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Article in *Journal of Petrology* · November 2001

DOI: 10.1093/petrology/42.11.2109

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# Early Proterozoic Calc-Alkaline and Middle Proterozoic Tholeiitic Dyke Swarms from Central–Eastern Argentina: Petrology, Geochemistry, Sr–Nd Isotopes and Tectonic Implications

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RECEIVED JANUARY 2, 2001; REVISED TYPESCRIPT ACCEPTED MAY 11, 2001

*The Rio de La Plata craton in Argentina (Azul and Tandil regions) is characterized by Early Proterozoic (2.0 Ga) calc-alkaline and Middle Proterozoic (1.6 Ga) tholeiitic dyke swarms intruding the crystalline basement involved in the Transamazonian Orogeny (2.2–1.9 Ga). The calc-alkaline dykes have andesitic and rhyolitic compositions and trend east–west, whereas the tholeiitic dykes mainly trend N30°W and are represented by basalts with low (0.9–1.7 wt %) and high TiO<sub>2</sub> (up to 3.7 wt %). The calc-alkaline dykes have primitive mantle (PM)-normalized trace element patterns enriched in Rb, Ba, K, La, Ce and Nd, and significant negative Nb and Ti anomalies. These dykes are characterized by  $\epsilon'(Nd)$  values of  $-3$  to  $-4$ , similar to those of the EMI mantle component. Low-TiO<sub>2</sub> tholeiitic dykes have low incompatible-element (IE) contents and PM-IE patterns with slightly positive or negative*

*Nb spikes. They have variable  $\epsilon'(Nd)$  values ( $-0.5$  to  $12.1$ ), which mainly reflect derivation from a depleted source mantle. High-TiO<sub>2</sub> tholeiitic dykes have more enriched IE-PM patterns and are characterized by  $\epsilon'(Nd)$  values ( $-1.4$  to  $-7.5$ ) typical of an enriched source mantle. Chemical and isotopic data and melting modelling indicate that both calc-alkaline and tholeiitic dykes originated by different melting degrees of a heterogeneous source mantle, the variable IE enrichment of which may have occurred in Late Archaean to Early Proterozoic times. The emplacement of the calc-alkaline dykes is associated with the transtensional stage of the Transamazonian Orogeny, whereas the tholeiitic dykes reflect extensional tectonics succeeding the Transamazonian event. The calc-alkaline and tholeiitic dykes are similar in emplacement age and characteristics to metamorphosed granites and volcanic rocks outcropping within the*

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*Namaqua fold belts of southwestern Africa (Richtersveld and Witberg–Aggeneys–Gamsberg provinces); this may indicate that the Rio de La Plata craton and southwestern Africa were contiguous in Early–Middle Proterozoic times.*

KEY WORDS: *Argentina; geochemistry; petrology; Proterozoic dykes; Rio de La Plata craton*

## INTRODUCTION

Precambrian dyke swarms are common in cratonic blocks, providing useful correlations between different tectonic units and yielding information on related geodynamic processes (Halls, 1987; Parker *et al.*, 1990; Baer & Heimann, 1995).

The crystalline basement of the Azul and Tandil regions (Buenos Aires province, Argentina; Fig. 1) belongs to the Rio de La Plata craton and is intruded by unmetamorphosed Proterozoic mafic, intermediate and felsic dyke swarms of 2.0 and 1.6 Ga age, respectively (Teixeira *et al.*, in preparation). Similarly, the portion of the Rio de La Plata craton in Uruguay is characterized by unmetamorphosed Proterozoic dyke swarms (Fig. 1) dated at 1.7 Ga (Florida region; Bossi

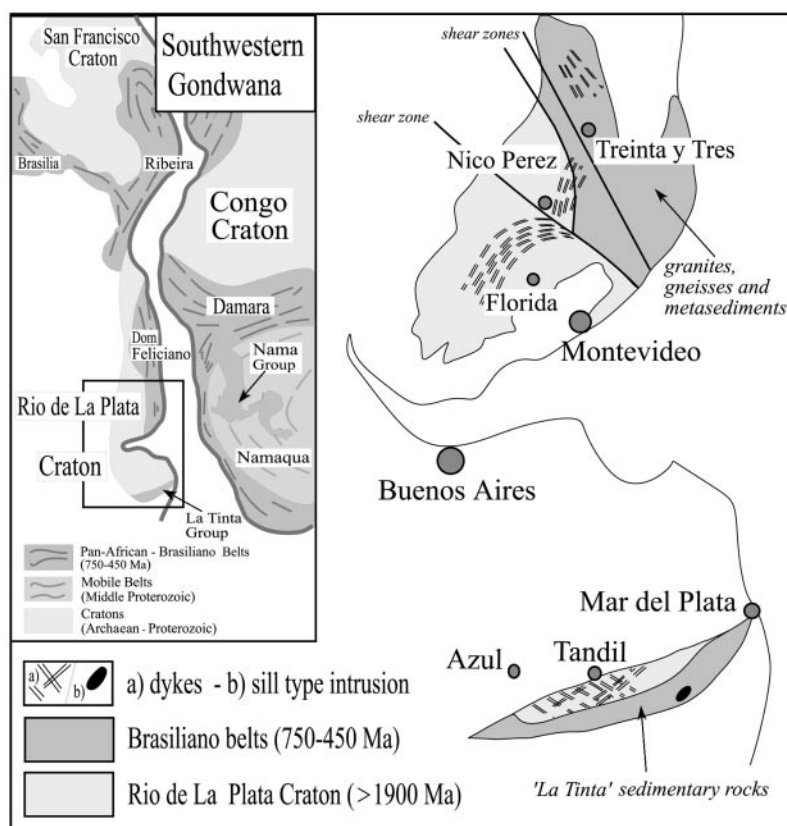
*et al.*, 1993; Teixeira *et al.*, 1999) and ~0.7 Ga (Nico Perez and Treinta y Tres regions; Mazzucchelli *et al.*, 1995; Rivalenti *et al.*, 1995; Girardi *et al.*, 1996).

Study of the Argentinian dykes is important for an understanding of source mantle characteristics in Early–Middle Proterozoic times, as well as for better insight into the processes responsible for craton stabilization in the southernmost part of the South American platform.

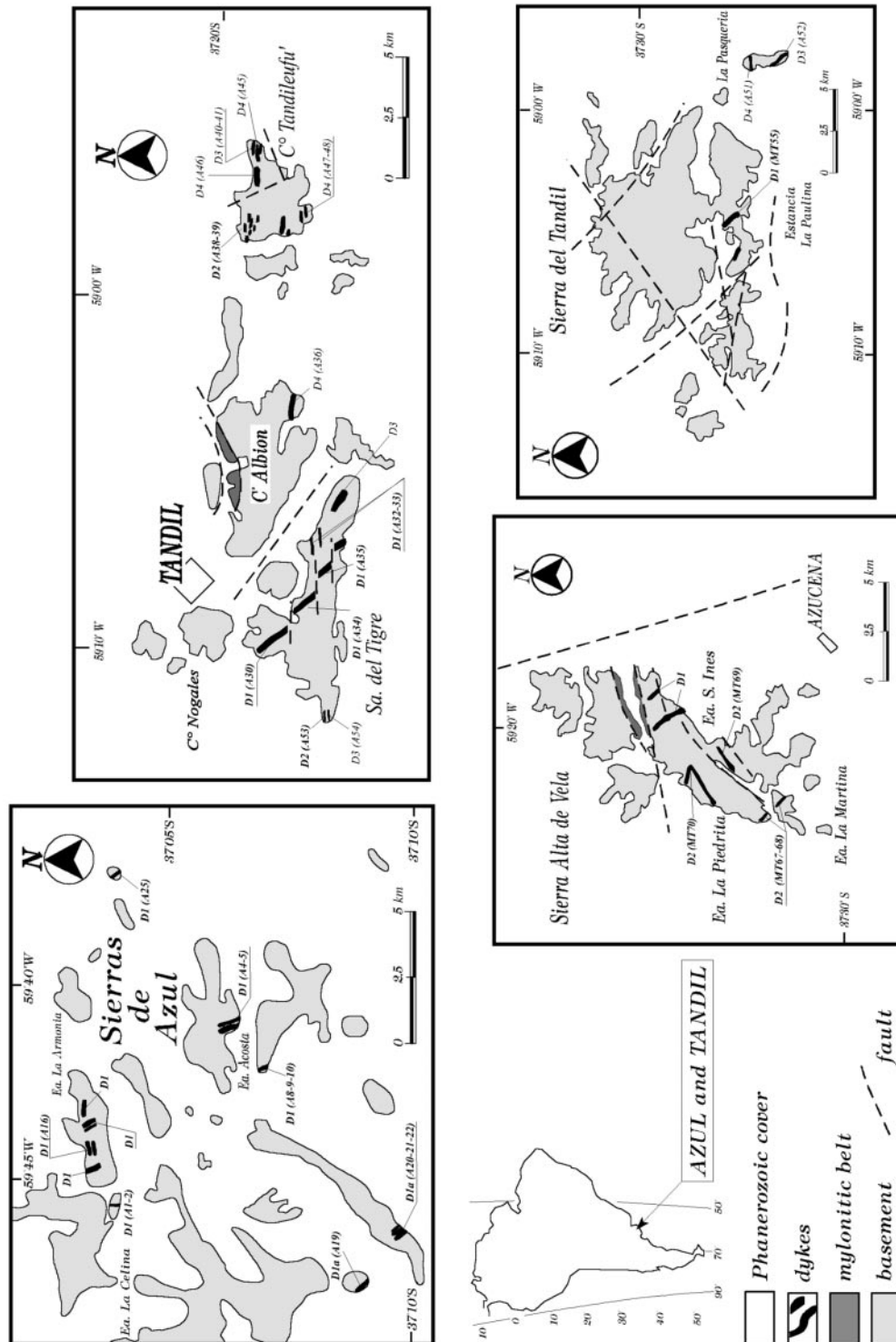
In this paper we characterize the Proterozoic calc-alkaline and tholeiitic dyke swarms of central–eastern Argentina in terms of petrological and genetic implications, and relationships between tectonics and magmatism. We show that these dyke swarms originated from depleted- to incompatible element-enriched mantle sources, and that their emplacement occurred during the transtensional to post-collisional stages of the Transamazonian Orogeny (2.2–1.9 Ga).

## GEOLOGICAL FRAMEWORK AND DYKE OCCURRENCE

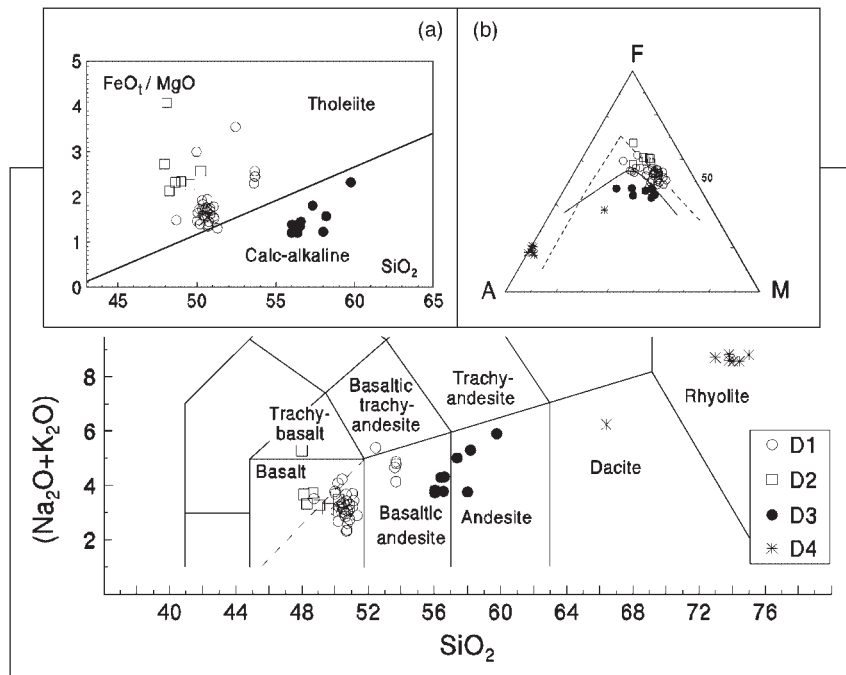
The southernmost Archaean–Proterozoic cratonic areas of South America (Fig. 1) are exposed in Uruguay



**Fig. 1.** Simplified geological sketch map of NE Argentina (Dalla Salda *et al.*, 1988) and Uruguay (Mazzucchelli *et al.*, 1995) showing Precambrian dyke occurrences. Inset: Pre-Atlantic South America–South Africa configuration modified after Trompette (1994).



**Fig. 2.** Simplified geological sketch map of dyke occurrences in Azul and Tandil regions (Argentina). D1, D1a, D2, tholeiitic basalts and basaltic andesites, respectively; D3, D4, calc-alkaline basaltic andesites, andesites, and rhyolites, respectively. Sample numbers in parentheses (see also Table 5).



**Fig. 3.** Classification diagrams of the Azul and Tandil dyke swarms in terms of TAS (Zanettin, 1984; Le Bas *et al.*, 1986),  $\text{SiO}_2$  vs  $\text{Fe}_t/\text{MgO}$  (Miyashiro, 1974) and AFM: continuous curve (calc-alkaline–tholeiitic boundary) after Irvine & Baragar (1971); dashed curve (tholeiitic trend) after Macdonald & Katsura (1964). D1, D2, tholeiitic suite; D3, D4, calc-alkaline suites.

*Table 1: Plagioclase microprobe compositions of tholeiitic (D1) and calc-alkaline (D3) dykes from the Azul and Tandil regions*

| Rock:                     | ThB    | ThB    | ThB    | ThB   | BAnd   | ThB    | ThB    | And    |
|---------------------------|--------|--------|--------|-------|--------|--------|--------|--------|
| Group:                    | D1     | D1     | D1     | D1    | D1     | D1     | D1     | D3     |
| Sample:                   | A2     | A5     | A8     | A16   | A19    | A25    | A34    | A54    |
| Plagioclase:              | r      | c1     | c2     | r     | r      | c      | r      | r      |
| $\text{SiO}_2$            | 52.83  | 50.28  | 55.40  | 53.66 | 50.98  | 53.21  | 53.32  | 51.07  |
| $\text{TiO}_2$            | 0.05   | 0.05   | 0.05   | 0.06  | 0.04   | 0.06   | 0.12   | 0.04   |
| $\text{Al}_2\text{O}_3$   | 30.01  | 31.59  | 28.30  | 29.06 | 31.05  | 29.85  | 29.26  | 29.85  |
| $\text{FeO}_{\text{tot}}$ | 0.76   | 0.60   | 0.46   | 0.58  | 0.76   | 0.28   | 0.98   | 1.51   |
| CaO                       | 12.43  | 14.33  | 10.42  | 11.45 | 13.72  | 12.22  | 11.70  | 12.71  |
| $\text{Na}_2\text{O}$     | 4.43   | 3.35   | 5.64   | 4.95  | 3.53   | 4.35   | 4.84   | 2.43   |
| $\text{K}_2\text{O}$      | 0.12   | 0.02   | 0.08   | 0.10  | 0.27   | 0.47   | 0.04   | 2.54   |
| Total                     | 100.63 | 100.22 | 100.35 | 99.86 | 100.35 | 100.44 | 100.26 | 100.15 |
| Or (wt %)                 | 0.71   | 0.12   | 0.46   | 0.58  | 1.63   | 2.75   | 0.23   | 15.24  |
| Ab (wt %)                 | 37.53  | 28.49  | 47.77  | 42.17 | 30.02  | 36.72  | 41.24  | 20.82  |
| An (wt %)                 | 61.76  | 71.40  | 51.78  | 57.26 | 68.36  | 60.53  | 58.53  | 63.94  |

ThB, tholeiitic basalt; BAnd, basaltic andesite; And, andesite;  $\text{FeO}_{\text{tot}}$ , total Fe as  $\text{FeO}$ ; c, core; r, rim.

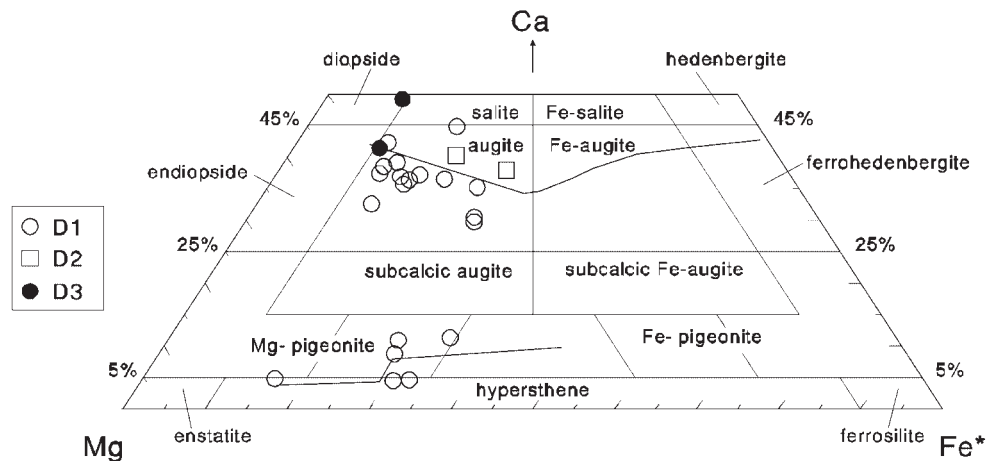


Fig. 4. Ca-rich and Ca-poor pyroxenes of Azul and Tandil basic-intermediate dyke swarms. Continuous curves, Skaergaard trends.  $\circ$ ,  $\square$  and  $\bullet$ , D1, D2 (tholeiitic) and D3 (calc-alkaline) dykes, respectively.  $\text{Fe}^* = \text{Fe}^{2+} + \text{Mn} + \text{Fe}^{3+}$ .

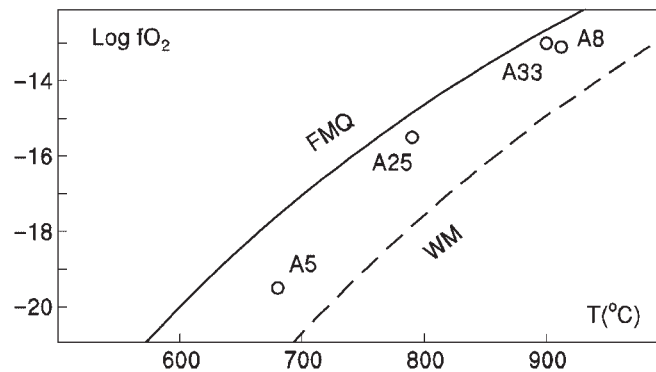


Fig. 5.  $\text{Log}(f\text{O}_2)$  vs  $T$  ( $^{\circ}\text{C}$ ) for homogeneous magnetite-ilmenite pairs from D1 tholeiitic dykes. FMQ, fayalite-magnetite-quartz; WM, wüstite-magnetite.

and Argentina (Azul and Tandil regions) and belong to the Rio de La Plata craton, characterized by terranes involved in the tectonic-metamorphic and magmatic events of the Transamazonian Orogeny (2.2–1.9 Ga) and the Brasiliano cycle (0.8–0.5 Ga). The Rio de La Plata crystalline basement is mainly composed of gneisses, migmatite and granitoids. The Precambrian unmetamorphosed dyke swarms in Uruguay (Fig. 1; Bossi *et al.*, 1993; Mazzucchelli *et al.*, 1995; Rivalenti *et al.*, 1995; Girardi *et al.*, 1996) intrude the crystalline terranes affected by both Transamazonian and Brasiliano events, and are dated as Early and Late Proterozoic, respectively: Florida 1.73 and 1.77 Ga (Teixeira *et al.*, 1999) and Nico Perez and Treinta y Tres 0.7–0.6 Ga (Mazzucchelli *et al.*, 1995; Rivalenti *et al.*, 1995; Girardi *et al.*, 1996). In contrast, the Precambrian unmetamorphosed dyke swarms in Argentina intrude exclusively crystalline terranes of the Rio de La Plata craton and are dated at 2.0 Ga and 1.6 Ga (biotite  $^{40}\text{Ar}/^{39}\text{Ar}$  and baddeleyite U–Pb ages,

respectively; Teixeira *et al.*, in preparation). It is also noted that the Late Proterozoic sedimentary rocks of the Argentina ‘La Tinta Group’ (0.8–0.5 Ga, Fig. 1) in the southern Azul–Tandil regions are locally (Sierra de Los Barrientos) intruded by tholeiitic sills (0.5 Ga; Rapela *et al.*, 1974).

