	17.	933 ± 0.031	0.003 ± 0.0	03	53.9	$934 \pm 0.$	185	230.46	11.	.13	20.67		1.5	996 ± 0.053	3	$7.19 \pm$	0.19
5	3.6%	11.	382 ± 0.013	0.029 ± 0.0	04	31.3	$360 \pm 0.$	115	20.11	18.	.62	19.12		2.	119 ± 0.034	4	$7.63 \pm$	0.12
6	4.0%	5.	766 ± 0.007	0.038 ± 0.0	03	12.4	$469 \pm 0.$	045	15.57	36.	.18	21.19		2.	086 ± 0.013	3	$7.51 \pm$	0.06
7	4.4%	4.	368 ± 0.005	0.017 ± 0.0	02	7.0	$648 \pm 0.$	032	35.41	48.	.31	18.17		2.	110 ± 0.010)	$7.59 \pm$	0.05
8	4.8%	4.	663 ± 0.006	0.013 ± 0.0	03	8.5	$536 \pm 0.$	065	46.12	45.	.94	9.18		2.	142 ± 0.019	9	$7.71 \pm$	80.0
9	5.2%	4.	364 ± 0.012	0.031 ± 0.0	07	7.3	394 ± 0 .	153	19.25	50.	.02	5.27		2.	183 ± 0.046	õ	$7.86 \pm$	0.17
J = 0.001	1999	⁴⁰ Ar/	³⁹ Ar	³⁷ Ar/ ³⁹ Ar		³⁶ Ar/ ³⁹ / (x10 ⁻³)			K/Ca	40 _f (%)		³⁹ Ar _K Fractio	n (%)	⁴⁰ Ar	*/ ³⁹ Ar _K		Age(±1:	s)
Steps	Laser outpu	t				(X10	,			(70,	,	Tractio	11 (%)			(ivia)	
1	1.6%	709.8	317 ± 4.657	0.101 ± 0.0	80	2361.93	6 ± 15.7	700	5.80	1.	.67	0.14		11.8	78 ± 0.823	4	12.34 ± 2	.90
2	2.2%	96.7	92 ± 0.169	0.135 ± 0.0	10	322.37	77 ± 0.57	70	4.37		.60	2.58		1.5	47 ± 0.139		5.57 ± 0	.50
3	2.8%	14.1	71 ± 0.022	0.055 ± 0.0	03	40.62	28 ± 0.09	97	10.71	15.	.33	8.71		2.1	72 ± 0.029		7.82 ± 0	.11
4	3.2%	7.1	73 ± 0.013	0.037 ± 0.0	03	16.94	15 ± 0.05	59	15.70	30.	.26	10.70		2.1	71 ± 0.018		7.81 ± 0	.07
5	3.6%	6.3	356 ± 0.007	0.053 ± 0.0	04		0.06 ± 0.06		11.10	35.		10.53		2.2	27 ± 0.018		8.01 ± 0	.07
6	4.0%		20 ± 0.004	0.070 ± 0.0	03	13.34	12 ± 0.03	30	8.43	35.		15.74		2.1	86 ± 0.009		7.87 ± 0	.05
7	4.4%	4.9	023 ± 0.007	0.043 ± 0.0	01	9.55	55 ± 0.01	3	13.52	42.		22.16		2.1	05 ± 0.006		7.58 ± 0	.04
8	4.8%	4.0	010 ± 0.003	0.035 ± 0.0	02	6.45	66 ± 0.02	23	16.93	52.		17.33		2.1	06 ± 0.007		7.58 ± 0	.04
9	5.2%	3.6	510 ± 0.006	0.020 ± 0.0	03	5.03	35 ± 0.02	27	29.24	58.	.85	12.11		2.1	24 ± 0.009		7.65 ± 0	.04
Sample	Unit	Location			Platea	nu age			Isochro	n age						Integr	ated age	
		Latitute	Longitude	dated	Age (Ma)	1σ	³⁹ Ar (%)	*steps	Age (Ma)	1σ		r/ ³⁶ Ar ercept	1σ	*steps	MSWD	Age (Ma)	1σ	Ca/I
CMS-28	Lavas II	27 06 47S	66 40 25.7 W	Biotite	7.59	0.03	93.6	6 (9)	7.66	0.06	293	3.72	0.92	6 (9)	1.52	7.51	0.08	0.00
		26 09 19S	66 42 51.5 W		7.60	0.02	51.6	3 (9)		0.07	292		2.12	3 (9)	1.27	7.72	0.02	

fused with lithium metaborate/tetraborate and solved and diluted in nitric acid. Detection limits are 0.1 to 0.01 ppm for REE and of 0.1 to 1 ppm for other trace elements.

The bulk-rock Sm-Nd isotopic analyses were carried out at the Geochronology Laboratory of the University of Brasília. Sample dissolution was carried in Teflon Savillex beakers or in Parrtype Teflon bombs. Sm and Nd extraction from whole-rock powders concentrates followed the technique of Gioia and Pimentel (2000), in which the separation of the REE as a group using cation-exchange columns precedes reversed-phase chromatography for the separation of Sm and Nd using columns loaded with HDEHP (di-2-ethylhexyl phosphoric acid) supported on Teflon powder. REE and Sm-Nd separation was performed using RE-Spec and Ln-Spec resins. A mixed ¹⁴⁹Sm-¹⁵⁰Nd spike was used. Sm and Nd samples were loaded onto Re filaments of a double filament assembly. Sm and Nd isotopic analyses were carried out using a Finnigan MAT-262 mass spectrometer. Uncertainties on Sm/Nd and 143Nd/144Nd ratios are considered to be better than $\pm 0.05\%$ (1 σ) and $\pm 0.003\%$ (1 σ), respectively, based on repeated analyses of international rock standards BCR-1 and BHVO-1. The ¹⁴³Nd/¹⁴⁴Nd ratios were normalized to a ¹⁴⁶Nd/¹⁴⁴Nd ratio of 0.7219. The Nd procedure blanks were smaller than 100 pg.

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