The stratigraphy of the Azul and Tandil regions (Fig. 2) is not well constrained in terms of radiometric ages. The oldest rocks are gneisses, migmatites, mica-schists and medium-grade amphibolites (Teruggi *et al.*, 1974). They are intruded by granitoid bodies whose Rb/Sr ages cluster around 2.2–2.0 and 1.8–1.6 Ga (Varela *et al.*, 1988; Dalla Salda *et al.*, 1992). The ‘old granitoids’ (gabbroic to granitic rocks) outcrop in the northern part of the ‘Azul and Tandil Sierras’; the ‘young granitoids’ (mainly monzogranites) are located in the southern part of these ‘Sierras’. The emplacement of both ‘old’ and ‘young’ granitoids has been related to the syn/late- to post-collisional stages, respectively, of the Transamazonian Orogeny (Varela *et al.*, 1985,

Table 2: *Ca-rich (cpx) and Ca-poor (pig and opx) pyroxene microprobe compositions of the tholeiitic (D1, D1a and D2) and calc-alkaline (D3) dykes from the Azul and Tandil regions*

| Rock:                            | ThB    |       | ThB    |       | ThB    |        |       | ThB    |        | ThB    |        |       |       |
|----------------------------------|--------|-------|--------|-------|--------|--------|-------|--------|--------|--------|--------|-------|-------|
| Group:                           | D1     |       | D1     |       | D1     |        |       | D1     |        | D1     |        |       |       |
| Sample:                          | A2     |       | A5     |       | A8     |        |       | A16    |        | A25    |        |       |       |
| Pyroxene:                        | cpx/c  | cpx/r | cpx/c  | cpx/r | cpx/c  | cpx/r  | pig/r | cpx/r  | opx/r  | cpx/c  | cpx/r  | opx/c | opx/r |
| SiO <sub>2</sub>                 | 50.99  | 51.09 | 52.57  | 50.34 | 52.94  | 52.41  | 53.52 | 52.16  | 55.37  | 52.13  | 51.57  | 53.03 | 52.29 |
| TiO <sub>2</sub>                 | 0.49   | 0.55  | 0.09   | 0.70  | 0.34   | 0.46   | 0.23  | 0.42   | 0.12   | 0.52   | 0.57   | 0.30  | 0.39  |
| Al <sub>2</sub> O <sub>3</sub>   | 4.46   | 1.94  | 0.35   | 1.69  | 2.20   | 1.91   | 0.97  | 3.59   | 2.71   | 2.07   | 2.19   | 0.94  | 1.44  |
| FeO <sub>tot</sub>               | 6.78   | 16.53 | 11.40  | 15.49 | 7.80   | 10.15  | 17.90 | 9.37   | 10.42  | 10.87  | 12.84  | 19.31 | 20.22 |
| MnO                              | 0.16   | 0.42  | 0.21   | 0.37  | 0.20   | 0.21   | 0.41  | 0.25   | 0.20   | 0.27   | 0.28   | 0.37  | 0.47  |
| MgO                              | 16.06  | 14.43 | 13.09  | 13.58 | 17.59  | 17.02  | 22.32 | 16.65  | 29.23  | 16.05  | 14.92  | 23.30 | 22.36 |
| CaO                              | 20.31  | 14.60 | 22.18  | 16.99 | 19.22  | 17.65  | 4.33  | 17.93  | 2.47   | 18.36  | 17.83  | 2.27  | 2.31  |
| Na <sub>2</sub> O                | 0.28   | 0.17  | 0.13   | 0.24  | 0.23   | 0.25   | 0.07  | 0.21   | 0.04   | 0.18   | 0.23   | 0.01  | 0.05  |
| Cr <sub>2</sub> O <sub>3</sub>   | 0.58   | 0.00  | 0.00   | 0.00  | 0.24   | 0.08   | 0.02  | 0.09   | 0.35   | 0.03   | 0.04   | 0.10  | 0.05  |
| Total                            | 100.10 | 99.74 | 100.02 | 99.40 | 100.76 | 100.14 | 99.77 | 100.67 | 100.91 | 100.48 | 100.47 | 99.63 | 99.58 |
| Fe <sub>2</sub> O <sub>3</sub> * | 1.61   | 0.69  | 0.68   | 1.87  | 1.58   | 1.56   | 0.00  | 0.55   | 0.00   | 1.47   | 1.63   | 0.55  | 0.96  |
| Si                               | 1.869  | 1.937 | 1.981  | 1.919 | 1.926  | 1.932  | 1.977 | 1.910  | 1.940  | 1.926  | 1.921  | 1.962 | 1.945 |
| Al(IV)                           | 0.131  | 0.063 | 0.015  | 0.076 | 0.074  | 0.068  | 0.023 | 0.090  | 0.060  | 0.074  | 0.079  | 0.038 | 0.055 |
| Total                            | 2.000  | 2.000 | 1.996  | 1.995 | 2.000  | 2.000  | 2.000 | 2.000  | 2.000  | 2.000  | 2.000  | 2.000 | 2.000 |
| Al(VI)                           | 0.062  | 0.024 | 0.000  | 0.000 | 0.020  | 0.015  | 0.019 | 0.065  | 0.052  | 0.016  | 0.017  | 0.003 | 0.008 |
| Fe <sup>2+</sup>                 | 0.163  | 0.504 | 0.340  | 0.440 | 0.194  | 0.270  | 0.553 | 0.272  | 0.305  | 0.295  | 0.354  | 0.582 | 0.602 |
| Fe <sup>3+</sup>                 | 0.044  | 0.020 | 0.019  | 0.054 | 0.043  | 0.043  | 0.000 | 0.015  | 0.000  | 0.041  | 0.046  | 0.015 | 0.027 |
| Cr                               | 0.017  | 0.000 | 0.000  | 0.000 | 0.007  | 0.002  | 0.001 | 0.003  | 0.010  | 0.001  | 0.001  | 0.003 | 0.001 |
| Mg                               | 0.878  | 0.816 | 0.735  | 0.772 | 0.954  | 0.935  | 1.229 | 0.909  | 1.526  | 0.884  | 0.829  | 1.285 | 1.240 |
| Mn                               | 0.005  | 0.013 | 0.007  | 0.012 | 0.006  | 0.007  | 0.013 | 0.008  | 0.006  | 0.009  | 0.009  | 0.012 | 0.015 |
| Ti                               | 0.013  | 0.016 | 0.003  | 0.020 | 0.009  | 0.013  | 0.006 | 0.011  | 0.003  | 0.015  | 0.016  | 0.008 | 0.011 |
| Ca                               | 0.798  | 0.593 | 0.895  | 0.694 | 0.749  | 0.697  | 0.171 | 0.703  | 0.093  | 0.727  | 0.712  | 0.090 | 0.092 |
| Na                               | 0.020  | 0.013 | 0.009  | 0.018 | 0.016  | 0.018  | 0.005 | 0.015  | 0.003  | 0.013  | 0.017  | 0.001 | 0.004 |
| Total                            | 2.000  | 1.999 | 2.008  | 2.010 | 1.998  | 2.000  | 1.997 | 2.001  | 1.998  | 2.001  | 2.001  | 1.999 | 2.000 |
| Ca                               | 42.27  | 30.47 | 44.84  | 35.19 | 38.49  | 35.71  | 8.70  | 36.86  | 4.82   | 37.17  | 36.51  | 4.54  | 4.66  |
| Mg                               | 46.50  | 41.93 | 36.82  | 39.15 | 49.02  | 47.90  | 62.51 | 47.67  | 79.07  | 45.19  | 42.51  | 64.77 | 62.75 |
| Fe†                              | 11.23  | 27.60 | 18.34  | 25.66 | 12.49  | 16.39  | 28.79 | 15.47  | 16.11  | 17.64  | 20.97  | 30.70 | 32.59 |

1988; Dalla Salda *et al.*, 1992). Mylonitic belts associated with the 'old granitoids' (Fig. 2) may reflect wrench or overthrusting tectonics (Dalla Salda, 1981; Ramos *et al.*, 1990; Dalla Salda *et al.*, 1992).

The southern parts of the Azul and Tandil 'Sierras' are covered by Late Proterozoic sedimentary rocks of the 'La Tinta Group' (up to 400 m thick). These rocks correspond to those of the 'Nama Group' in South Africa (Fig. 1), suggesting a common sedimentary basin origin (Dalla Salda, 1982; Kawashita *et al.*, 1999).

The unmetamorphosed dykes studied here intrude the crystalline basement terranes and the 'old' and 'young granitoids', and mainly strike N30°W in Azul,

and N30°W and east-west in Tandil. The dykes (Fig. 2) are divided into two main groups according to field and chemical features (see below). The dykes of the first group (D1 and D2) mainly trend N30°W and have tholeiitic characteristics; those of the second group (D3 and D4) have an east-west strike and a calc-alkaline signature (see Fernandez & Echeveste, 1995).

Most of the tholeiitic dykes are low in TiO<sub>2</sub> (<1.7 wt %, D1), although a few have TiO<sub>2</sub> up to 3.7 wt % (D2). D1 dykes occur in both the Azul and Tandil 'Sierras', are subvertical, vary in thickness from 10 to 50 m, and extend for distances of up to 5 km. They are fine grained at the border and coarse grained

| Rock:                            | ThB    |       | ThB   |        | BAnd   |       |       | Trb    |       | BAnd   |       |
|----------------------------------|--------|-------|-------|--------|--------|-------|-------|--------|-------|--------|-------|
| Group:                           | D1     |       | D1    |        | D1a    |       |       | D2     |       | D3     |       |
| Sample:                          | A33    |       | A34   |        | A19    |       |       | A38    |       | A41    |       |
| Pyroxene:                        | cpx/c  | cpx/r | cpx/c | cpx/r  | cpx/r  | pig/c | pig/r | cpx/c  | cpx/r | cpx/c  | cpx/r |
| SiO <sub>2</sub>                 | 52.05  | 51.85 | 52.29 | 53.26  | 51.76  | 53.32 | 52.44 | 51.06  | 49.38 | 54.03  | 52.89 |
| TiO <sub>2</sub>                 | 0.42   | 0.47  | 0.25  | 0.25   | 0.45   | 0.19  | 0.34  | 1.25   | 1.39  | 0.05   | 0.34  |
| Al <sub>2</sub> O <sub>3</sub>   | 2.71   | 2.39  | 2.88  | 1.78   | 1.23   | 1.05  | 1.00  | 2.27   | 2.89  | 0.29   | 2.04  |
| FeO <sub>tot</sub>               | 8.41   | 10.18 | 7.73  | 8.83   | 17.17  | 17.37 | 20.79 | 12.38  | 16.16 | 5.88   | 6.51  |
| MnO                              | 0.26   | 0.29  | 0.13  | 0.17   | 0.34   | 0.47  | 0.52  | 0.36   | 0.42  | 0.21   | 0.14  |
| MgO                              | 16.60  | 16.44 | 17.50 | 19.20  | 14.82  | 21.70 | 19.02 | 13.62  | 11.45 | 14.78  | 16.80 |
| CaO                              | 19.25  | 17.75 | 18.19 | 16.30  | 14.50  | 5.38  | 5.48  | 19.43  | 17.62 | 24.45  | 20.20 |
| Na <sub>2</sub> O                | 0.25   | 0.21  | 0.21  | 0.15   | 0.11   | 0.02  | 0.07  | 0.23   | 0.17  | 0.26   | 0.23  |
| Cr <sub>2</sub> O <sub>3</sub>   | 0.21   | 0.12  | 0.04  | 0.18   | 0.04   | 0.07  | 0.00  | 0.14   | 0.08  | 0.33   | 0.50  |
| Total                            | 100.16 | 99.70 | 99.22 | 100.12 | 100.42 | 99.57 | 99.66 | 100.74 | 99.56 | 100.28 | 99.65 |
| Fe <sub>2</sub> O <sub>3</sub> * | 1.61   | 1.19  | 0.66  | 1.00   | 0.83   | 0.00  | 0.00  | 0.91   | 0.08  | 0.34   | 0.00  |
| Si                               | 1.913  | 1.924 | 1.928 | 1.944  | 1.952  | 1.976 | 1.975 | 1.908  | 1.898 | 1.992  | 1.947 |
| Al(IV)                           | 0.087  | 0.076 | 0.072 | 0.056  | 0.048  | 0.024 | 0.025 | 0.092  | 0.102 | 0.008  | 0.053 |
| Total                            | 2.000  | 2.000 | 2.000 | 2.000  | 2.000  | 2.000 | 2.000 | 2.000  | 2.000 | 2.000  | 2.000 |
| Al(VI)                           | 0.030  | 0.029 | 0.053 | 0.021  | 0.007  | 0.022 | 0.019 | 0.008  | 0.029 | 0.005  | 0.036 |
| Fe <sup>2+</sup>                 | 0.214  | 0.283 | 0.220 | 0.242  | 0.518  | 0.538 | 0.655 | 0.361  | 0.517 | 0.172  | 0.200 |
| Fe <sup>3+</sup>                 | 0.045  | 0.033 | 0.018 | 0.028  | 0.024  | 0.000 | 0.000 | 0.026  | 0.002 | 0.009  |       |
| Cr                               | 0.006  | 0.004 | 0.001 | 0.005  | 0.001  | 0.002 | 0.000 | 0.004  | 0.002 | 0.010  | 0.014 |
| Mg                               | 0.910  | 0.909 | 0.962 | 1.045  | 0.833  | 1.199 | 1.068 | 0.759  | 0.656 | 0.812  | 0.922 |
| Mn                               | 0.008  | 0.009 | 0.004 | 0.005  | 0.011  | 0.015 | 0.017 | 0.012  | 0.014 | 0.006  | 0.004 |
| Ti                               | 0.012  | 0.013 | 0.007 | 0.007  | 0.013  | 0.005 | 0.010 | 0.035  | 0.040 | 0.001  | 0.009 |
| Ca                               | 0.758  | 0.706 | 0.719 | 0.638  | 0.586  | 0.214 | 0.221 | 0.778  | 0.726 | 0.966  | 0.797 |
| Na                               | 0.018  | 0.015 | 0.015 | 0.010  | 0.008  | 0.001 | 0.005 | 0.017  | 0.013 | 0.019  | 0.016 |
| Total                            | 2.001  | 2.001 | 1.999 | 2.001  | 2.001  | 1.996 | 1.995 | 2.000  | 1.999 | 2.000  | 1.998 |
| Ca                               | 39.17  | 36.39 | 37.39 | 32.58  | 29.72  | 10.89 | 11.27 | 40.19  | 37.91 | 49.16  | 41.45 |
| Mg                               | 47.03  | 46.86 | 50.03 | 53.37  | 42.24  | 60.99 | 54.46 | 39.20  | 34.26 | 41.32  | 47.95 |
| Fe†                              | 13.80  | 16.75 | 12.58 | 14.04  | 28.04  | 28.13 | 34.27 | 20.61  | 27.83 | 9.52   | 10.61 |

Fe<sub>2</sub>O<sub>3</sub>\* calculated according to Papike *et al.* (1974). Abbreviations as in Table 1, and Trb, trachybasalt. Fe† is Fe<sup>2+</sup> + Mn + Fe<sup>3+</sup>.

in the centre. D2 dykes are exposed only in the Tandil 'Sierras', are subvertical, and trend east–west in the 'Don Pedro' quarry (Cerro Tandileufú) and in the southwestern part of the 'Sierra del Tigre', and NW–SE and ~NE–SW in the 'Sierra Alta de Vela'. D2 dykes vary in thickness from 0.5 to 10 m and are medium to fine grained.

The calc-alkaline dykes have andesitic (D3) and rhyolitic (D4) composition. D3 dykes outcrop in the Tandil 'Sierras', mostly along the belt comprising Sierra del Tigre, Cerro Albion and Cerro Tandileufú. These dykes, 0.5–10 m thick, strike east–west, are vertical to moderately inclined (45°S), and have a medium–fine

texture. D4 rhyolitic dykes are associated with, and sometimes intruded by, D3 dykes (Tandileufú quarry). D4 dykes strike east–west, are subvertical, and may reach 30 m in thickness (Brigitte quarry).

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Total alkalis–SiO<sub>2</sub> (TAS) and total alkalis–total Fe–MgO (AFM) diagrams (Fig. 3) reveal that dykes with SiO<sub>2</sub> <54 wt % (D1 and D2) are characterized by pronounced FeO<sub>tot</sub> enrichment and straddle the tholeiitic trend of

Table 3: Olivine microprobe compositions of tholeiitic (D1) dykes from the Azul and Tandil regions

|                                |        |       |        |        |        |        |        |
|--------------------------------|--------|-------|--------|--------|--------|--------|--------|
| Rock:                          | ThB    |       |        | ThB    |        |        |        |
| Sample:                        | A5     |       |        | A34    |        |        |        |
| Group:                         | D1     |       |        | D1     |        |        |        |
| Olivine:                       | c1     | c2    | r2     | c1     | r1     | c2     | r2     |
| SiO <sub>2</sub>               | 36.56  | 37.40 | 36.68  | 38.99  | 38.75  | 38.79  | 39.07  |
| Al <sub>2</sub> O <sub>3</sub> | 0.00   | 0.00  | 0.01   | 0.03   | 0.04   | 0.05   | 0.00   |
| FeO <sub>tot</sub>             | 31.77  | 28.14 | 30.77  | 20.12  | 20.29  | 20.60  | 20.02  |
| MnO                            | 0.40   | 0.37  | 0.41   | 0.39   | 0.32   | 0.30   | 0.19   |
| MgO                            | 31.18  | 33.66 | 32.16  | 40.94  | 40.47  | 40.50  | 40.56  |
| CaO                            | 0.22   | 0.28  | 0.17   | 0.26   | 0.33   | 0.26   | 0.25   |
| Total                          | 100.12 | 99.85 | 100.20 | 100.73 | 100.20 | 100.50 | 100.09 |
| Si                             | 0.997  | 1.003 | 0.994  | 0.996  | 0.997  | 0.996  | 1.003  |
| Al(IV)                         | 0.000  | 0.000 | 0.000  | 0.001  | 0.001  | 0.001  | 0.000  |
| Total                          | 0.997  | 1.003 | 0.994  | 0.997  | 0.998  | 0.997  | 1.003  |
| Fe <sup>2+</sup>               | 0.724  | 0.631 | 0.698  | 0.430  | 0.436  | 0.442  | 0.430  |
| Mn                             | 0.009  | 0.009 | 0.010  | 0.008  | 0.007  | 0.007  | 0.004  |
| Mg                             | 1.267  | 1.346 | 1.299  | 1.560  | 1.552  | 1.550  | 1.552  |
| Ca                             | 0.006  | 0.008 | 0.005  | 0.007  | 0.009  | 0.007  | 0.007  |
| Total                          | 2.006  | 1.994 | 2.012  | 2.005  | 2.004  | 2.006  | 1.993  |
| Fo                             | 63.63  | 68.08 | 65.07  | 78.39  | 78.05  | 77.80  | 78.32  |
| Fa                             | 36.37  | 31.92 | 34.93  | 21.61  | 21.95  | 22.20  | 21.68  |

Abbreviations as in Table 1. Fo, forsterite; Fa, fayalite.

Hawaii (Macdonald & Katsura, 1964); dykes with SiO<sub>2</sub> >56 wt % (D3 and D4) plot in the calc-alkaline field of Irvine & Baragar (1971) (Fig. 3b). These data and the SiO<sub>2</sub>–(FeO<sub>tot</sub>/MgO) relationships (Fig. 3a) indicate that the Azul–Tandil dykes belong to different magmatic suites with tholeiitic and calc-alkaline characteristics, respectively.

The dykes of the tholeiitic suite mainly correspond to basalts and basaltic andesites, the high-TiO<sub>2</sub> (HTi) samples (D2) having lower SiO<sub>2</sub> and higher incompatible element abundances (see below) than the low-TiO<sub>2</sub> (LTi) analogues (D1). The dykes of the calc-alkaline suite plot in the fields of basaltic andesite and andesite (D3), and in those of dacite and rhyolite (D4).

LTi tholeiitic dykes (D1) have a subophitic texture dominated by plagioclase (Table 1) and pyroxenes (Table 2). Plagioclase is labradoritic (An<sub>71–52</sub>) and may be altered to clay minerals and sericite. Augite (Wo = 42–30), often associated with orthopyroxene (Wo = 4.5–4.8) or pigeonite (Wo = 8.7–11.3), has total compositional variations that mimic those of the Skaergaard intrusion (Fig. 4). The TiO<sub>2</sub> content of augites is low and ranges from 0.25 to 0.70 wt %.

Pyroxenes may be partly or completely replaced by hornblende, tremolite and chlorite. In general, olivine is scarce and restricted to low-Ca pyroxene-free samples. Olivine phenocryst compositions (Table 3) are either virtually constant (sample A34, Fo = 77.8–78.4%) or have a small range (sample A5) from 63.6 to 68.1%. The forsterite content of tholeiitic basalt A34 is in equilibrium ( $K_{\text{dFe-Mg}} = 0.30$ ; Takahashi & Kushiro, 1983) with the bulk-rock composition  $\{100 \times [\text{at. Mg}/(\text{Mg} + \text{Fe}^{2+})], \text{mg-number} = 51.3\}$ , whereas this is not the case for the olivine of tholeiitic basalt A5 (mg-number = 60.2), which would require equilibrium olivine with Fo = 83%. Magnetite (ulvöspinel 26–63%) and ilmenite (R<sub>2</sub>O<sub>3</sub> 2.3–9.4%) are usually confined to the groundmass (Table 4). Quartz, apatite and epidote are common accessory phases. The crystallization temperatures (Kretz, 1982) of D1 pyroxenes range from 1130 to 1205°C, with Cpx–Opx equilibration at 1121°C (Iacumin, 1998). According to the Loucks (1996) thermometer, the olivine–augite equilibrium temperature for sample A34 is ~1165°C. The homogeneous magnetite–ilmenite pairs yielded subsolidus equilibration temperatures ranging from 680 to 900°C.

Table 4: Magnetite and ilmenite microprobe compositions of tholeiitic (D1, D1a, D2) and calc-alkaline (D3, D4) dykes from the Azul and Tandil regions

|                                 |        |        |        |        |        |        |        |        |
|---------------------------------|--------|--------|--------|--------|--------|--------|--------|--------|
| Rock:                           | ThB    | ThB    | ThB    | ThB    | ThB    | ThB    | ThB    |        |
| Sample:                         | A5     | A8     | A25    | A33    | A34    | A2     | A16    |        |
| Group:                          | D1     | D1     | D1     | D1     | D1     | D1     | D1     |        |
|                                 |        |        |        |        |        |        |        |        |
| Magnetite                       |        |        |        |        |        |        |        |        |
| SiO <sub>2</sub>                | 0.10   | 0.13   | 0.25   | 0.03   | 0.26   | 0.11   | 0.03   |        |
| TiO <sub>2</sub>                | 15.42  | 21.78  | 15.78  | 19.12  | 9.03   | 9.69   | 10.74  |        |
| Al <sub>2</sub> O <sub>3</sub>  | 1.71   | 2.23   | 1.79   | 2.86   | 1.80   | 1.63   | 0.80   |        |
| FeO <sub>tot</sub>              | 77.59  | 71.76  | 72.28  | 74.64  | 84.50  | 84.67  | 84.41  |        |
| MnO                             | 0.67   | 0.02   | 0.46   | 0.00   | 0.14   | 0.02   | 0.43   |        |
| MgO                             | 0.04   | 0.04   | 0.54   | 0.00   | 0.29   | 0.00   | 0.05   |        |
| CaO                             | 0.02   | 0.16   | 5.78   | 0.14   | 0.76   | 0.12   | 0.07   |        |
| Cr <sub>2</sub> O <sub>3</sub>  | 0.04   | 0.00   | 0.05   | 0.03   | 0.00   | 0.18   | 0.04   |        |
| Total                           | 95.59  | 96.12  | 96.93  | 96.82  | 96.78  | 96.42  | 96.57  |        |
| FeO                             | 44.81  | 51.08  | 38.38  | 49.09  | 39.17  | 40.72  | 41.03  |        |
| Fe <sub>2</sub> O <sub>3</sub>  | 36.42  | 22.97  | 37.67  | 28.38  | 50.37  | 48.83  | 48.19  |        |
| Total*                          | 99.23  | 98.42  | 100.71 | 99.66  | 101.82 | 101.30 | 101.39 |        |
| % Ulv.                          | 44.83  | 63.41  | 44.87  | 54.94  | 26.40  | 27.95  | 30.51  |        |
|                                 |        |        |        |        |        |        |        |        |
| Rock:                           | ThB    | ThB    | ThB    | ThB    | BAnd   | Trb    | BAnd   | Dacite |
| Sample:                         | A5     | A8     | A25    | A33    | A19    | A38    | A41    | A36    |
| Group:                          | D1     | D1     | D1     | D1     | D1a    | D2     | D3     | D4     |
|                                 |        |        |        |        |        |        |        |        |
| Ilmenite                        |        |        |        |        |        |        |        |        |
| SiO <sub>2</sub>                | 0.07   | 0.25   | 0.01   | 0.05   | 0.01   | 0.06   | 0.24   | 0.25   |
| TiO <sub>2</sub>                | 50.83  | 49.91  | 50.04  | 49.32  | 50.62  | 50.50  | 48.13  | 48.91  |
| Al <sub>2</sub> O <sub>3</sub>  | 0.00   | 0.05   | 0.02   | 0.36   | 0.02   | 0.07   | 0.22   | 0.18   |
| FeO <sub>tot</sub>              | 45.75  | 44.98  | 46.87  | 45.30  | 46.20  | 45.96  | 39.73  | 41.95  |
| MnO                             | 1.95   | 2.68   | 2.10   | 3.70   | 1.46   | 2.39   | 4.75   | 6.25   |
| MgO                             | 0.00   | 0.07   | 0.03   | 0.07   | 0.05   | 0.00   | 0.05   | 0.03   |
| CaO                             | 0.19   | 0.54   | 0.01   | 0.09   | 0.12   | 0.16   | 6.00   | 1.70   |
| Cr <sub>2</sub> O <sub>3</sub>  | 0.01   | 0.00   | 0.04   | 0.00   | 0.02   | 0.05   | 0.00   | 0.11   |
| Total                           | 98.80  | 98.48  | 99.12  | 98.89  | 98.50  | 99.19  | 99.12  | 99.38  |
| FeO                             | 43.57  | 41.64  | 42.81  | 40.42  | 43.80  | 42.85  | 30.98  | 35.71  |
| Fe <sub>2</sub> O <sub>3</sub>  | 2.42   | 3.70   | 4.51   | 5.42   | 2.66   | 3.45   | 9.72   | 6.93   |
| Total                           | 99.04  | 98.85  | 99.57  | 99.44  | 98.77  | 99.53  | 100.09 | 100.07 |
| % R <sub>2</sub> O <sub>3</sub> | 2.33   | 3.63   | 4.39   | 5.74   | 2.61   | 3.46   | 9.44   | 6.94   |
|                                 |        |        |        |        |        |        |        |        |
| T (°C)                          | 680    | 912    | 790    | 900    |        |        |        |        |
| Log( fO <sub>2</sub> )          | − 19.5 | − 13.1 | − 15.5 | − 13.0 |        |        |        |        |

Temperature and log(*f*O<sub>2</sub>) according to Buddington & Lindsley (1964) for homogeneous magnetite-ilmenite pairs. FeO, Fe<sub>2</sub>O<sub>3</sub>, ulvöspinel (Ulv.) and R<sub>2</sub>O<sub>3</sub> calculated according to Carmichael (1967).

(Table 4). The corresponding log(*f*O<sub>2</sub>) values range from −19.5 to −13.0 (Buddington & Lindsley, 1964), corresponding to conditions intermediate between fayalite-magnetite-quartz (FMQ) and wüstite-magnetite (WM) buffers (Fig. 5).

The scarce HTi tholeiitic dykes (D2) have ophitic textures. Labradoritic plagioclase is often altered to clay minerals and sericite and sometimes albitized (Ab up to 98%). Pyroxenes are augites (Wo 40–38) with high TiO<sub>2</sub> content (1.2–1.4 wt %; Table 2) and

Table 5: Major and trace element compositions of tholeiitic (D1, D1a, D2) and calc-alkaline (D3, D4) dykes from the Azul and Tandil regions

| Sample:                               | A1    | A2    | A4     | A5    | A8    | A9    | A10   | A16   | A25   |
|---------------------------------------|-------|-------|--------|-------|-------|-------|-------|-------|-------|
| Locality:                             | AZUL  | AZUL  | AZUL   | AZUL  | AZUL  | AZUL  | AZUL  | AZUL  | AZUL  |
| Direction:                            | N-S   | N-S   | N30°W  | N40°W | N40°W | N40°W | N40°W | N80°E | N35°W |
| Rock:                                 | ThB   | ThB   | ThB    | ThB   | ThB   | ThB   | ThB   | ThB   | ThB   |
| Group:                                | D1    | D1    | D1     | D1    | D1    | D1    | D1    | D1    | D1    |
| SiO <sub>2</sub> (wt %)               | 50.55 | 50.20 | 50.76  | 50.72 | 50.86 | 50.42 | 50.03 | 50.48 | 51.34 |
| TiO <sub>2</sub>                      | 1.26  | 1.29  | 1.16   | 0.99  | 1.19  | 1.36  | 1.21  | 0.92  | 0.87  |
| Al <sub>2</sub> O <sub>3</sub>        | 14.22 | 14.85 | 14.07  | 14.66 | 14.17 | 15.17 | 14.72 | 17.73 | 14.69 |
| FeO <sub>tot</sub>                    | 12.11 | 12.11 | 11.61  | 10.72 | 11.91 | 12.31 | 12.42 | 10.21 | 10.56 |
| MnO                                   | 0.20  | 0.19  | 0.19   | 0.18  | 0.19  | 0.19  | 0.20  | 0.16  | 0.19  |
| MgO                                   | 7.73  | 7.00  | 8.00   | 8.02  | 7.40  | 6.71  | 8.49  | 6.47  | 8.16  |
| CaO                                   | 10.77 | 11.55 | 11.46  | 12.24 | 11.39 | 9.46  | 9.09  | 10.56 | 11.21 |
| Na <sub>2</sub> O                     | 2.15  | 2.27  | 2.16   | 2.17  | 2.27  | 2.30  | 2.15  | 2.40  | 2.06  |
| K <sub>2</sub> O                      | 0.87  | 0.42  | 0.46   | 0.19  | 0.48  | 1.93  | 1.56  | 0.97  | 0.83  |
| P <sub>2</sub> O <sub>5</sub>         | 0.14  | 0.12  | 0.13   | 0.11  | 0.14  | 0.15  | 0.13  | 0.10  | 0.09  |
| mg-no.                                | 56.35 | 53.89 | 58.22  | 60.20 | 55.68 | 52.43 | 58.02 | 56.17 | 60.98 |
| LOI (wt %)                            | 2.02  | 1.40  | 1.84   | 1.47  | 2.36  | 2.63  | 3.55  | 1.36  | 2.41  |
| FeO (wt %)                            | 9.95  | 10.15 | 9.48   | 9.00  | 9.27  | 9.20  | 9.76  | 9.14  | 8.82  |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 2.40  | 2.18  | 2.37   | 1.91  | 2.93  | 3.46  | 2.96  | 1.19  | 1.93  |
| Cr (ppm)                              | 328   | 292   | 345    | 356   | 336   | 196   | 272   | 239   | 157   |
| Ni                                    | 137   | 144   | 140    | 147   | 127   | 115   | 131   | 134   | 110   |
| Rb                                    | 51    | 17    | 18     | 7     | 20    | 71    | 62    | 108   | 96    |
| Ba                                    | 87    | 99    | 61     | 31    | 47    | 165   | 274   | 136   | 65    |
| Sr                                    | 176   | 154   | 160.00 | 138   | 164   | 166   | 263   | 171   | 201   |
| Nb                                    | 4     | 6     | 6      | 5     | 6     | 6     | 9     | 2     | 3     |
| Zr                                    | 87    | 77    | 73     | 74    | 91    | 89    | 85    | 89    | 62    |
| Y                                     | 27    | 26    | 20     | 22    | 30    | 30    | 28    | 26    | 20    |
| La                                    | 6     | 3.62* | 3      | 3.60* | 4.41* | 6     | 5     | 7.33* | 2.77* |
| Ce                                    | 19    | 9.35  | 21     | 9.56  | 11.82 | 14    | 15    | 15.41 | 7.28  |
| Pr                                    |       | 1.30  |        | 1.33  | 1.78  |       |       | 1.95  | 1.05  |
| Nd                                    | 11    | 7.29  | 10     | 7.09  | 9.57  | 14    | 10    | 9.08  | 5.22  |
| Sm                                    |       | 2.57  |        | 2.31  | 2.98  |       |       | 2.46  | 1.78  |
| Eu                                    |       | 1.00  |        | 0.80  | 0.98  |       |       | 0.87  | 0.65  |
| Gd                                    |       | 3.35  |        | 2.99  | 3.37  |       |       | 2.96  | 2.01  |
| Tb                                    |       | 0.49  |        | 0.43  | 0.53  |       |       | 0.43  | 0.33  |
| Dy                                    |       | 3.16  |        | 2.95  | 3.69  |       |       | 2.69  | 2.26  |
| Ho                                    |       | 0.71  |        | 0.70  | 0.88  |       |       | 0.64  | 0.49  |
| Er                                    |       | 1.86  |        | 1.75  | 2.22  |       |       | 1.73  | 1.25  |
| Tm                                    |       | 0.27  |        | 0.26  | 0.32  |       |       | 0.27  | 0.18  |
| Yb                                    |       | 1.96  |        | 1.80  | 2.14  |       |       | 1.89  | 1.31  |
| Lu                                    |       | 0.28  |        | 0.27  | 0.30  |       |       | 0.27  | 0.20  |
| Q (CIPW)                              | 0.00  | 0.00  | 0.00   | 0.00  | 0.00  | 0.00  | 0.00  | 0.00  | 0.00  |
| OI/Hy (CIPW)                          | 0.20  | 0.15  | 0.10   | 0.10  | 0.07  | 0.76  | 0.71  | 0.34  | 0.09  |

| Sample:                               | A30    | A32    | A33    | A34    | A35    | MT65   | MT66   | MT55   | Mean     | SD            |
|---------------------------------------|--------|--------|--------|--------|--------|--------|--------|--------|----------|---------------|
| Locality:                             | TANDIL | TANDIL | TANDIL | TANDIL | TANDIL | TANDIL | TANDIL | TANDIL | $n = 34$ | (1 $\sigma$ ) |
| Direction:                            | N30°W  | E-W    | E-W    | N30°W  | N30°W  | N10°W  | N10°W  | N30°W  |          |               |
| Rock:                                 | BTrand | ThB    | ThB    | ThB    | ThB    | ThB    | ThB    | ThB    |          |               |
| Group:                                | D1     | D1     | D1     | D1     | D1     | D1     | D1     | D1     | D1       | D1            |
| SiO <sub>2</sub> (wt %)               | 52.46  | 48.72  | 50.86  | 50.36  | 50.67  | 50.62  | 50.69  | 50.00  | 50.61    | 0.59          |
| TiO <sub>2</sub>                      | 1.55   | 0.91   | 1.23   | 1.31   | 1.28   | 1.43   | 1.24   | 1.66   | 1.19     | 0.18          |
| Al <sub>2</sub> O <sub>3</sub>        | 15.55  | 15.65  | 14.28  | 14.48  | 15.41  | 12.99  | 14.34  | 16.17  | 14.89    | 1.09          |
| FeO <sub>tot</sub>                    | 13.56  | 12.46  | 12.78  | 13.29  | 12.29  | 13.24  | 11.89  | 13.87  | 11.93    | 1.01          |
| MnO                                   | 0.15   | 0.20   | 0.20   | 0.21   | 0.20   | 0.22   | 0.21   | 0.20   | 0.19     | 0.02          |
| MgO                                   | 3.82   | 8.44   | 7.57   | 6.93   | 6.32   | 7.46   | 7.59   | 4.63   | 7.28     | 0.89          |
| CaO                                   | 7.28   | 9.66   | 9.64   | 10.01  | 10.03  | 10.87  | 11.56  | 9.45   | 10.49    | 1.06          |
| Na <sub>2</sub> O                     | 4.75   | 2.30   | 2.16   | 2.15   | 2.35   | 2.33   | 2.00   | 2.60   | 2.31     | 0.48          |
| K <sub>2</sub> O                      | 0.64   | 1.22   | 1.14   | 1.14   | 1.33   | 0.64   | 0.33   | 1.19   | 0.96     | 0.42          |
| P <sub>2</sub> O <sub>5</sub>         | 0.24   | 0.44   | 0.14   | 0.12   | 0.12   | 0.14   | 0.10   | 0.18   | 0.14     | 0.06          |
| mg-no.                                | 36.29  | 57.80  | 54.50  | 51.32  | 50.98  | 53.26  | 56.35  | 40.30  | 55.11    | 4.43          |
| LOI (wt %)                            | 3.22   | 3.42   | 2.61   | 2.02   | 2.88   | 3.50   | 2.33   | 3.11   | 2.36     | 0.65          |
| FeO (wt %)                            | 9.44   | 10.44  | 9.58   | 10.68  | 9.85   | 7.49   | 4.73   | 10.15  | 9.40     | 1.24          |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 4.58   | 2.24   | 3.56   | 2.90   | 2.71   | 1.96   | 1.06   | 4.13   | 2.44     | 0.86          |
| Cr (ppm)                              | 14     | 144    | 144    | 139    | 94     | 165    | 171    | 71     | 220      | 85            |
| Ni                                    | 29     | 76     | 129    | 122    | 85     | 101    | 100    | 57     | 117      | 26            |
| Rb                                    | 29     | 64     | 29     | 48     | 48     | 30     | 15     | 35     | 62       | 47            |
| Ba                                    | 63     | 318    | 90     | 233    | 110    | 127    | 30     | 64     | 130      | 70            |
| Sr                                    | 326    | 302    | 173    | 156    | 156    | 205    | 189    | 238    | 198      | 56            |
| Nb                                    | 14     | 10     | 6      | 9      | 4      | 4      | 3      | 7      | 6        | 3             |
| Zr                                    | 142    | 96     | 78     | 96     | 76     | 97     | 59     | 96     | 85       | 15            |
| Y                                     | 44     | 27     | 26     | 26     | 24     | 22     | 15     | 18     | 26       | 5             |
| La                                    | 11     | 12     | 3.51*  | 5.26*  | 4      | 5      | 4      | 3      | 6        | 3             |
| Ce                                    | 30     | 33     | 9.63   | 13.19  | 18     | 13     | 8      | 13     | 18       | 5             |
| Pr                                    |        |        | 1.53   | 1.90   |        |        |        |        |          |               |
| Nd                                    | 19     | 18     | 8.02   | 9.17   | 9      | 10     | 7      | 15     | 11       | 3             |
| Sm                                    |        |        | 2.54   | 2.96   |        |        |        |        |          |               |
| Eu                                    |        |        | 0.94   | 0.97   |        |        |        |        |          |               |
| Gd                                    |        |        | 2.93   | 3.47   |        |        |        |        |          |               |
| Tb                                    |        |        | 0.47   | 0.55   |        |        |        |        |          |               |
| Dy                                    |        |        | 3.30   | 3.76   |        |        |        |        |          |               |
| Ho                                    |        |        | 0.75   | 0.84   |        |        |        |        |          |               |
| Er                                    |        |        | 2.00   | 2.23   |        |        |        |        |          |               |
| Tm                                    |        |        | 0.28   | 0.33   |        |        |        |        |          |               |
| Yb                                    |        |        | 1.92   | 2.26   |        |        |        |        |          |               |
| Lu                                    |        |        | 0.29   | 0.35   |        |        |        |        |          |               |
| Q (CIPW)                              | 0.00   | 0.00   | 0.00   | 0.00   | 0.00   | 0.00   | 0.43   | 0.00   |          |               |
| OI/Hy (CIPW)                          | 1.13   | 1.63   | 0.13   | 0.19   | 0.28   | 0.16   | 0.00   | 0.18   |          |               |

Table 5: continued

| Sample:                               | A19    | A20   | A21   | A22   | Mean    | SD            | A38    | A39    | A53    |
|---------------------------------------|--------|-------|-------|-------|---------|---------------|--------|--------|--------|
| Locality:                             | AZUL   | AZUL  | AZUL  | AZUL  | $n = 4$ | (1 $\sigma$ ) | TANDIL | TANDIL | TANDIL |
| Direction:                            | N30°W  | N30°W | N30°W | N30°W |         |               | E-W    | E-W    | E-W    |
| Rock:                                 | BAnd   | BAnd  | BAnd  | BAnd  |         |               | Trb    | ThB    | ThB    |
| Group:                                | D1a    | D1a   | D1a   | D1a   | D1a     | D1a           | D2     | D2     | D2     |
| SiO <sub>2</sub> (wt %)               | 53.69  | 53.62 | 53.72 | 53.68 | 53.68   | 0.04          | 47.99  | 48.11  | 49.58  |
| TiO <sub>2</sub>                      | 1.20   | 1.20  | 1.22  | 1.21  | 1.21    | 0.01          | 3.74   | 3.61   | 2.14   |
| Al <sub>2</sub> O <sub>3</sub>        | 15.34  | 15.58 | 15.27 | 15.75 | 15.49   | 0.22          | 14.16  | 13.99  | 13.06  |
| FeO <sub>tot</sub>                    | 11.52  | 11.45 | 11.52 | 11.49 | 11.50   | 0.03          | 14.76  | 15.83  | 15.25  |
| MnO                                   | 0.18   | 0.18  | 0.18  | 0.18  | 0.18    | 0.00          | 0.23   | 0.22   | 0.23   |
| MgO                                   | 4.71   | 4.98  | 4.71  | 4.47  | 4.72    | 0.21          | 5.42   | 3.87   | 6.72   |
| CaO                                   | 8.86   | 7.97  | 8.24  | 7.97  | 8.26    | 0.42          | 6.94   | 9.17   | 8.71   |
| Na <sub>2</sub> O                     | 2.89   | 2.95  | 2.89  | 2.95  | 2.92    | 0.03          | 2.71   | 2.05   | 2.25   |
| K <sub>2</sub> O                      | 1.26   | 1.72  | 1.91  | 1.93  | 1.71    | 0.31          | 2.58   | 1.63   | 0.90   |
| P <sub>2</sub> O <sub>5</sub>         | 0.35   | 0.35  | 0.34  | 0.37  | 0.35    | 0.01          | 1.47   | 1.52   | 1.16   |
| mg-no.                                | 45.26  | 46.79 | 45.26 | 44.03 | 45.34   | 1.13          | 42.61  | 33.08  | 47.12  |
| LOI (wt %)                            | 2.02   | 2.54  | 2.62  | 2.46  | 2.41    | 0.27          | 3.61   | 3.64   | 3.11   |
| FeO (wt %)                            | 10.85  | 10.20 | 10.08 | 10.79 | 10.48   | 0.40          | 11.22  | 12.80  | 12.24  |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 0.74   | 1.39  | 1.60  | 0.78  | 1.13    | 0.43          | 3.93   | 3.37   | 3.34   |
| Cr (ppm)                              | 109    | 104   | 110   | 102   | 106.25  | 3.86          | 54     | 58     | 149    |
| Ni                                    | 22     | 18    | 21    | 19    | 20.00   | 1.83          | 33     | 24     | 74     |
| Rb                                    | 36     | 55    | 63    | 58    | 53.00   | 11.80         | 81     | 47     | 34     |
| Ba                                    | 681    | 776   | 750   | 865   | 768.00  | 76.08         | 932    | 976    | 477    |
| Sr                                    | 298    | 335   | 353   | 340   | 331.50  | 23.59         | 309    | 488    | 367    |
| Nb                                    | 15     | 14    | 13    | 12    | 13.50   | 1.29          | 47     | 51     | 15     |
| Zr                                    | 192    | 190   | 195   | 188   | 191.25  | 2.99          | 390    | 415    | 205    |
| Y                                     | 39     | 36    | 39    | 39    | 38.25   | 1.50          | 62     | 66     | 37     |
| La                                    | 30.60* | 37    | 41    | 41    | 37.40   | 4.91          | 43.66* | 48     | 49     |
| Ce                                    | 62.00  | 72    | 75    | 72    | 70.25   | 5.68          | 87.12  | 96     | 91     |
| Pr                                    | 6.93   |       |       |       |         |               | 9.82   |        |        |
| Nd                                    | 26.95  | 38    | 39    | 36    | 34.99   | 5.50          | 44.49  | 58     | 55     |
| Sm                                    | 5.37   |       |       |       |         |               | 9.87   |        |        |
| Eu                                    | 1.61   |       |       |       |         |               | 3.55   |        |        |
| Gd                                    | 5.19   |       |       |       |         |               | 10.45  |        |        |
| Tb                                    | 0.73   |       |       |       |         |               | 1.41   |        |        |
| Dy                                    | 4.80   |       |       |       |         |               | 7.73   |        |        |
| Ho                                    | 1.06   |       |       |       |         |               | 1.66   |        |        |
| Er                                    | 2.80   |       |       |       |         |               | 3.95   |        |        |
| Tm                                    | 0.43   |       |       |       |         |               | 0.61   |        |        |
| Lu                                    | 0.46   |       |       |       |         |               | 0.68   |        |        |
| Q (CIPW)                              | 3.43   | 2.06  | 2.09  | 2.12  |         |               | 0.00   | 2.92   | 1.32   |
| OI/Hy (CIPW)                          | 0.00   | 0.00  | 0.00  | 0.00  |         |               | 0.44   | 0.00   | 0.00   |

| Sample:                               | MT67   | MT68   | MT69   | MT70   | Mean    | SD            | A40    | A41    | A49    | A52    |
|---------------------------------------|--------|--------|--------|--------|---------|---------------|--------|--------|--------|--------|
| Locality:                             | TANDIL | TANDIL | TANDIL | TANDIL | $n = 7$ | (1 $\sigma$ ) | TANDIL | TANDIL | TANDIL | TANDIL |
| Direction:                            | N15°W  | N15°W  | N70°W  | N70°E  |         |               | N80°E  | N80°E  | E-W    | N40°W  |
| Rock:                                 | ThB    | ThB    | ThB    | ThB    |         |               | BAnd   | BAnd   | BAnd   | BAnd   |
| Group:                                | D2     | D2     | D2     | D2     | D2      | D2            | D3     | D3     | D3     | D3     |
| SiO <sub>2</sub> (wt %)               | 48.28  | 49.04  | 50.24  | 48.66  | 48.84   | 0.83          | 56.55  | 58.20  | 56.04  | 56.03  |
| TiO <sub>2</sub>                      | 1.90   | 1.71   | 2.45   | 1.96   | 2.50    | 0.83          | 0.77   | 0.80   | 0.74   | 0.43   |
| Al <sub>2</sub> O <sub>3</sub>        | 14.42  | 14.02  | 14.02  | 14.16  | 13.98   | 0.43          | 13.59  | 14.26  | 14.12  | 15.26  |
| FeO <sub>tot</sub>                    | 14.62  | 15.31  | 14.31  | 14.95  | 15.00   | 0.50          | 9.39   | 8.45   | 9.53   | 8.67   |
| MnO                                   | 0.22   | 0.22   | 0.23   | 0.22   | 0.22    | 0.01          | 0.19   | 0.17   | 0.19   | 0.17   |
| MgO                                   | 6.88   | 6.55   | 5.57   | 6.43   | 5.92    | 1.06          | 6.97   | 5.40   | 6.88   | 7.22   |
| CaO                                   | 9.33   | 8.97   | 8.93   | 8.82   | 8.70    | 0.80          | 8.36   | 7.12   | 8.28   | 8.36   |
| Na <sub>2</sub> O                     | 2.37   | 2.34   | 2.57   | 2.69   | 2.43    | 0.24          | 2.86   | 3.10   | 2.23   | 2.43   |
| K <sub>2</sub> O                      | 0.95   | 0.95   | 0.93   | 1.04   | 1.28    | 0.63          | 0.93   | 2.21   | 1.61   | 1.31   |
| P <sub>2</sub> O <sub>5</sub>         | 0.96   | 0.80   | 0.67   | 0.98   | 1.08    | 0.32          | 0.39   | 0.29   | 0.38   | 0.12   |
| mg-no.                                | 48.76  | 46.38  | 44.04  | 46.52  | 44.07   | 5.25          | 60.02  | 56.37  | 59.35  | 62.74  |
| LOI (wt %)                            | 2.44   | 3.13   | 1.96   | 3.29   | 3.03    | 0.62          | 3.03   | 2.52   | 3.01   | 3.29   |
| FeO (wt %)                            | 9.48   | 10.88  | 9.56   | 12.06  | 11.18   | 1.30          | 7.10   | 6.44   | 7.11   | 7.31   |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 3.21   | 2.62   | 2.59   | 2.84   | 3.13    | 0.48          | 2.54   | 2.23   | 2.69   | 1.51   |
| Cr (ppm)                              | 92     | 117    | 132    | 78     | 97      | 37            | 241    | 114    | 194    | 60     |
| Ni                                    | 104    | 84     | 77     | 83     | 68      | 29            | 19     | 14     | 20     | 21     |
| Rb                                    | 46     | 46     | 51     | 41     | 49      | 15            | 49     | 93     | 59     | 48     |
| Ba                                    | 493    | 505    | 402    | 529    | 616     | 234           | 353    | 754    | 493    | 354    |
| Sr                                    | 448    | 392    | 388    | 422    | 402     | 58            | 564    | 666    | 595    | 456    |
| Nb                                    | 17     | 16     | 15     | 14     | 25      | 16            | 9      | 10     | 9      | 5      |
| Zr                                    | 167    | 187    | 230    | 183    | 254     | 104           | 119    | 126    | 117    | 74     |
| Y                                     | 24     | 24     | 24     | 25     | 37      | 19            | 23     | 21     | 21     | 14     |
| La                                    | 30     | 28     | 32     | 30     | 39      | 11            | 30     | 26.32* | 23     | 12     |
| Ce                                    | 72     | 57     | 78     | 63     | 80      | 17            | 63     | 56.85  | 55     | 33     |
| Pr                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Nd                                    | 37     | 37     | 46     | 40     | 48      | 10            | 31     | 28.01  | 30     | 14     |
| Sm                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Eu                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Gd                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Tb                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Dy                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Ho                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Er                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Tm                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Yb                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Lu                                    |        |        |        |        |         |               |        | 6.53   |        |        |
| Q (CIPW)                              | 0.00   | 0.00   | 1.44   | 0.00   |         |               | 6.88   | 7.10   | 7.02   | 5.74   |
| OI/Hy (CIPW)                          | 0.32   | 0.14   | 0.00   | 0.39   |         |               | 0.00   | 0.00   | 0.00   | 0.00   |

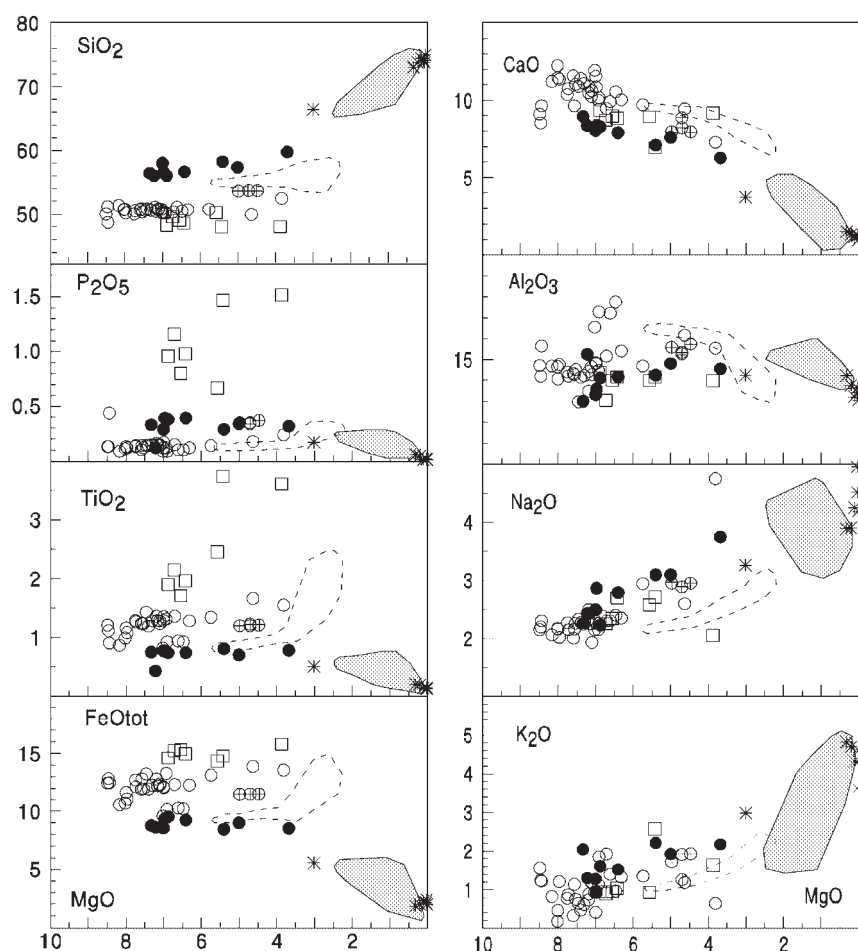
Table 5: continued

| Sample:                               | A54    | T12    | Mean    | SD            | A45    | A46    | A47    | A48    | A51    |
|---------------------------------------|--------|--------|---------|---------------|--------|--------|--------|--------|--------|
| Locality:                             | TANDIL | TANDIL | $n = 9$ | (1 $\sigma$ ) | TANDIL | TANDIL | TANDIL | TANDIL | TANDIL |
| Direction:                            | E-W    | E-W    |         |               | N80°E  | E-W    | E-W    | E-W    | E-W    |
| Rock:                                 | And    | And    |         |               | Rhy    | Rhy    | Rhy    | Rhy    | Rhy    |
| Group:                                | D3     | D3     | D3      | D3            | D4     | D4     | D4     | D4     | D4     |
| SiO <sub>2</sub> (wt %)               | 59.77  | 58.03  | 57.22   | 1.25          | 74.44  | 73.84  | 75.01  | 73.82  | 76.21  |
| TiO <sub>2</sub>                      | 0.78   | 0.77   | 0.72    | 0.11          | 0.14   | 0.19   | 0.12   | 0.15   | 0.11   |
| Al <sub>2</sub> O <sub>3</sub>        | 14.57  | 13.33  | 14.12   | 0.72          | 13.23  | 13.77  | 13.06  | 13.42  | 12.72  |
| FeO <sub>tot</sub>                    | 8.52   | 8.58   | 8.91    | 0.40          | 2.28   | 1.97   | 1.96   | 2.38   | 1.88   |
| MnO                                   | 0.16   | 0.18   | 0.18    | 0.01          | 0.05   | 0.04   | 0.06   | 0.06   | 0.03   |
| MgO                                   | 3.68   | 7      | 6.21    | 1.25          | 0.11   | 0.21   | 0.01   | 0.03   | 0.17   |
| CaO                                   | 6.28   | 8.05   | 7.88    | 0.79          | 1.17   | 1.33   | 0.98   | 1.31   | 1.30   |
| Na <sub>2</sub> O                     | 3.75   | 2.49   | 2.78    | 0.49          | 4.25   | 3.91   | 4.19   | 4.52   | 3.84   |
| K <sub>2</sub> O                      | 2.17   | 1.28   | 1.67    | 0.45          | 4.32   | 4.70   | 4.61   | 4.30   | 3.72   |
| P <sub>2</sub> O <sub>5</sub>         | 0.32   | 0.29   | 0.32    | 0.08          | 0.01   | 0.04   | 0.01   | 0.01   | 0.02   |
| mg-no.                                | 46.62  | 61.77  | 57.72   | 5.30          | 8.89   | 17.73  | 1.02   | 2.48   | 15.46  |
| LOI (wt %)                            | 1.65   | 2.66   | 2.73    | 0.56          | 0.76   | 0.70   | 0.65   | 0.57   | 0.62   |
| FeO (wt %)                            | 6.80   | 6.31   | 6.88    | 0.39          | 2.18   | 1.75   | 1.57   | 2.19   | 1.62   |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 1.91   | 1.64   | 1.85    | 0.56          | 0.11   | 0.24   | 0.43   | 0.21   | 0.29   |
| Cr (ppm)                              | 45     | 361    | 186     | 119           | 1      | 3      | 3      | 1      | 2      |
| Ni                                    | 12     | 35     | 20      | 8             | 3      | 2      | 1      | 1      | 5      |
| Rb                                    | 154    | 63     | 82      | 35            | 145    | 178    | 148    | 139    | 141    |
| Ba                                    | 517    | 474    | 520     | 130           | 1219   | 1185   | 1355   | 1326   | 802    |
| Sr                                    | 375    | 570    | 552     | 88            | 181    | 261    | 135    | 182    | 178    |
| Nb                                    | 9      | 5      | 8       | 2             | 12     | 13     | 12     | 12     | 11     |
| Zr                                    | 135    | 132    | 113     | 20            | 273    | 213    | 227    | 233    | 169    |
| Y                                     | 29     | 22     | 22      | 4             | 33     | 22     | 30     | 29     | 26     |
| La                                    | 24.02* | 30     | 26      | 6             | 57     | 56.62* | 53.80* | 55     | 56     |
| Ce                                    | 54.46  | 63     | 58      | 12            | 108    | 105.10 | 106.40 | 101    | 107    |
| Pr                                    | 6.43   |        |         |               |        | 10.35  | 11.36  |        |        |
| Nd                                    | 25.83  | 32     | 30      | 7             | 49     | 36.86  | 42.04  | 48     | 49     |
| Sm                                    | 4.85   |        |         |               |        | 6.05   | 6.94   |        |        |
| Eu                                    | 1.02   |        |         |               |        | 1.11   | 1.21   |        |        |
| Gd                                    | 3.86   |        |         |               |        | 4.53   | 5.30   |        |        |
| Tb                                    | 0.53   |        |         |               |        | 0.55   | 0.74   |        |        |
| Dy                                    | 3.18   |        |         |               |        | 2.74   | 4.34   |        |        |
| Ho                                    | 0.66   |        |         |               |        | 0.50   | 0.78   |        |        |
| Er                                    | 1.50   |        |         |               |        | 1.36   | 1.96   |        |        |
| Tm                                    | 0.24   |        |         |               |        | 0.22   | 0.29   |        |        |
| Yb                                    | 1.61   |        |         |               |        | 1.47   | 2.11   |        |        |
| Lu                                    | 0.26   |        |         |               |        | 0.22   | 0.33   |        |        |
| Q (CIPW)                              | 8.91   | 9.90   |         |               | 29.36  | 28.85  | 29.94  | 27.27  | 35.41  |
| OI/Hy (CIPW)                          | 0.00   | 0.00   |         |               | 0.00   | 0.00   | 0.00   | 0.00   | 0.00   |

| Sample:                               | A36           | BR1           | BR2    | mean    | SD            | A50     | A44     | LP3     | SL1    |
|---------------------------------------|---------------|---------------|--------|---------|---------------|---------|---------|---------|--------|
| Locality:                             | TANDIL        | TANDIL        | TANDIL | $n = 8$ | (1 $\sigma$ ) | TANDIL  | TANDIL  | TANDIL  | TANDIL |
| Direction:                            | E-W           | E-W           | E-W    |         |               |         |         |         |        |
| Rock:                                 | Dacite        | Rhy           | Rhy    |         |               | Granite | Diorite | Diorite | Gabbro |
| Group:                                | D4            | D4            | D4     | D4      | D4            |         |         |         |        |
| SiO <sub>2</sub> (wt %)               | 66.41         | 72.98         | 74.06  | 73.35   | 2.96          | 69.71   | 65.82   | 54.81   | 49.45  |
| TiO <sub>2</sub>                      | 0.50          | 0.20          | 0.14   | 0.19    | 0.13          | 0.34    | 0.55    | 0.94    | 0.44   |
| Al <sub>2</sub> O <sub>3</sub>        | 14.26         | 14.23         | 13.66  | 13.54   | 0.54          | 16.08   | 15.00   | 19.69   | 14.59  |
| FeO <sub>tot</sub>                    | 5.56          | 1.87          | 2.16   | 2.51    | 1.25          | 2.64    | 5.29    | 7.44    | 8.42   |
| MnO                                   | 0.11          | 0.04          | 0.05   | 0.06    | 0.02          | 0.03    | 0.11    | 0.13    | 0.16   |
| MgO                                   | 3.01          | 0.32          | 0.04   | 0.49    | 1.02          | 1.09    | 2.48    | 3.05    | 11.22  |
| CaO                                   | 3.73          | 1.50          | 1.22   | 1.57    | 0.89          | 3.79    | 4.66    | 8.99    | 13.71  |
| Na <sub>2</sub> O                     | 3.26          | 3.89          | 4.95   | 4.10    | 0.50          | 4.73    | 4.28    | 3.05    | 1.21   |
| K <sub>2</sub> O                      | 2.99          | 4.82          | 3.63   | 4.14    | 0.63          | 1.48    | 1.57    | 1.34    | 0.50   |
| P <sub>2</sub> O <sub>5</sub>         | 0.17          | 0.06          | 0.02   | 0.04    | 0.05          | 0.11    | 0.24    | 0.45    | 0.19   |
| <i>mg</i> -no.                        | 52.26         | 25.71         | 3.61   | 15.90   | 17.00         | 45.50   | 48.67   | 45.32   | 72.93  |
| LOI (wt %)                            | 2.52          | 3.24          | 3.01   | 1.51    | 1.19          | 0.86    | 1.66    | n.m.    | n.m.   |
| FeO (wt %)                            | 4.20          | 12.51         | 12.37  | 4.80    | 4.79          | 2.64    | 4.68    | n.m.    | n.m.   |
| Fe <sub>2</sub> O <sub>3</sub> (wt %) | 1.51          | 3.11          | 2.16   | 1.01    | 1.13          | 0.00    | 0.68    | n.m.    | n.m.   |
| Cr (ppm)                              | 66            | 4             | 2      | 10      | 23            | 1       | 19      | 54      | 327    |
| Ni                                    | 10            | 0             | 0      | 3       | 3             | 2       | 10      | 7       | 40     |
| Rb                                    | 112           | 201           | 123    | 148     | 29            | 37      | 80      | 39      | 13     |
| Ba                                    | 912           | 1220          | 1184   | 1150    | 194           | 913     | 567     | 432     | 301    |
| Sr                                    | 519           | 326           | 242    | 253     | 123           | 1021    | 751     | 862     | 548    |
| Nb                                    | 10            | 9             | 10     | 11      | 1             | 3       | 11      | 11      | 7      |
| Zr                                    | 174           | 201           | 268    | 220     | 39            | 80      | 108     | 62      | 51     |
| Y                                     | 26            | 15            | 35     | 27      | 6             | 3       | 16      | 12      | 8      |
| La                                    | <i>38.70*</i> | <i>59.36*</i> | 49     | 54      | 6             | 14      | 16      | 15      | 19     |
| Ce                                    | <i>79.44</i>  | <i>117.20</i> | 98     | 101     | 12            | 28      | 35      | 46      | 47     |
| Pr                                    | <i>8.41</i>   | <i>11.94</i>  |        |         |               |         |         |         |        |
| Nd                                    | <i>30.69</i>  | <i>42.33</i>  | 41     | 45      | 6             | 9       | 15      | 19      | 28     |
| Sm                                    | <i>5.42</i>   | <i>5.96</i>   |        |         |               |         |         |         |        |
| Eu                                    | <i>1.15</i>   | <i>1.13</i>   |        |         |               |         |         |         |        |
| Gd                                    | <i>4.27</i>   | <i>3.38</i>   |        |         |               |         |         |         |        |
| Tb                                    | <i>0.58</i>   | <i>0.45</i>   |        |         |               |         |         |         |        |
| Dy                                    | <i>3.40</i>   | <i>2.23</i>   |        |         |               |         |         |         |        |
| Ho                                    | <i>0.71</i>   | <i>0.43</i>   |        |         |               |         |         |         |        |
| Er                                    | <i>1.70</i>   | <i>1.11</i>   |        |         |               |         |         |         |        |
| Tm                                    | <i>0.25</i>   | <i>0.15</i>   |        |         |               |         |         |         |        |
| Yb                                    | <i>1.80</i>   | <i>1.01</i>   |        |         |               |         |         |         |        |
| Lu                                    | <i>0.28</i>   | <i>0.15</i>   |        |         |               |         |         |         |        |
| Q (CIPW)                              | 20.88         | 27.21         | 27.67  |         |               |         |         |         |        |
| OI/Hy (CIPW)                          | 0.00          | 0.00          | 0.00   |         |               |         |         |         |        |

\*REE values in italics were measured by ICP-AES.

ThB, tholeiitic basalt; BTrand, basaltic trachyandesite; BAnd, basaltic andesite; Trb, trachybasalt; And, andesite; Rhy, rhyolite. FeO<sub>tot</sub>, total Fe as FeO; *mg*-number,  $100 \times \text{at. Mg}/(\text{Mg} + \text{Fe}^{2+})$  for Fe<sub>2</sub>O<sub>3</sub>/FeO = 0.15. Q, OI and Hy are CIPW normative quartz, olivine and hypersthene. n.m., not measured.



**Fig. 6.** MgO (wt %) vs major elements (wt %) for Azul and Tandil dyke swarms. Symbols as in Fig. 4; also, crossed circles, D1a tholeiitic dykes; \*, rhyolite (D4) dykes. Fields outlined with continuous line, dyke country rocks; fields outlined with dashed line, 1.7 Ga Florida (Uruguay) tholeiitic dyke swarms (Bossi *et al.*, 1993).

pigeonite. Pyroxenes may be replaced by hornblende, actinolite–tremolite and chlorite. Ilmenite (Table 4), magnetite, rutile and pyrite (Echeveste & Fernandez, 1994), and apatite are abundant in the groundmass.

The basic–intermediate calc-alkaline dykes (D3) have porphyritic to intergranular textures. The porphyritic dykes have phenocrysts of plagioclase ( $An_{55-34}$ , Table 1), often altered to sericite and clay minerals, and salitic–augitic pyroxenes (Table 2), almost completely replaced by amphiboles and opaque minerals. The holocrystalline matrix is composed of microphenocrysts of altered plagioclase, epidote, biotite–chlorite, opaques (ilmenite, Table 4; pyrrhotite, Echeveste & Fernandez, 1994), altered alkali-feldspar and quartz. The dykes with intergranular texture have plagioclase and augite completely replaced by secondary minerals. Quartz–feldspar intergrowths are present in the groundmass.

The dacitic and rhyolitic dykes (D4) have a porphyritic texture, phenocrysts of altered plagioclase, alkali-feldspar

and augite being replaced by amphibole–chlorite aggregates. The same minerals are found in the holomicrocrystalline matrix. Secondary zeolites or quartz and K-feldspars may fill vesicles or fractures.

## GEOCHEMISTRY

Major and trace elements were determined using a PW 1404 XRF spectrometer, following the procedures of Philips (1994) for the correction of matrix effects. Results are accurate to within 2–3% for major elements and better than 7–10% for trace elements. Ferrous iron was determined by redox titration; loss on ignition (LOI) was determined at 1100°C (12 h) and corrected for  $Fe^{2+}$  oxidation. Rare earth elements (REE) were determined by inductively coupled plasma-atomic emission spectrometry (ICP-AES) at the Centre de Recherches Petrographiques et Géochimiques, CNRS,

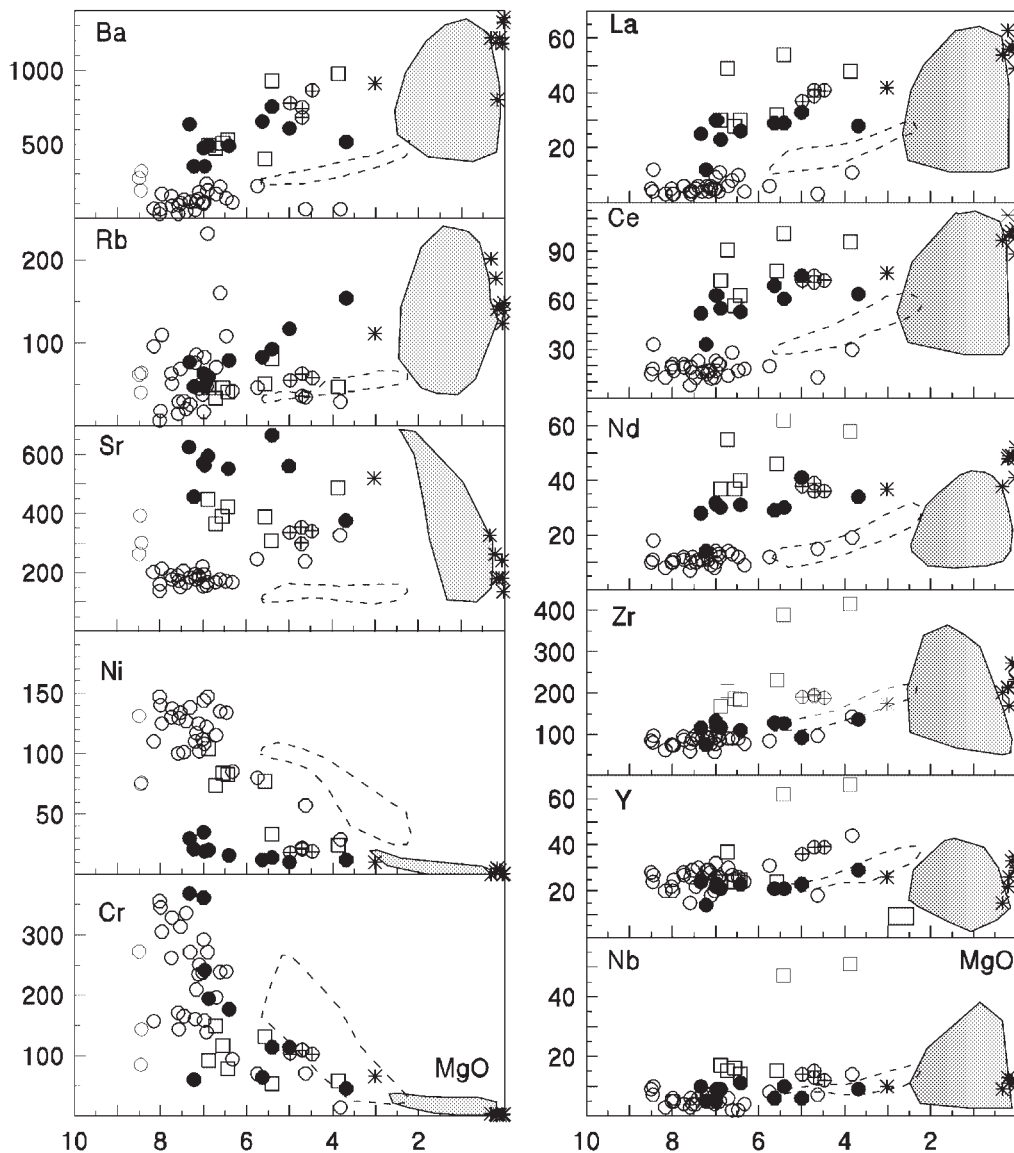


Fig. 7. MgO (wt %) vs trace elements (ppm) for Azul and Tandil dyke swarms. Symbols as in Figs 4 and 6.

Vandoeuvre (France) (Govindaraju & Mevelle, 1987).

Low-TiO<sub>2</sub> (LTi) tholeiites (Table 5) have SiO<sub>2</sub> and MgO contents ranging from 48.7 to 53.7 wt % (mean  $51.0 \pm 1.2$  wt %) and 8.8 to 3.8 wt %, respectively, and *mg*-number values from 61 to 36. As shown in the variation diagrams (Figs 6 and 7), with respect to MgO, the trend of SiO<sub>2</sub> is rather flat, like that of P<sub>2</sub>O<sub>5</sub> ( $0.12 \pm 0.03$  wt %), TiO<sub>2</sub> ( $1.34 \pm 0.13$  wt %) and FeO<sub>tot</sub> ( $12.6 \pm 1.0$  wt %). The D1 dykes plotting in the basalt field have low contents of incompatible elements (IE), whereas those with basaltic andesite composition (D1a), for similar MgO, have higher IE (e.g. La, Ce, Nd, Zr; Fig. 7). D1a dykes also have La/Y ( $\sim 1.0$ ) and Zr/Y ( $\sim 5.0$ ) that are clearly higher

(Fig. 8) than those of the D1 dykes ( $\sim 0.3$  and 3–4, respectively), excluding the possibility that they derived from D1 basalt dykes through simple fractional crystallization. It should be noted that the D1 tholeiites have REE patterns (Fig. 9) similar to the E-MORB of Sun & McDonough (1989), with mean chondrite-normalized (La/Yb)<sub>CN</sub> of  $1.55 \pm 0.48$ , whereas the D1a-type tholeiite is more fractionated, i.e. (La/Yb)<sub>CN</sub> = 6.83. Considerable differences between D1 and D1a tholeiites are also apparent, excluding mobile elements such as Rb, Ba and K, in the multi-elemental diagram relative to the Primitive Mantle (PM) of Sun & McDonough (1989; Fig. 9), in which the (La/Nb)<sub>PM</sub> and (Zr/Ti)<sub>PM</sub> ratios of D1 dykes are lower than those

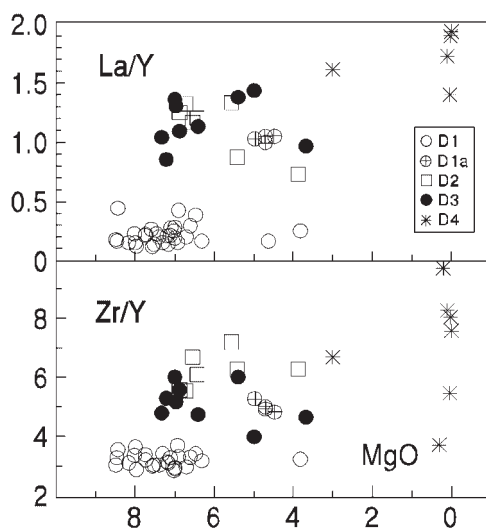


Fig. 8. MgO (wt %) vs Zr/Y and La/Y ratios for Azul and Tandil dyke swarms.

of IE-enriched D1a dykes ( $1.06 \pm 0.78$  vs  $2.92 \pm 0.63$ , and  $1.31 \pm 0.36$  vs  $3.07 \pm 0.04$ , respectively).

High-TiO<sub>2</sub> (HTi) tholeiites have MgO ranging from 6.9 to 3.9 wt % (*mg*-number 49–33) and relatively low SiO<sub>2</sub> contents ( $48.8 \pm 0.8$  wt %). They are characterized by high concentrations of TiO<sub>2</sub> ( $1.71$ – $3.74$ , mean  $2.50 \pm 0.83$  wt %), P<sub>2</sub>O<sub>5</sub> ( $0.67$ – $1.52$  wt %), REE, Zr ( $254 \pm 104$  ppm) and Nb ( $25 \pm 16$  ppm). The REE pattern (A38) of a D2 dyke parallels that of the D1a basaltic andesite (LTi), and is similarly characterized by high (La/Yb)<sub>CN</sub> ( $6.72$ ). In the multi-elemental diagram, the HTi dykes have enriched IE patterns with respect to LTi dykes, being characterized by positive P and negative Sr and Ti ( $Zr/Ti_{PM} = 1.93 \pm 0.19$ ) spikes. It should be noted that (La/Nb)<sub>PM</sub> ranges from 1.0 to 3.4 (mean =  $1.92 \pm 0.83$ ).

The calc-alkaline basaltic andesites and andesites have MgO and SiO<sub>2</sub> ranging from 7.0 to 3.7 wt % (*mg*-number 60–47) and 56.0 to 59.8 wt %, respectively. These dykes (D3), relative to the D1 tholeiites, have lower mean TiO<sub>2</sub> ( $0.72$  vs  $1.20$  wt %) and FeO<sub>tot</sub> ( $8.9$

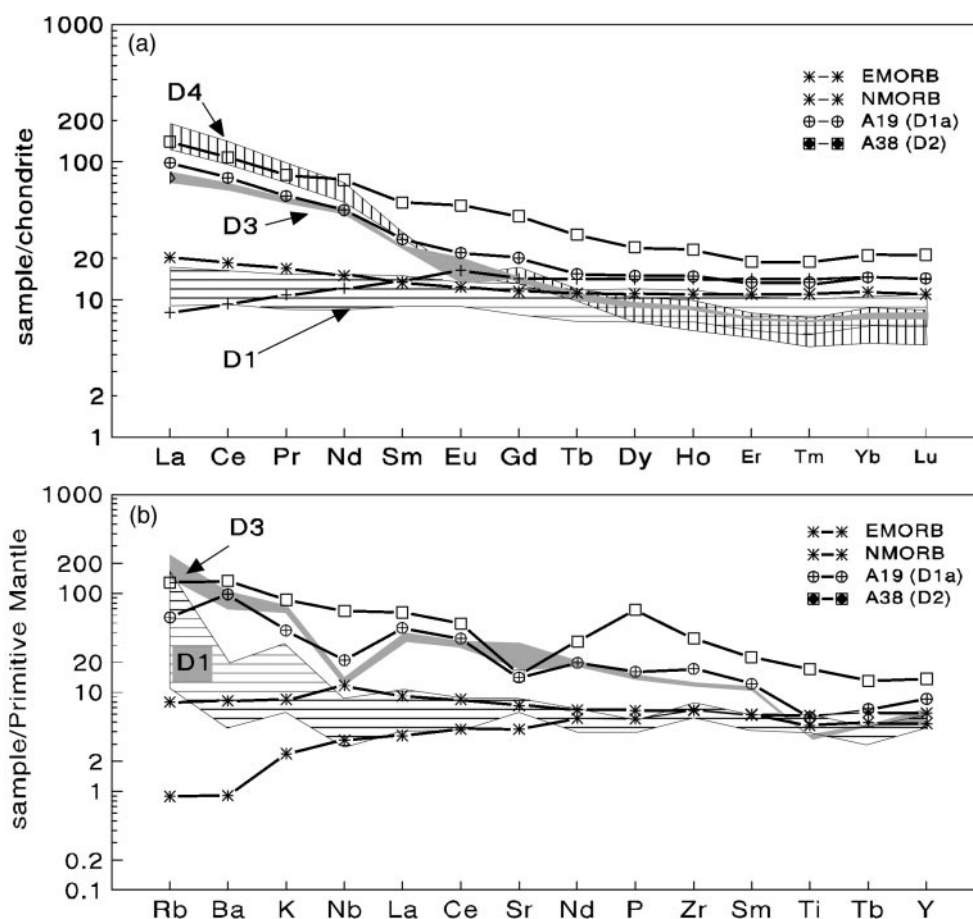


Fig. 9. (a) Chondrite-normalized (Boynton, 1984) REE abundance patterns of tholeiitic (D1, D1a, D2) and calc-alkaline (D3, D4) dyke swarms from Azul and Tandil regions. (b) Incompatible element patterns, normalized to Primitive Mantle of Sun & McDonough (1989), for tholeiitic and calc-alkaline dyke swarms from the Azul and Tandil regions.

Table 6:  $^{87}\text{Sr}/^{86}\text{Sr}$  isotopic compositions ( $2\sigma\text{SD}$  in parentheses) of tholeiitic (D1, D2), calc-alkaline (D3, D4) dykes and country rocks from the Azul and Tandil regions

| Sample                         | Rb (ppm) | Sr (ppm) | $^{87}\text{Sr}/^{86}\text{Sr}$ | $\text{Sr}_i$ | $\epsilon\text{Sr}$ |
|--------------------------------|----------|----------|---------------------------------|---------------|---------------------|
| <i>D1 Group (1588 Ma)</i>      |          |          |                                 |               |                     |
| A1 (ThB)                       | 40.89    | 166.93   | 0.72416(20)                     | 0.70797       | 76                  |
| A2 (ThB)                       | 16.50    | 149.62   | 0.71025(16)                     | 0.70297       | 5                   |
| A4 (ThB)                       | 16.85    | 166.20   | 0.71370(14)                     | 0.70701       | 63                  |
| A5 (ThB)                       | 5.74     | 134.16   | 0.70717(18)                     | 0.70435       | 24                  |
| A8 (ThB)                       | 16.70    | 157.18   | 0.71314(24)                     | 0.70613       | 50                  |
| A33 (ThB)                      | 29.79    | 195.91   | 0.71508(18)                     | 0.70504       | 34                  |
| A35 (ThB)                      | 42.09    | 171.38   | 0.72639(20)                     | 0.71016       | 107                 |
| MT65 (ThB)                     | 29.71    | 192.44   | 0.71540(14)                     | 0.70521       | 35                  |
| <i>D2 Group (1588 Ma)</i>      |          |          |                                 |               |                     |
| A38 (Trb)                      | 62.48    | 237.91   | 0.72187(18)                     | 0.70452       | 27                  |
| A39 (ThB)                      | 37.53    | 453.13   | 0.71218(18)                     | 0.70671       | 58                  |
| A53 (ThB)                      | 29.84    | 330.84   | 0.70895(18)                     | 0.70300       | 5                   |
| MT67 (ThB)                     | 39.26    | 404.26   | 0.71705(18)                     | 0.71064       | 114                 |
| MT68 (ThB)                     | 38.68    | 370.56   | 0.71785(16)                     | 0.71096       | 118                 |
| MT69 (ThB)                     | 45.87    | 360.57   | 0.71228(18)                     | 0.70388       | 18                  |
| MT70 (ThB)                     | 33.21    | 393.17   | 0.71425(34)                     | 0.70867       | 86                  |
| <i>D3 group (2020 Ma)</i>      |          |          |                                 |               |                     |
| A40 (BAnd)                     | 42.00    | 577.84   | 0.70987(14)                     | 0.70375       | 23                  |
| A41 (BAnd)                     | 58.79    | 663.02   | 0.71250(11)                     | 0.70503       | 41                  |
| A49 (BAnd)                     | 51.96    | 577.19   | 0.71125(11)                     | 0.70367       | 22                  |
| A52 (BAnd)                     | 43.76    | 507.00   | 0.71046(16)                     | 0.70319       | 15                  |
| A54 (And)                      | 147.76   | 378.90   | 0.73667(16)                     | 0.70373       | 23                  |
| <i>D4 group (2020 Ma)</i>      |          |          |                                 |               |                     |
| A47 (Rhy)                      | 142.52   | 142.55   | 0.78779(16)                     | 0.70293       | 12                  |
| BR2 (Rhy)                      | 117.50   | 251.61   | 0.74389(18)                     | 0.70440       | 32                  |
| <i>Country rocks (2150 Ma)</i> |          |          |                                 |               |                     |
| A50 (granite)                  | 31.83    | 1057.71  | 0.70524(24)                     | 0.70254       | 8                   |
| SL2 (gabbro)                   | 13.37    | 551.99   | 0.70457(12)                     | 0.70240       | 6                   |
| LP3 (diorite)                  | 37.18    | 899.70   | 0.70693(18)                     | 0.70322       | 18                  |
| A44 (diorite)                  | 64.54    | 671.64   | 0.71270(16)                     | 0.70476       | 30                  |

Rb and Sr measured by isotope dilution. Abbreviations as in Table 5.

vs 11.9 wt %) and higher  $\text{P}_2\text{O}_5$  (0.32 vs 0.14 wt %), Sr (552 vs 199 ppm), REE (e.g. La 26 vs 6 ppm; Nd 30 vs 11 ppm) and Zr (113 vs 86). The REE patterns of D3 dykes (Fig. 9) are different from those of tholeiitic D1 and D2 dykes, and are characterized by the highest  $(\text{La}/\text{Yb})_{\text{CN}}$  values, i.e. 10–12. In the multi-elemental diagram (Fig. 9), D3 dykes are distinct from the others by having pronounced negative spikes of Nb ( $\text{La}/\text{Nb}_{\text{PM}} = 3.39 \pm 1.43$ ) and Ti ( $\text{Zr}/\text{Ti}_{\text{PM}} = 3.18 \pm 0.15$ ).

The silicic dykes ( $\text{SiO}_2$  65–75 wt %) have the highest IE contents and are characterized by very high  $(\text{La}/\text{Yb})_{\text{CN}}$  (15–40) and a significant Eu negative anomaly ( $\text{Eu}/\text{Eu}^* = 0.61\text{--}0.76$ ).

## ISOTOPE GEOCHEMISTRY

Sr and Nd isotope analyses of selected samples were performed at the ‘Centro de Pesquisas Geocronológicas’

Table 7:  $^{143}\text{Nd}/^{144}\text{Nd}$  isotopic compositions ( $2\sigma\text{SD}$  in parentheses) of tholeiitic (D1, D2) and calc-alkaline (D3, D4) dykes from the Azul and Tandil regions

| Sample                    | Sm (ppm) | Nd (ppm) | $^{143}\text{Nd}/^{144}\text{Nd}$ | $\text{Nd}_i$ | $\epsilon\text{Nd}$ |
|---------------------------|----------|----------|-----------------------------------|---------------|---------------------|
| <i>D1 Group (1588 Ma)</i> |          |          |                                   |               |                     |
| A4 (ThB)                  | 1.900    | 7.635    | 0.512778(39)                      | 0.511206      | 12.18               |
| A5 (ThB)                  | 2.280    | 6.926    | 0.512893(41)                      | 0.510814      | 4.49                |
| A16 (ThB)                 | 2.571    | 9.633    | 0.512250(20)                      | 0.510564      | -0.39               |
| <i>D2 Group (1588 Ma)</i> |          |          |                                   |               |                     |
| A38 (Trb)                 | 10.433   | 48.003   | 0.511895(20)                      | 0.510518      | -1.30               |
| MT70 (ThB)                | 7.285    | 38.939   | 0.511394(37)                      | 0.510242      | -6.71               |
| <i>D3 Group (2020 Ma)</i> |          |          |                                   |               |                     |
| A40 (BAnd)                | -4.667   | 25.056   | 0.511375(38)                      | 0.509878      | -2.84               |
| A41 (BAnd)                | 4.925    | 26.675   | 0.511323(40)                      | 0.509839      | -3.59               |
| A54 (And)                 | 5.179    | 26.699   | 0.511392(10)                      | 0.509832      | -3.73               |
| <i>D4 Group (2020 Ma)</i> |          |          |                                   |               |                     |
| A48 (Rhy)                 | 4.260    | 22.531   | 0.511372(40)                      | 0.509852      | -3.34               |
| A51 (Rhy)                 | 7.063    | 40.709   | 0.511190(10)                      | 0.509795      | -4.45               |

Sm and Nd measured by isotope dilution. Abbreviations as in Table 5.

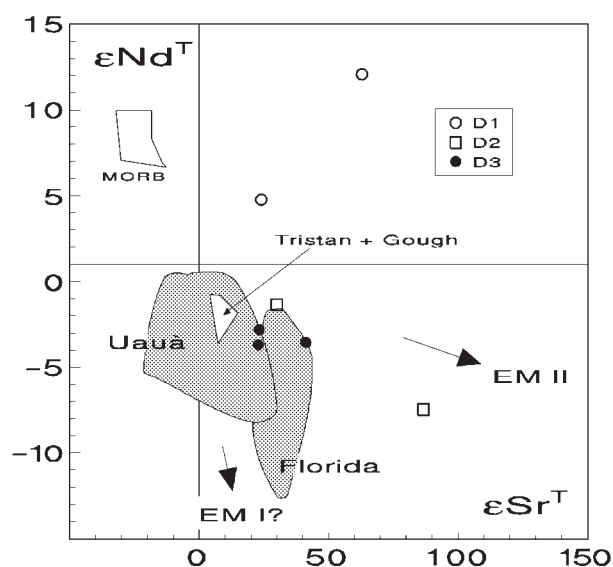


Fig. 10.  $\epsilon\text{Sr}^T$  vs  $\epsilon\text{Nd}^T$  diagram for tholeiitic and calc-alkaline dyke swarms of the Azul and Tandil regions. Fields of MORB, Uauá (2.2–2.0 Ga, São Francisco craton, Brazil; Bellieni *et al.*, 1995) and Florida (1.7 Ga, Rio de La Plata craton, Uruguay; Mazzucchelli *et al.*, 1995; Teixeira *et al.*, 1999) are shown for comparison. EM I and EM II after Hart & Zindler (1989).

of São Paulo, following the procedure of Ludwig (1985). Rb, Sr, Sm and Nd concentrations were measured by isotopic dilution. The average  $^{87}\text{Sr}/^{86}\text{Sr}$

for NBS-987 and  $^{143}\text{Nd}/^{144}\text{Nd}$  for La Jolla standards were 0.710254(22) and 0.511857(46), respectively, with  $2\sigma$  standard deviations reported in parentheses.

The initial  $^{87}\text{Sr}/^{86}\text{Sr}$  ( $\text{Sr}_i$ ) and  $^{143}\text{Nd}/^{144}\text{Nd}$  ( $\text{Nd}_i$ ) were calculated to 1588 Ma (baddeleyite U–Pb age: Teixeira *et al.*, in preparation) for D1 and D2 dykes, 2020 Ma (biotite  $^{40}\text{Ar}/^{39}\text{Ar}$  age: Teixeira *et al.*, in preparation) for D3 and D4 dykes, and 2150 Ma for the country rocks.

The studied dykes have  $\text{Sr}_i$  ratios (Table 6) higher than those of Bulk Earth (BE): tholeiitic dykes: 0.7030–0.7102 for D1 ( $\epsilon\text{Sr}$  5–107), 0.7030–0.7110 for D2 ( $\epsilon\text{Sr}$  6–118); calc-alkaline dykes: 0.7032–0.7050 for D3 ( $\epsilon\text{Sr}$  15–41) and 0.7029–0.7044 for D4 ( $\epsilon\text{Sr}$  12–32). The gabbroic to granitic country rocks have  $\text{Sr}_i$  ranging from 0.7025 to 0.7048 ( $\epsilon\text{Sr}$  8–30), as reported by Varela *et al.* (1988) for the ‘old granitoids’. The ‘young granitoids’ of the basement have  $\text{Sr}_{i(1750\text{ Ma})}$  in the range 0.7181–0.7303 ( $\epsilon\text{Sr}$  221–395; Varela *et al.*, 1985, 1988; Dalla Salda *et al.*, 1992).

The initial  $^{143}\text{Nd}/^{144}\text{Nd}$  (Table 7) of the LTi tholeiitic dykes (D1) ranges from 0.51056 to 0.51121 and is slightly lower to higher relative to BE ( $\epsilon\text{Nd}$  -0.4 to 12.2). In comparison, the HTi tholeiitic dykes (D2) are moderately ( $\text{Nd}_i = 0.51052$ ,  $\epsilon\text{Nd} = -1.3$ ) to much ( $\text{Nd}_i = 0.51024$ ,  $\epsilon\text{Nd} = -6.71$ ) lower than BE. D3 and D4 andesitic and rhyolitic dykes have similar  $\text{Nd}_i$  (0.50980–0.50988) and are moderately lower than BE ( $\epsilon\text{Nd} = -2.8$  to  $-4.5$ ).

Table 8: Mineral–liquid partition coefficients used for fractional crystallization modelling (see text for explanation)

|    | Ol/liq | Cpx/liq | Opx/liq | Pl/liq | Mt/liq | Ap/liq |
|----|--------|---------|---------|--------|--------|--------|
| Cr | 0.7    | 6       | 10      | 0.01   | 32     | 0.01   |
| Ni | 5      | 2       | 5       | 0.01   | 10     | 0.01   |
| Rb | 0.01   | 0.02    | 0.02    | 0.07   | 0.01   | 0.56   |
| Ba | 0.01   | 0.02    | 0.01    | 0.23   | 0.01   | 0.95   |
| Sr | 0.01   | 0.06    | 0.04    | 1.8    | 0.01   | 1.67   |
| Nb | 0.01   | 0.005   | 0.15    | 0.01   | 0.4    | 0.09   |
| Zr | 0.01   | 0.1     | 0.18    | 0.048  | 0.1    | 0.01   |
| Y  | 0.01   | 0.9     | 0.18    | 0.03   | 0.2    | 5.08   |
| La | 0.007  | 0.06    | 0.02    | 0.15   | 0.53   | 5.16   |
| Ce | 0.007  | 0.1     | 0.02    | 0.1    | 0.56   | 6.34   |
| Nd | 0.007  | 0.23    | 0.03    | 0.08   | 0.55   | 6.6    |

Ol, olivine; Cpx, Ca-rich pyroxene; Opx, orthopyroxene; Pl, plagioclase; Mt, magnetite; Ap, apatite.

$\epsilon\text{Sr}$ – $\epsilon\text{Nd}$  relationships (Fig. 10) reveal that the HTi-D2 tholeiitic dyke A38 and the calc-alkaline andesitic D3 dykes plot in the enriched quadrant with respect to BE, suggesting an EMI-type isotopic signature, similar to the Florida (Uruguay) and Uauá (Brazil, São Francisco craton) Proterozoic tholeiitic dykes. Conversely, and particularly for sample A4, the LTi-D1 tholeiitic dykes plot in the enriched quadrant in terms of  $\epsilon\text{Sr}$ , but in the depleted one in terms of  $\epsilon\text{Nd}$ .

## PETROGENESIS

### Crystal fractionation

A possible mechanism for interdyke differentiation is fractional crystallization. This process was tested by performing MELTS (Ghiorso & Sack, 1995) and XLFRAC (Stormer & Nicholls, 1978) major element modelling. MELTS can model the crystal fractionation process of a magma on the basis of its major element composition, total pressure,  $f\text{O}_2$  and water content. A Rayleigh fractionation model was then tested to compare calculated vs observed trace element contents (Tables 9–12), using the solid–liquid partition coefficients listed in Table 8.

MELTS results indicate that the mineral compositional variation of D1 LTi tholeiites is compatible with olivine/Ca-poor pyroxene + augite + plagioclase fractional crystallization at low pressure (0.1–0.4 GPa), low  $f\text{O}_2$  (FMQ – 1 log unit) and  $\text{H}_2\text{O}$  <0.3 wt % (Fig. 11). Similarly, XLFRAC results (Table 9) indicate that differentiation from dykes with 8 wt % MgO (A5 and A25) to those with 6–7 wt % MgO (M.E.D1) is

compatible with 27–32% removal of plagioclase (9–12%), augite (15–16%) and olivine (2%, A5) or orthopyroxene (6%, A25). In general, calculated/observed trace element values are satisfactory, with the exception of Rb and Ba. It should be noted that, starting from parental magma A5 (Rb and Ba 7 and 31 ppm, respectively), it is impossible to use fractional crystallization to match the Rb and Ba concentrations of the mean composition of the moderately evolved D1 dykes (i.e. M.E.D1: Rb 106 ppm, Ba 166 ppm), probably as a result of alteration that did not affect elements less sensitive to alteration (e.g. Zr, Nb).

D1a LTi tholeiitic dykes, characterized by high REE, Zr, Y and Nb contents, require parental magmas enriched in these elements. Among the D1 dykes with 8 wt % MgO, only sample A32 has the high content of incompatible elements. XLFRAC calculations (Table 10) indicate that this transition from A32 to the mean composition of D1a dykes is compatible with 47% fractionation of olivine (11%), clinopyroxene (8%), plagioclase (24%) and magnetite (4%). Calculated/observed trace element values are broadly consistent with such a differentiation process. The MELTS program (Fig. 11) indicates that this fractional crystallization may have occurred at low  $f\text{O}_2$  (FMQ – 1 log unit) and 0.1 GPa pressure, in near-anhydrous conditions ( $\text{H}_2\text{O}$  0.1 wt %).

MELTS modelling for the D2 HTi tholeiites (Fig. 11) indicates pigeonite as the dominant fractionating phase, at both ‘high’ (0.8 GPa) and low (0.1 GPa) pressures, for low  $\text{H}_2\text{O}$  (0.1 wt %) and  $f\text{O}_2$  (FMQ – 1 log unit) conditions. XLFRAC mass balances, in conjunction with calculated/observed trace element

Table 9: Mass balance calculations (Stormer & Nicholls, 1978) for fractional crystallization transition from less (A5, A25) to the mean of the more evolved (M.E.D1) D1 tholeiitic dykes (see text for explanation)

|                                | A5    | M.E.D1 | OL(83) | CPX   | PL(71) |    | A5  | calc. | calc./obs. |
|--------------------------------|-------|--------|--------|-------|--------|----|-----|-------|------------|
| SiO <sub>2</sub>               | 50.72 | 50.51  | 39.93  | 52.56 | 50.17  | Cr | 356 | 157   | 0.82       |
| TiO <sub>2</sub>               | 0.99  | 1.15   | 0.00   | 0.09  | 0.05   | Ni | 147 | 122   | 1.01       |
| Al <sub>2</sub> O <sub>3</sub> | 14.66 | 16.02  | 0.00   | 0.35  | 31.52  | Rb | 7   | 10    | 0.09       |
| FeO <sub>tot</sub>             | 10.72 | 11.51  | 14.32  | 11.40 | 0.60   | Ba | 31  | 42    | 0.25       |
| MnO                            | 0.18  | 0.18   | 0.00   | 0.21  | 0.00   | Sr | 138 | 155   | 0.91       |
| MgO                            | 8.02  | 6.71   | 45.00  | 13.09 | 0.00   | Nb | 5   | 7     | 1.39       |
| CaO                            | 12.24 | 10.15  | 0.74   | 22.18 | 14.30  | Zr | 74  | 100   | 1.13       |
| Na <sub>2</sub> O              | 2.17  | 2.29   | 0.00   | 0.13  | 3.34   | Y  | 22  | 26    | 0.94       |
| K <sub>2</sub> O               | 0.19  | 1.36   | 0.00   | 0.00  | 0.02   | La | 5   | 7     | 0.94       |
| P <sub>2</sub> O <sub>5</sub>  | 0.11  | 0.12   | 0.00   | 0.00  | 0.00   | Ce | 19  | 25    | 1.27       |
|                                |       |        |        |       |        | Nd | 10  | 13    | 1.11       |
| wt %                           |       |        | 2.26   | 15.96 | 9.23   |    |     |       |            |
| Σres <sup>2</sup> = 0.75       |       |        |        |       |        |    |     |       |            |
|                                | A25   | M.E.D1 | OPX    | CPX   | PL(71) |    | A25 | calc. | calc./obs. |
| SiO <sub>2</sub>               | 51.34 | 50.51  | 53.28  | 51.90 | 50.17  | Cr | 157 | 37    | 0.20       |
| TiO <sub>2</sub>               | 0.87  | 1.15   | 0.30   | 0.52  | 0.05   | Ni | 110 | 78    | 0.64       |
| Al <sub>2</sub> O <sub>3</sub> | 14.69 | 16.02  | 0.94   | 2.06  | 31.52  | Rb | 96  | 140   | 1.32       |
| FeO <sub>tot</sub>             | 10.56 | 11.51  | 19.40  | 10.82 | 0.60   | Ba | 65  | 93    | 0.56       |
| MnO                            | 0.19  | 0.18   | 0.37   | 0.27  | 0.00   | Sr | 201 | 228   | 1.33       |
| MgO                            | 8.16  | 6.71   | 23.41  | 15.98 | 0.00   | Nb | 3   | 4     | 0.88       |
| CaO                            | 11.21 | 10.15  | 2.28   | 18.28 | 14.30  | Zr | 62  | 88    | 1.00       |
| Na <sub>2</sub> O              | 2.06  | 2.29   | 0.01   | 0.18  | 3.34   | Y  | 20  | 25    | 0.92       |
| K <sub>2</sub> O               | 0.83  | 1.36   | 0.00   | 0.00  | 0.02   | La | 3   | 4     | 0.61       |
| P <sub>2</sub> O <sub>5</sub>  | 0.09  | 0.12   | 0.00   | 0.00  | 0.00   | Ce | 13  | 19    | 0.93       |
|                                |       |        |        |       |        | Nd | 8   | 11    | 0.93       |
| wt %                           |       |        | 6.43   | 14.47 | 11.55  |    |     |       |            |
| Σres <sup>2</sup> = 0.33       |       |        |        |       |        |    |     |       |            |

In parentheses: forsterite and anorthite contents of olivine (OL) and plagioclase (PL); CPX, Ca-rich pyroxene; OPX, orthopyroxene; Calc., calculated; obs., observed; Σres<sup>2</sup>, sum of squares of major element residuals.

values (Table 11), indicate that the transition from less evolved (MgO 7 wt %) to more evolved (MgO 4 wt %) tholeiites requires fractionation of pigeonite (19%) and plagioclase (8%).

The transition from the calc-alkaline D3 andesitic basalts (MgO 7 wt %) to andesites (MgO 4 wt %) requires removal of olivine or orthopyroxene. For the fractional crystallization of andesitic basalt T12, MELTS (Fig. 11) indicates that orthopyroxene is a liquidus phase even at 0.1 GPa [ $fO_2 = FMQ + 2 \log$  unit, H<sub>2</sub>O 1.0 wt %]. Mass balance calculations and calculated/observed trace element abundances (Table

12) indicate that the transition from T12 (andesitic basalt) to A54 (andesite) is compatible with 26% removal of orthopyroxene (7%), clinopyroxene (5%), plagioclase (13%) and magnetite (1%).

XLFRAC results indicate that the silicic D4 dykes may derive from D3 andesites through 73% fractionation of clinopyroxene (36%), plagioclase (28%), magnetite (1%) and alkali-feldspar (8%), but there are great differences between calculated and observed trace element values (Table 12).

Variation diagrams (Figs 6 and 7) and Zr, La vs Ni relationships (Fig. 12) show that dykes D1, D1a,

Table 10: Mass balance calculations (Storner & Nicholls, 1978) for the fractional crystallization transition from the less evolved D1 dyke (A32) with highest incompatible element contents (A32) to the mean of the tholeiitic D1a dykes (see text for explanation)

|                                | A32   | Mean D1a | OL(82) | CPX   | PL(71) | MT    |    | A32 | calc. | calc./obs. |
|--------------------------------|-------|----------|--------|-------|--------|-------|----|-----|-------|------------|
| SiO <sub>2</sub>               | 48.72 | 53.67    | 39.76  | 52.56 | 50.17  | 0.10  | Cr | 144 | 18    | 0.17       |
| TiO <sub>2</sub>               | 0.91  | 1.21     | 0.00   | 0.09  | 0.05   | 16.14 | Ni | 76  | 30    | 1.48       |
| Al <sub>2</sub> O <sub>3</sub> | 15.65 | 15.49    | 0.00   | 0.35  | 31.52  | 1.79  | Rb | 64  | 119   | 2.24       |
| FeO <sub>tot</sub>             | 12.46 | 11.50    | 15.21  | 11.40 | 0.60   | 81.20 | Ba | 318 | 561   | 0.73       |
| MnO                            | 0.2   | 0.18     | 0.00   | 0.21  | 0.00   | 0.70  | Sr | 302 | 321   | 0.97       |
| MgO                            | 8.44  | 4.72     | 44.28  | 13.09 | 0.00   | 0.04  | Nb | 10  | 19    | 1.37       |
| CaO                            | 9.66  | 8.26     | 0.74   | 22.18 | 14.30  | 0.02  | Zr | 96  | 177   | 0.92       |
| Na <sub>2</sub> O              | 2.3   | 2.92     | 0.00   | 0.13  | 3.34   | 0.00  | Y  | 27  | 45    | 1.19       |
| K <sub>2</sub> O               | 1.22  | 1.71     | 0.00   | 0.00  | 0.02   | 0.00  | La | 12  | 21    | 0.53       |
| P <sub>2</sub> O <sub>5</sub>  | 0.44  | 0.35     | 0.00   | 0.00  | 0.00   | 0.00  | Ce | 33  | 58    | 0.80       |
|                                |       |          |        |       |        |       | Nd | 18  | 31    | 0.84       |
| wt %                           |       | 11.07    | 8.43   | 23.57 | 4.46   |       |    |     |       |            |
| Σres <sup>2</sup> = 0.40       |       |          |        |       |        |       |    |     |       |            |

MT, magnetite; other abbreviations and notation as in Table 9.

Table 11: Mass balance calculations (Storner & Nicholls, 1978) for the fractional crystallization transition from the less differentiated (A53) to the more evolved (A39) D2 tholeiitic dykes (see text for explanation)

|                                | A53   | A39   | PIG   | PL(71) |    | A39 | calc. | calc./obs. |
|--------------------------------|-------|-------|-------|--------|----|-----|-------|------------|
| SiO <sub>2</sub>               | 49.58 | 48.11 | 53.59 | 50.17  | Cr | 58  | 24    | 0.41       |
| TiO <sub>2</sub>               | 2.14  | 3.61  | 0.19  | 0.05   | Ni | 24  | 34    | 1.43       |
| Al <sub>2</sub> O <sub>3</sub> | 13.06 | 13.99 | 1.06  | 31.52  | Rb | 47  | 46    | 0.97       |
| FeO <sub>tot</sub>             | 15.25 | 15.83 | 17.46 | 0.60   | Ba | 976 | 633   | 0.65       |
| MnO                            | 0.23  | 0.22  | 0.47  | 0.00   | Sr | 488 | 419   | 0.86       |
| MgO                            | 6.72  | 3.87  | 21.81 | 0.00   | Nb | 51  | 20    | 0.39       |
| CaO                            | 8.71  | 9.17  | 5.41  | 14.30  | Zr | 415 | 267   | 0.64       |
| Na <sub>2</sub> O              | 2.25  | 2.05  | 0.02  | 3.34   | Y  | 66  | 48    | 0.73       |
| K <sub>2</sub> O               | 0.90  | 1.63  | 0.00  | 0.02   | La | 48  | 65    | 1.36       |
| P <sub>2</sub> O <sub>5</sub>  | 1.16  | 1.52  | 0.00  | 0.00   | Ce | 96  | 122   | 1.27       |
|                                |       |       |       |        | Nd | 58  | 74    | 1.27       |
| wt %                           |       | 18.53 | 7.89  |        |    |     |       |            |
| Σres <sup>2</sup> = 0.88       |       |       |       |        |    |     |       |            |

PIG, pigeonite; other abbreviations and notation as in Table 9.

D2 and D3 are related to different parental magmas.

Primary magma compositions for these dykes (P-D1, P-D1a, P-D2 and P-D3, respectively; Table 13) were calculated starting from samples A5 (*mg*-number 60), A32 (*mg*-number 58), A53 (*mg*-number 53) and T12

(*mg*-number 62), respectively, by adding olivine, clinopyroxene and orthopyroxene (in proportions 5:3:1, respectively) to achieve *mg*-number values of 72–73. Plagioclase was not included as fractionating phase, because the MELTS results (Fig. 11) show that

Table 12: Mass balance calculations (Stormer & Nicholls, 1978) for the fractional crystallization transition from the less differentiated (T12) to the more evolved (A54) D3 calc-alkaline dykes, and to silicic (A51) D4 dykes (see text for explanation)

|                                | T12 (D3) | A54 (D3) | OPX   | CPX    | PL(60) | MT     |    | A54 | calc. | calc./obs. |
|--------------------------------|----------|----------|-------|--------|--------|--------|----|-----|-------|------------|
| SiO <sub>2</sub>               | 58.03    | 59.77    | 53.28 | 54.06  | 52.98  | 0.14   | Cr | 45  | 97    | 2.16       |
| TiO <sub>2</sub>               | 0.77     | 0.78     | 0.30  | 0.05   | 0.06   | 22.66  | Ni | 12  | 24    | 2.00       |
| Al <sub>2</sub> O <sub>3</sub> | 13.33    | 14.57    | 0.94  | 0.29   | 29.72  | 2.32   | Rb | 154 | 84    | 0.55       |
| FeO <sub>tot</sub>             | 8.58     | 8.52     | 19.40 | 5.88   | 0.28   | 74.66  | Ba | 517 | 617   | 1.19       |
| MnO                            | 0.18     | 0.16     | 0.37  | 0.21   | 0.00   | 0.02   | Sr | 375 | 583   | 1.55       |
| MgO                            | 7.00     | 3.68     | 23.41 | 14.79  | 0.00   | 0.04   | Nb | 9   | 7     | 0.78       |
| CaO                            | 8.05     | 6.28     | 2.28  | 24.46  | 12.17  | 0.17   | Zr | 135 | 173   | 1.28       |
| Na <sub>2</sub> O              | 2.49     | 3.75     | 0.01  | 0.26   | 4.33   | 0.00   | Y  | 29  | 28    | 0.97       |
| K <sub>2</sub> O               | 1.28     | 2.17     | 0.00  | 0.00   | 0.47   | 0.00   | La | 28  | 39    | 1.39       |
| P <sub>2</sub> O <sub>5</sub>  | 0.29     | 0.32     | 0.00  | 0.00   | 0.00   | 0.00   | Ce | 64  | 83    | 1.30       |
| Total                          | 100.00   | 100.00   | 99.99 | 100.00 | 100.01 | 100.01 | Nd | 34  | 42    | 1.24       |
| wt %                           |          | 6.78     | 4.84  | 13.01  | 1.33   |        |    |     |       |            |
| $\Sigma \text{res}^2 = 0.19$   |          |          |       |        |        |        |    |     |       |            |
|                                | A54 (D3) | A51(D4)  | CPX   | PL(35) | AF     | MT     |    | A51 | calc. | calc./obs. |
| SiO <sub>2</sub>               | 59.77    | 76.21    | 48.06 | 59.97  | 64.02  | 0.14   | Cr | 2   | 2     | 1.00       |
| TiO <sub>2</sub>               | 0.78     | 0.11     | 0.96  | 0.00   | 0.03   | 22.66  | Ni | 5   | 11    | 2.20       |
| Al <sub>2</sub> O <sub>3</sub> | 14.57    | 12.72    | 7.08  | 25.34  | 18.93  | 2.32   | Rb | 141 | 349   | 2.48       |
| FeO <sub>tot</sub>             | 8.52     | 1.88     | 21.42 | 0.00   | 0.30   | 74.66  | Ba | 802 | 1590  | 1.98       |
| MnO                            | 0.16     | 0.03     | 0.62  | 0.00   | 0.00   | 0.02   | Sr | 178 | 292   | 1.64       |
| MgO                            | 3.68     | 0.17     | 8.84  | 0.00   | 0.00   | 0.04   | Nb | 11  | 31    | 2.82       |
| CaO                            | 6.28     | 1.30     | 12.06 | 6.91   | 0.68   | 0.17   | Zr | 169 | 436   | 2.58       |
| Na <sub>2</sub> O              | 3.75     | 3.84     | 0.95  | 7.76   | 0.63   | 0.00   | Y  | 26  | 57    | 2.19       |
| K <sub>2</sub> O               | 2.17     | 3.72     | 0.00  | 0.02   | 15.41  | 0.00   | La | 56  | 91    | 1.63       |
| P <sub>2</sub> O <sub>5</sub>  | 0.32     | 0.02     | 0.00  | 0.00   | 0.00   | 0.00   | Ce | 107 | 204   | 1.91       |
| Total                          | 100.00   | 100.00   | 99.99 | 100.00 | 100.00 | 100.01 | Nd | 49  | 102   | 2.08       |
| wt %                           |          | 35.85    | 28.27 | 8.29   | 0.77   |        |    |     |       |            |
| $\Sigma \text{res}^2 = 0.68$   |          |          |       |        |        |        |    |     |       |            |

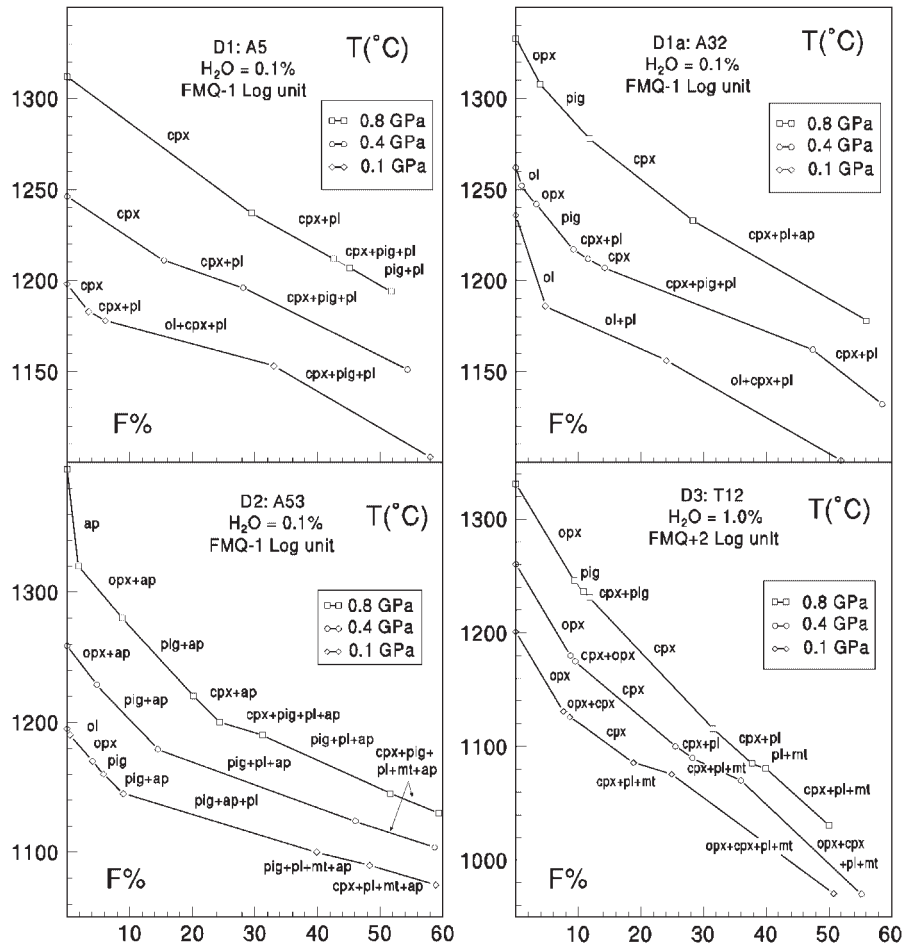
MT, magnetite; AF, alkali feldspar; other abbreviations and notation as in Table 9.

plagioclase is not a liquidus phase in the early stages of crystal fractionation, even at low pressure. The relatively low Cr and Ni contents do not take account of sulfide (pyrite, pyrrhotine, calc-pyrite, pentlandite) fractionation (Echeveste & Fernandez, 1994).

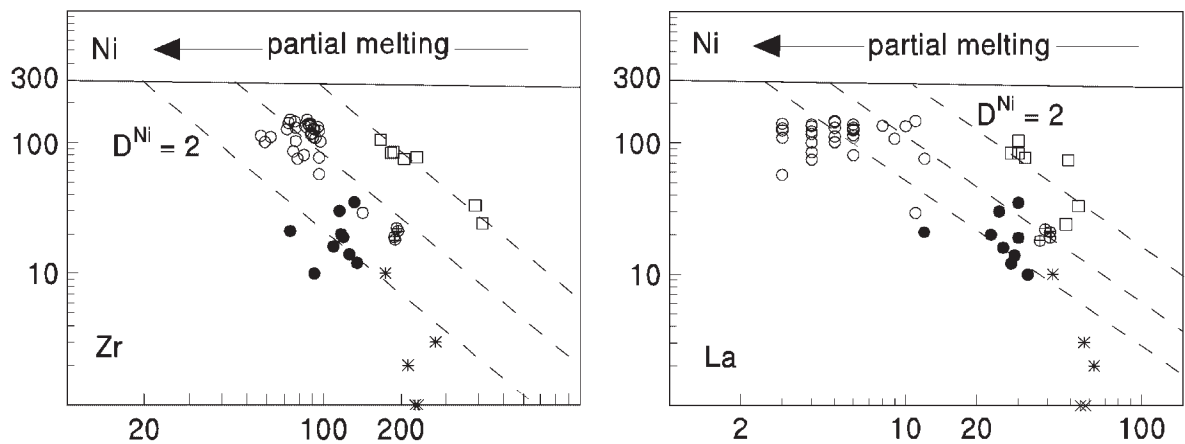
### Crustal contamination

In general, the Azul and Tandil dykes have  $Sr_i$  higher than that of BE. The calc-alkaline D3 and D4 dykes are characterized by IE and  $Sr_i$  values (0.7029–0.7050) lower than or similar to those of the ‘old granitoids’ (0.7034–0.7060; Varela *et al.*, 1988) and therefore

crustal contamination is not easy to detect. D1 and D2 tholeiitic dykes have  $Sr_i$  ratios ranging from 0.7031 to 0.7111. Figure 13 reveals that  $Sr_i$  increases in both D1 and D2 dykes are not correlated with SiO<sub>2</sub>, MgO, K<sub>2</sub>O, Rb, Ba, La, Zr and Nb. It should be noted that the country rocks have  $Sr_i$  ratios <0.7070 and cannot represent end-members in terms of simple mixing. Therefore, we modelled AFC contamination (DePaolo, 1981) for D1 and D2 dykes assuming the ‘old’ and ‘young granitoids’ of Varela *et al.* (1988) as contaminants and as starting compositions those of the D1 (A2, A5) and D2 (A53, A38) dykes with the lowest  $Sr_i$  values. Calculations were carried out with bulk  $D(Sr)$  partition



**Fig. 11.** Percentages of solid removal ( $F\%$ ) vs temperature ( $T(^{\circ}\text{C})$ ) according to the MELTS fractional crystallization model of Ghiorso & Sack (1995) for the Azul and Tandil dyke swarms. ol, olivine; opx, orthopyroxene; cpx, augite; pig, pigeonite; pl, plagioclase; ap, apatite; mt, magnetite. FMQ, fayalite–magnetite–quartz buffer. (See text for explanation.)



**Fig. 12.** Log La and Zr (ppm) vs log Ni (ppm) for the Azul and Tandil dyke swarms. Symbols as in Fig. 7. Dashed straight lines: crystal fractionation trends for bulk  $D^{\text{Ni}}$  partition coefficient = 2. It should be noted that, assuming the same parental source mantle, D1 dykes were generated by a lower melting degree for Ni–Zr but not for Ni–La ratios relative to D3 dykes.

Table 13: Calculated 'primary' compositions of D1 (P-D1), D1a (P-D1a), D2 (P-D2) and D3 (P-D3) dykes, starting from the bulk-rock compositions of A5, A32, A53 and T12, respectively

|                                | P-D1  | P-D1a | P-D2  | P-D3  | OL (89) | CPX   | OPX   |
|--------------------------------|-------|-------|-------|-------|---------|-------|-------|
| SiO <sub>2</sub>               | 49.71 | 48.03 | 48.45 | 56.25 | 40.69   | 52.50 | 56.23 |
| TiO <sub>2</sub>               | 0.81  | 0.72  | 1.43  | 0.68  | 0.00    | 0.36  | 0.00  |
| Al <sub>2</sub> O <sub>3</sub> | 11.92 | 12.13 | 8.98  | 11.60 | 0.00    | 3.71  | 1.71  |
| FeO <sub>tot</sub>             | 10.38 | 11.60 | 13.10 | 8.65  | 10.71   | 6.00  | 10.53 |
| MnO                            | 0.14  | 0.15  | 0.15  | 0.15  | 0.00    | 0.00  | 0.00  |
| MgO                            | 13.87 | 15.35 | 17.12 | 11.31 | 48.60   | 17.62 | 31.53 |
| CaO                            | 11.21 | 9.04  | 7.95  | 7.89  | 0.00    | 19.67 | 0.00  |
| Na <sub>2</sub> O              | 1.73  | 1.74  | 1.48  | 2.14  | 0.00    | 0.14  | 0.00  |
| K <sub>2</sub> O               | 0.15  | 0.92  | 0.58  | 1.09  | 0.00    | 0.00  | 0.00  |
| P <sub>2</sub> O <sub>5</sub>  | 0.09  | 0.33  | 0.75  | 0.25  | 0.00    | 0.00  | 0.00  |
| mg-no.                         | 72.92 | 72.80 | 72.55 | 72.56 |         |       |       |
| Cr                             | 600   | 269   | 439   | 503   |         |       |       |
| Ni                             | 288   | 174   | 270   | 56    |         |       |       |
| Rb                             | 6     | 48    | 22    | 54    |         |       |       |
| Ba                             | 25    | 240   | 312   | 406   |         |       |       |
| Sr                             | 110   | 229   | 241   | 489   |         |       |       |
| Nb                             | 4     | 8     | 10    | 4     |         |       |       |
| Zr                             | 59    | 73    | 137   | 114   |         |       |       |
| Y                              | 19    | 22    | 28    | 20    |         |       |       |
| La                             | 3     | 9     | 32    | 26    |         |       |       |
| Ce                             | 8     | 25    | 60    | 54    |         |       |       |
| Nd                             | 6     | 14    | 37    | 28    |         |       |       |

Primary compositions were calculated to match at.  $100 \times \text{Mg}/(\text{Mg} + \text{Fe}^{2+})$  of 73. The following mineral amounts were added: P-D1, A5 + 15% OL + 10% CPX + 1.5% OPX; P-D1a, A32 + 19% OL + 12% CPX + 2% OPX; P-D2, A53 + 30% OL + 18% CPX + 6% OPX; P-D3, T12 + 10% OL + 6% CPX + 1% OPX. Abbreviations and notation as in Table 9.

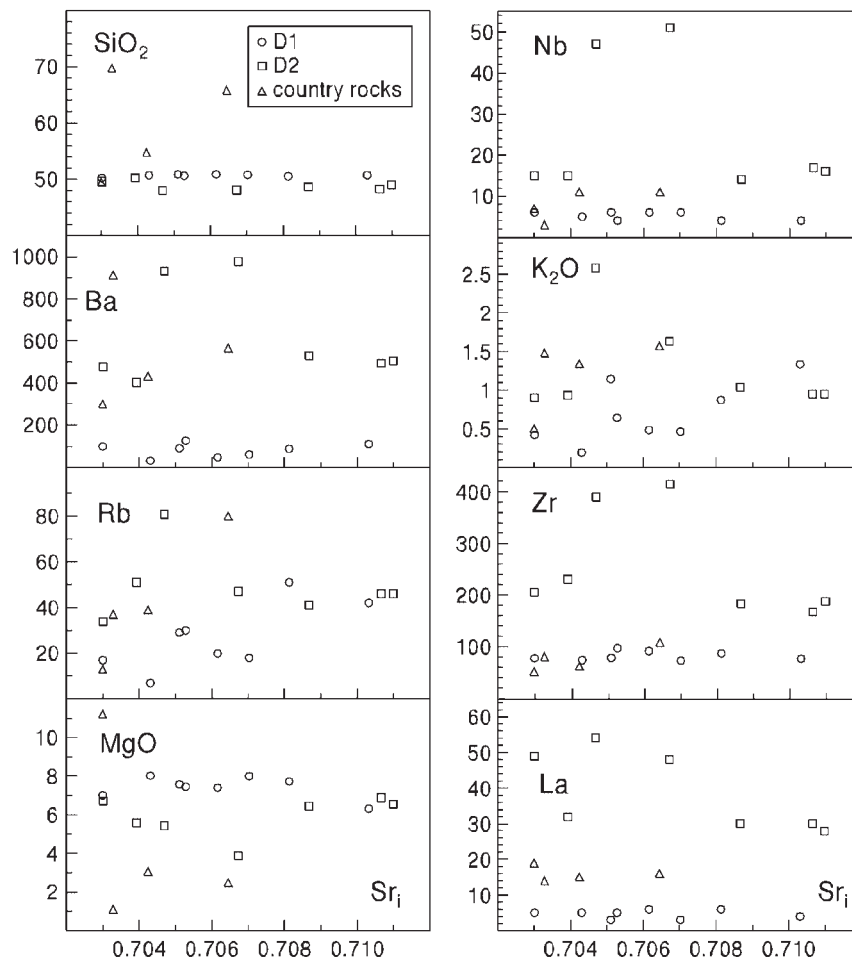
coefficients of 0.5 and 0.9, considering that plagioclase was not a liquidus phase in the earliest stages of differentiation (Fig. 11). The results (Fig. 14) indicate that Sr<sub>i</sub> variations in both LTi-D1 and HTi-D2 dykes are compatible with assimilation of small amounts ( $r = 0.2$ ) of both 'young' and 'old granitoids' (Cavazzini, 1996). It should be noted, however, that D1 and D2 dykes with Sr<sub>i</sub> values >0.707 do not conform to AFC mixing curves, even for residual liquid fractions as low as 0.2–0.3. This suggests that alteration may have contributed towards reaching the highest Sr<sub>i</sub> values. In summary, major and trace elements, mass balance calculations and isotopic data all indicate that D1 and D2 tholeiitic dykes with Sr<sub>i</sub> <0.7050 were not significantly affected by crustal contamination.

### Source mantle

The melting degree for the 'primary' D1 to D3 basalts was calculated (XLFRAC) using a peridotitic major

element composition from Ringwood (1966; pyrolite), Chen (1971) and Ionov & Hofmann (1995). Calculations were carried out for dry and amphibole-bearing spinel- and garnet-peridotite mineral assemblages. The results (Table 14) indicate that the highest degrees of melting (15–9% and 14–6% for garnet and spinel peridotites) refer to the calc-alkaline D3 dykes, whereas the lowest ones (7–3% and 6–4%: garnet and spinel peridotites, respectively) apply to the high-TiO<sub>2</sub> D2 tholeiitic dykes (Fig. 15).

Incompatible trace elements (IE) in the peridotitic sources (Table 15) were computed (batch melting; Hanson, 1978) using the partition coefficients of Table 16 and the calculated degrees of melting and solid residua (Table 14). The multi-elemental diagrams of Fig. 16 reveal that the IE of D1 peridotitic sources are depleted with respect to PM of Sun & McDonough (1989), irrespective of whether such sources are enriched (i.e. pyrolite) or relatively depleted in terms of major



**Fig. 13.** Initial (1565 Ma)  $^{87}\text{Sr}/^{86}\text{Sr}$  ( $\text{Sr}_i$ ) ratios vs  $\text{MgO}$ ,  $\text{K}_2\text{O}$  and  $\text{SiO}_2$  (wt %), and Rb, Ba, La, Zr and Nb (ppm).

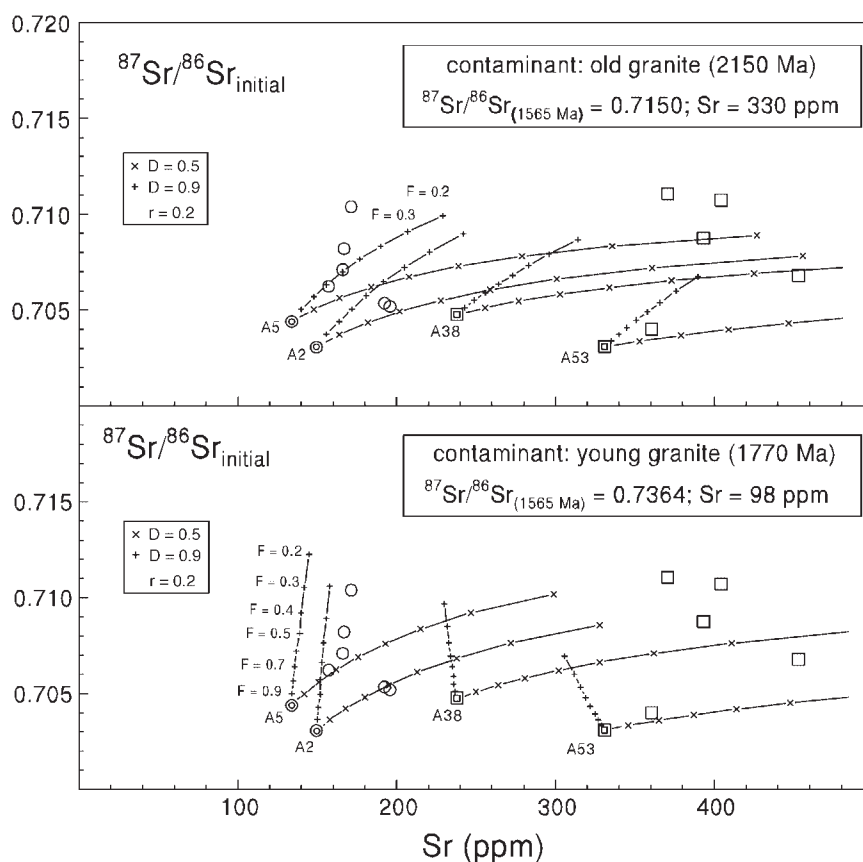
elements. The relatively low IE contents of a major-element pyrolite-type mantle indicate that IE depletion of such peridotites reflects a low degree (e.g. <1%) of melting, which did not significantly change the original mineral assemblage or major element composition. D1a sources (Fig. 16b) have IE higher than D1 sources, are especially enriched in Rb, Ba and K, and display negative Nb and Sr spikes and positive P spikes. D2 source patterns (Fig. 16c) are similar to those of D1a and have distinct P and Ti positive spikes, as well as a more pronounced Sr negative anomaly. D3 source patterns (Fig. 16d) are globally enriched with respect to PM and are characterized by high Nb and less marked Ti negative anomalies. The high (LILE, LREE)/(HFSE, HREE) values of D3 calc-alkaline dykes with respect to D1 tholeiitic dykes do not necessarily imply different mantle peridotite residua, i.e. garnet- vs spinel-peridotites (see Fig. 15 for melting degrees). Such incompatible element ratio differences may reflect LILE–LREE enrichment in D3 mantle source as a

result of ‘fluid/small-volume melts’ metasomatism that may have occurred in Archaean–Proterozoic times [ $T_{\text{DM}}(\text{Nd}) \sim 2.5 \text{ Ga}$ ].

## DISCUSSION AND CONCLUSIONS

### Early Proterozoic dykes

The oldest Precambrian dyke magmatism of the Rio de La Plata craton outcrops in Argentina (Tandil region) and is dated at 2.0 Ga. The dykes crosscut the crystalline basement and are sub-coeval with the youngest ‘old granitoids’ that intruded the basement (2.2–2.0 Ga). This magmatism (andesites: D3; rhyolites: D4), intruded during the transamazonian stages of the Transamazonian Orogeny, has calc-alkaline geochemical features including negative Nb–Ti spikes. The calculated D3 primary andesite magmas may be derived (major elements) from either anhydrous or amphibole-bearing peridotites by 5–15% batch melting, in the



**Fig. 14.** Sr (ppm) vs initial  $^{87}\text{Sr}/^{86}\text{Sr}$  for D1 (○) and D2 (□) tholeiitic LTi and HTi, respectively, dykes from the Azul and Tandil regions. Curves calculated according to the AFC crustal contamination model of DePaolo (1981).  $D$ , bulk Sr partition coefficient;  $r$ , rate of assimilated/crystallized materials;  $F$ , liquid fraction. Contaminants: 'old granitoids' is sample 99 of Varela *et al.* (1988); 'young granitoids' is sample 40 of Varela *et al.* (1988).

spinel and garnet facies. The high LREE/HREE ratios of the D3 dykes are compatible with garnet-peridotite residua as well as with spinel peridotites enriched in LREE by 'metasomatic processes' involving fluids and/or small-volume melts related to peridotites contaminated by eclogite-derived melts (see Rivalenti *et al.*, 1998). These 'metasomatic processes' imply interaction with subducted rutile-bearing crustal material (e.g. Ryerson & Watson, 1987; Brenan *et al.*, 1994) and possibly ilmenite in the residua (Ayers *et al.*, 1997; Ayers, 1998). The anhydrous primary mineral assemblages of the andesitic dykes indicate that melting occurred in near-anhydrous water-undersaturated conditions. The transition from andesites to rhyolites (major and trace elements) is only partly consistent with fractional crystallization.

The initial Sr isotopic ratios of the andesites and rhyolites are similar to those of the country rocks and the 'old granitoids', and are higher than that of BE; crustal contamination was not appreciable. In terms of  $\epsilon\text{Sr}$  and  $\epsilon\text{Nd}$ , these dykes, as well as those of Uauá

(São Francisco craton, Brazil, 2.2–2.0 Ga), plot in the enriched quadrant and trend towards the EMI mantle component of Hart & Zindler (1989). Assuming that the Sm/Nd ratio was not significantly changed by melting, the  $T_{\text{DM}}(\text{Nd})$  of these Argentinian dykes indicate that they could be related to source mantle that underwent metasomatic processes responsible for their EMI Sr–Nd isotopic signature in Archaean–Proterozoic times ( $\sim 2.5$  Ga).

### Middle Proterozoic dykes

The youngest Precambrian dyke magmatism of the Rio de La Plata craton in Argentina (Azul and Tandil regions) dates at  $\sim 1.6$  Ga, an age similar to that (1.7–1.8 Ga; Bossi *et al.*, 1993; Teixeira *et al.*, 1999) of the oldest dykes from Uruguay (Florida region). The Argentina dyke swarms intrude both the 'old' and 'young granitoids' and have compositional characteristics typical of tholeiitic suites, with low and high

Table 14: Weight per cent mineral assemblages calculated (Stormer & Nicholls, 1978) from peridotite compositions of Ringwood (1966; P1 and P2), Chen (1971; P3 and P4) and Ionov & Hofmann (1995; P5 and P6), and mineral assemblages of residua after extraction of P-D1, P-D1a, P-D2 and P-D3 primary magmas (see text for explanation)

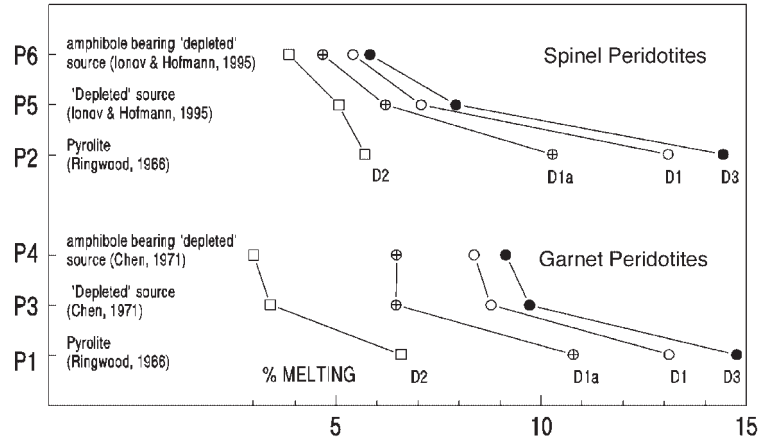
|                      | Peridotites        |                    |                    |                    |                    |                    |
|----------------------|--------------------|--------------------|--------------------|--------------------|--------------------|--------------------|
|                      | P1                 | P2                 | P3                 | P4                 | P5                 | P6                 |
| OL                   | 56.23 <sup>1</sup> | 49.52 <sup>1</sup> | 65.28 <sup>1</sup> | 65.22 <sup>1</sup> | 55.98 <sup>2</sup> | 59.78 <sup>2</sup> |
| OPX                  | 15.61 <sup>3</sup> | 28.87 <sup>3</sup> | 18.38 <sup>1</sup> | 18.99 <sup>1</sup> | 26.94 <sup>2</sup> | 20.28 <sup>2</sup> |
| CPX                  | 16.35 <sup>3</sup> | 17.20 <sup>3</sup> | 10.17 <sup>1</sup> | 9.16 <sup>1</sup>  | 13.78 <sup>2</sup> | 14.24 <sup>4</sup> |
| GT                   | 11.81 <sup>1</sup> |                    | 6.16 <sup>1</sup>  | 4.35 <sup>1</sup>  |                    |                    |
| SP                   |                    | 4.42 <sup>3</sup>  |                    |                    | 4.30 <sup>2</sup>  | 2.84 <sup>2</sup>  |
| AMP                  |                    |                    |                    | 2.03 <sup>5</sup>  |                    | 2.86 <sup>4</sup>  |
| $\Sigma\text{res}^2$ | 0.62               | 0.59               | 0.06               | 0.04               | 0.41               | 0.16               |
| <i>D1 residua</i>    |                    |                    |                    |                    |                    |                    |
| OL                   | 67.41              | 64.01              | 73.45              | 73.01              | 63.08              | 64.48              |
| OPX                  | 17.58              | 24.27              | 18.33              | 18.40              | 22.79              | 18.85              |
| CPX                  | 8.93               | 9.50               | 7.02               | 7.07               | 10.75              | 11.79              |
| GT                   | 6.08               |                    | 1.20               | 1.25               |                    |                    |
| SP                   |                    | 2.23               |                    |                    | 3.38               | 2.37               |
| AMP                  |                    |                    |                    | 0.26               |                    | 2.51               |
| $\Sigma\text{res}^2$ | 0.44               | 0.50               | 0.02               | 0.02               | 0.12               | 0.01               |
| <i>F</i>             | 13.09              | 13.06              | 8.76               | 8.34               | 7.06               | 5.40               |
| <i>D1a residua</i>   |                    |                    |                    |                    |                    |                    |
| OL                   | 64.39              | 60.28              | 70.70              | 70.72              | 61.81              | 63.58              |
| OPX                  | 16.95              | 24.20              | 18.47              | 18.45              | 23.09              | 19.07              |
| CPX                  | 12.30              | 13.08              | 8.61               | 8.61               | 11.70              | 12.94              |
| GT                   | 6.37               |                    | 2.23               | 2.23               |                    |                    |
| SP                   |                    | 2.44               |                    |                    | 3.40               | 2.45               |
| AMP                  |                    |                    |                    |                    |                    | 1.95               |
| $\Sigma\text{res}^2$ | 0.43               | 0.49               | 0.01               | 0.01               | 0.11               | 0.01               |
| <i>F</i>             | 10.75              | 10.25              | 6.45               | 6.46               | 6.19               | 4.67               |
| <i>D2 residua</i>    |                    |                    |                    |                    |                    |                    |
| OL                   | 60.60              | 52.75              | 67.87              | 67.53              | 60.17              | 62.48              |
| OPX                  | 15.89              | 27.38              | 18.22              | 18.39              | 27.78              | 19.47              |
| CPX                  | 14.20              | 16.01              | 9.43               | 9.26               | 12.27              | 13.49              |
| GT                   | 9.31               |                    | 4.49               | 4.24               |                    |                    |
| SP                   |                    | 3.87               |                    |                    | 3.77               | 2.72               |
| AMP                  |                    |                    |                    | 0.58               |                    | 1.84               |
| $\Sigma\text{res}^2$ | 0.44               | 0.55               | 0.02               | 0.02               | 0.12               | 0.01               |
| <i>F</i>             | 6.57               | 5.69               | 3.40               | 2.98               | 5.06               | 3.84               |
| <i>D3 residua</i>    |                    |                    |                    |                    |                    |                    |
| OL                   | 75.49              | 71.28              | 78.32              | 76.96              | 66.86              | 66.94              |
| OPX                  | 4.42               | 12.06              | 12.36              | 13.05              | 17.93              | 15.85              |
| CPX                  | 13.57              | 14.21              | 8.47               | 8.62               | 11.74              | 13.73              |
| GT                   | 6.52               |                    | 0.85               | 1.36               |                    |                    |
| SP                   |                    | 2.45               |                    |                    | 3.48               | 2.75               |
| AMP                  |                    |                    |                    |                    |                    | 0.73               |
| $\Sigma\text{res}^2$ | 0.34               | 0.39               | 0.01               | 0.01               | 0.08               | 0.01               |
| <i>F</i>             | 13.09              | 13.06              | 8.76               | 8.34               | 7.06               | 5.40               |

<sup>1</sup>MacGregor (1974); <sup>2</sup>Ionov & Hofmann (1995); <sup>3</sup>Ringwood (1966); <sup>4</sup>Takazawa *et al.* (1996); <sup>5</sup>Wilkinson & Le Maitre (1987). OL, olivine; OPX, orthopyroxene; CPX, Ca-rich pyroxene; GT, garnet; SP, spinel; AMP, amphibole; *F*, percentage of liquid fraction;  $\Sigma\text{res}^2$ , sum of squares of major element residuals.

Table 15: Calculated incompatible element (batch melting; Hanson, 1978) mantle sources, for tholeiitic (P-D1, P-D1a, P-D2) and calc-alkaline (P-D3) calculated primary magmas from the Azul and Tandil regions

|                      | P1      | P2      | P3      | P4      | P5      | P6      |
|----------------------|---------|---------|---------|---------|---------|---------|
| <i>P-D1 sources</i>  |         |         |         |         |         |         |
| Rb                   | 0.79    | 0.79    | 0.53    | 0.51    | 0.43    | 0.38    |
| Ba                   | 3.28    | 3.27    | 2.20    | 2.12    | 1.78    | 1.60    |
| K                    | 207.31  | 206.90  | 138.88  | 136.02  | 112.47  | 123.87  |
| Nb                   | 0.53    | 0.53    | 0.36    | 0.35    | 0.29    | 0.30    |
| La                   | 0.41    | 0.41    | 0.27    | 0.26    | 0.23    | 0.19    |
| Ce                   | 1.10    | 1.10    | 0.75    | 0.72    | 0.64    | 0.56    |
| Sr                   | 15.24   | 15.30   | 10.35   | 9.92    | 8.77    | 7.30    |
| Nd                   | 0.89    | 0.88    | 0.61    | 0.59    | 0.54    | 0.51    |
| P                    | 69.61   | 67.74   | 46.23   | 44.50   | 40.07   | 34.67   |
| Zr                   | 9.59    | 8.81    | 6.16    | 5.97    | 5.39    | 4.90    |
| Ti                   | 1321.04 | 1205.47 | 903.49  | 892.94  | 902.86  | 894.35  |
| Y                    | 5.17    | 3.39    | 2.75    | 2.70    | 2.40    | 2.21    |
| <i>P-D1a sources</i> |         |         |         |         |         |         |
| Rb                   | 5.21    | 4.99    | 3.15    | 3.15    | 3.04    | 2.59    |
| Ba                   | 25.89   | 24.70   | 15.57   | 15.60   | 14.95   | 13.09   |
| K                    | 825.47  | 789.36  | 497.09  | 497.85  | 478.84  | 505.47  |
| Nb                   | 0.87    | 0.83    | 0.53    | 0.53    | 0.51    | 0.51    |
| La                   | 1.01    | 0.98    | 0.62    | 0.62    | 0.61    | 0.51    |
| Ce                   | 2.87    | 2.82    | 1.79    | 1.79    | 1.79    | 1.56    |
| Sr                   | 26.13   | 25.67   | 16.32   | 16.34   | 16.27   | 13.44   |
| Nd                   | 1.76    | 1.75    | 1.14    | 1.14    | 1.16    | 1.09    |
| P                    | 175.66  | 167.93  | 109.22  | 109.37  | 109.14  | 93.78   |
| Zr                   | 10.01   | 9.00    | 6.18    | 6.18    | 6.02    | 5.41    |
| Ti                   | 870.42  | 817.47  | 613.18  | 613.47  | 638.20  | 625.63  |
| Y                    | 5.55    | 3.62    | 3.24    | 3.24    | 2.68    | 2.48    |
| <i>P-D2 sources</i>  |         |         |         |         |         |         |
| Rb                   | 2.46    | 1.74    | 1.27    | 1.21    | 1.71    | 1.47    |
| Ba                   | 25.73   | 18.17   | 13.30   | 12.69   | 17.83   | 15.78   |
| K                    | 410.57  | 299.14  | 212.04  | 213.91  | 284.66  | 306.98  |
| Nb                   | 0.88    | 0.64    | 0.46    | 0.45    | 0.61    | 0.62    |
| La                   | 1.70    | 1.24    | 0.90    | 0.85    | 1.20    | 1.01    |
| Ce                   | 4.35    | 3.28    | 2.34    | 2.23    | 3.12    | 2.75    |
| Sr                   | 26.34   | 19.81   | 14.27   | 13.40   | 19.06   | 15.91   |
| Nd                   | 2.73    | 2.13    | 1.54    | 1.49    | 1.96    | 1.87    |
| P                    | 276.17  | 199.94  | 150.41  | 142.81  | 190.17  | 165.38  |
| Zr                   | 13.66   | 9.13    | 7.58    | 7.31    | 8.55    | 7.78    |
| Ti                   | 1588.77 | 1249.69 | 1035.67 | 1024.57 | 1121.96 | 1104.56 |
| Y                    | 5.13    | 2.34    | 2.88    | 2.80    | 2.08    | 1.96    |
| <i>P-D3 sources</i>  |         |         |         |         |         |         |
| Rb                   | 8.03    | 7.86    | 5.29    | 4.98    | 4.34    | 3.34    |
| Ba                   | 59.89   | 58.61   | 39.49   | 37.13   | 32.23   | 24.91   |
| K                    | 1339.59 | 1311.02 | 881.87  | 829.50  | 721.75  | 597.32  |
| Nb                   | 0.60    | 0.58    | 0.39    | 0.37    | 0.32    | 0.26    |
| La                   | 4.00    | 3.92    | 2.63    | 2.48    | 2.20    | 1.73    |
| Ce                   | 8.50    | 8.35    | 5.59    | 5.29    | 4.76    | 3.85    |
| Sr                   | 75.99   | 74.86   | 50.34   | 47.59   | 42.77   | 33.73   |
| Nd                   | 4.78    | 4.65    | 3.13    | 2.98    | 2.76    | 2.38    |
| P                    | 179.97  | 172.46  | 115.98  | 110.39  | 100.63  | 81.70   |
| Zr                   | 20.45   | 18.58   | 12.70   | 12.24   | 11.16   | 9.33    |
| Ti                   | 978.11  | 886.50  | 640.50  | 634.73  | 637.91  | 598.60  |
| Y                    | 6.04    | 4.02    | 2.99    | 3.07    | 2.68    | 2.45    |

P1 to P6 as in Table 14. (See text for explanation.)



**Fig. 15.** Melting degrees for tholeiitic (D1, D1a and D2) and calc-alkaline (D3) calculated primary magmas for anhydrous and amphibole-bearing source mantle in the spinel and garnet-facies. (See text for explanation.)

*Table 16: Mineral/liquid partition coefficients used for melting modelling*

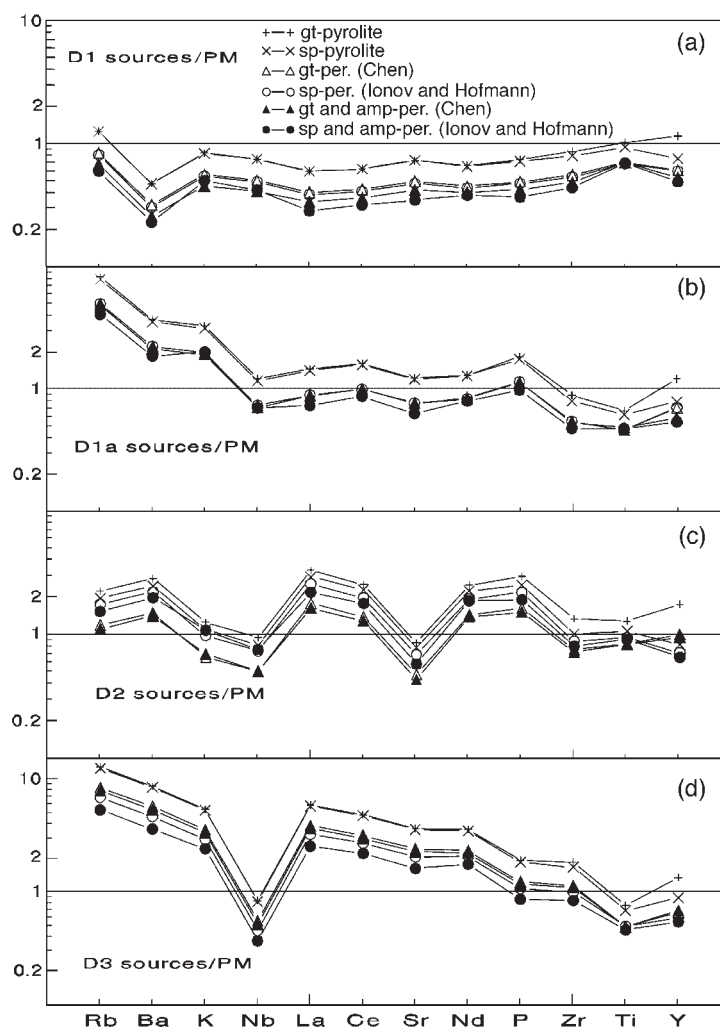
|    | Ol/liq   | Opx/liq | Cpx/liq | Gt/liq  | Sp/liq   | Amp/liq |
|----|----------|---------|---------|---------|----------|---------|
| Rb | 0.00018  | 0.0006  | 0.011   | 0.0007  | 0.0001   | 0.3     |
| Ba | 0.00032  | 0.00069 | 0.0007  | 0.00001 | 0.00001  | 0.4     |
| K  | 0.000001 | 0.00001 | 0.0072  | 0.00001 | 0.000001 | 1       |
| Nb | 0.0001   | 0.003   | 0.0077  | 0.01    | 0.01     | 0.8     |
| La | 0.000007 | 0.0005  | 0.0536  | 0.01    | 0.0006   | 0.17    |
| Ce | 0.00001  | 0.0009  | 0.0858  | 0.004   | 0.0006   | 0.26    |
| Sr | 0.00019  | 0.007   | 0.067   | 0.0011  | 0.005    | 0.12    |
| Nd | 0.00007  | 0.009   | 0.173   | 0.057   | 0.0006   | 0.44    |
| P  | 0.0008   | 0.004   | 0.1     | 0.1     | 0.05     | 0.2     |
| Zr | 0.0007   | 0.02    | 0.1234  | 0.3     | 0.07     | 0.35    |
| Ti | 0.015    | 0.14    | 0.384   | 0.6     | 0.15     | 0.69    |
| Y  | 0.001    | 0.06    | 0.4     | 1.9     | 0.07     | 0.2     |

Source data from McKenzie & O'Nions (1991) and Kelemen *et al.* (1993). Ol, olivine; Cpx, Ca-rich pyroxene; Opx, orthopyroxene; Gt, garnet; Sp, spinel; Amp, amphibole.

TiO<sub>2</sub> contents (LTi and HTi, respectively). Most LTi tholeiites (D1 dykes) have REE patterns similar to those of E-MORB, except for a few tholeiitic andesitic basalts (D1a dykes) characterized by higher LREE/HREE ratios (e.g. La/Yb<sub>CN</sub> ~7). D1 dykes have slightly positive or negative Nb spikes and HREE lower than those of E-MORB, whereas D1a dykes have a distinct negative Nb spike. The dykes with high TiO<sub>2</sub> and incompatible elements (D2) are characterized by high LREE/HREE and slightly positive or negative Nb spikes. The Middle Proterozoic dykes cannot be related to the same parental magma, but require different primary melts originating from different mantle sources. The genesis of D1 tholeiites requires

5–13% melting of peridotites depleted in incompatible elements relative to PM; that of D1a (5–11% melting) and D2 (3–7% melting) dykes would be related to source mantle enriched in LILE and LREE with respect to PM. The tholeiitic dykes that can be considered unaffected by appreciable crustal contamination are characterized by initial <sup>143</sup>Nd/<sup>144</sup>Nd ranging from slightly lower to higher than BE (D2 and D1 dykes, respectively).

In summary, petrological, geochemical and Sr–Nd isotope data support the hypothesis that the various Middle Proterozoic tholeiitic dykes require distinct mantle sources, ranging from IE-depleted (basalt melt extraction) to IE-enriched ('metasomatic processes')



**Fig. 16.** Multi-elemental plots of calculated (batch melting; Hanson, 1978) mantle sources, normalized to Primitive Mantle (PM) of Sun & McDonough (1989), for tholeiitic (D1, D1a, D2) and calc-alkaline (D3) calculated primary magmas from the Azul and Tandil regions. (See text for explanation.)

peridotites. It should be noted that, in a very restricted area, particularly in Tandil (Fig. 2), both calc-alkaline and tholeiitic dykes occur; this indicates important small-scale mantle heterogeneity.

Lastly, it is notable that the Middle Proterozoic tholeiitic dykes from Uruguay (Florida region, Rio de La Plata craton) have geochemical features (mainly trace elements) similar to their D1 Argentina analogues (Figs 6 and 7).

### Tectonic implications

The Early Proterozoic calc-alkaline dykes of Tandil are sub-coeval ( $\sim 2.0$  Ga) with the youngest intrusion of 'old granitoids' and represent transtensional stages of the Transamazonian Orogeny. On the other hand,

the emplacement of Middle Proterozoic tholeiitic dykes in the Azul and Tandil regions relates to the development of the Rio de La Plata craton, where these dykes are sub-coeval ( $\sim 1.6$  Ga) with the youngest intrusion of the 'young granitoids' and are associated with extensional tectonics during a post-collisional Transamazonian stage (Dalla Salda *et al.*, 1988, 1992), 300–400 m.y. after the end of the Transamazonian Orogeny. Rogers (1996) proposed a time of 2.0 Ga for the existence of the 'Atlantica' continent, which includes the cratons of Amazonia (Guaporè and Guyana), São Francisco, Congo–Kasai, West Africa and, possibly, West Nile and Rio de La Plata. In our opinion, Rogers' reconstruction should take into consideration the spatial position of the southwestern Africa region. Volcanic and granitic rocks from the 'Richtersveld Subprovince', within the Namaqua Fold

Belts facing the Rio de La Plata craton in the Gondwana reconstruction (Fig. 1), have been related to the Eburnean Orogeny at  $\sim 2.0$  Ga (Hartnady *et al.*, 1985; Reid, 1997) and are, therefore, coeval with the calc-alkaline rocks of Tandil and Azul. The African calc-alkaline rocks were subjected to metamorphic processes related to the Kibaran Orogeny (1.2–1.0 Ga; Hartnady *et al.*, 1985; Reid, 1997; Colliston & Schoch, 1998), but not appreciable in the corresponding Argentinian rocks. Also, within the Namaqua fold belts, in the Witberg–Aggeney–Gamsberg ore district, metamorphosed basic tholeiites similar to the Tandil and Florida dykes are present whose emplacement occurred at 1.6–1.7 Ga (Reid *et al.*, 1987). The metamorphism of the African basalts (amphibolite facies) has been related to the Kibaran Orogeny (Reid, 1997). The similarities between the Proterozoic calc-alkaline and tholeiitic rocks of Rio de La Plata and the southwestern Africa analogues suggest that these two regions were contiguous in Early Proterozoic as well as in Late Proterozoic times, consistent with the common sedimentary basin of La Tinta and Nama Groups (Dalla Salda, 1982).

In a global context, the generation of juvenile crust ('granitoids') within the Rio de La Plata craton occurred from 2.2 to 1.6 Ga. This time broadly corresponds to that attributed to the development of the Early Proterozoic Supercontinent proposed by Condie (1998) on the basis of the occurrence of greenstone belts. Yale & Carpenter (1998) partly confirmed this hypothesis, emphasizing the absence of giant dyke swarms in the periods that preceded and followed 2.4–2.2 and 1.6–1.4 Ga, respectively. These time gaps may well represent periods of continental dispersal after supercontinent break-up.

## ACKNOWLEDGEMENTS

Special thanks are due to A. Cundari and A. Marzoli for fruitful suggestions related to the manuscript, and to the constructive reviewer's comments and criticism. L. Furlan (Trieste University) and A. Giaretta and A. Carampin (Padova University) are acknowledged for the valuable technical collaboration in this research. The authors acknowledge the financial support by FAPESP and CNPq (Brazilian Agencies) and CNR and MURST (Italian Agencies).

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