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This work presents a sustainable approach to produce lithium carbonate from brines and DOI: 10.1039/D5SU00552C simultaneously capture carbon dioxide. The findings contribute to advancing sustainable mining technologies and support progress toward 4 UN Sustainable Development Goals: Affordable and Clean Energy (SDG 7); Industry, Innovation and Infrastructure (SDG 9); Responsible Consumption and Production (SDG 12); and Life on land (SDG 15). Lithium is fundamental for the energy transition. It would be an inconsistency if its increased production would be associated to non-sustainable practices. The new brine processing methodology fully avoids the use of soda ash for lithium carbonate precipitation. This is replaced by CO<sub>2</sub> absorption in an alkaline media produced via water electrolysis. CO<sub>2</sub> is permanently captured and stored in the mineralized products.

# Towards a higher level of circularity in lithium brine mining: CO<sub>2</sub> absorption in concentrate of size of the size

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#### **ABSTRACT**

The transition to sustainable lithium production from brines requires innovations that address chemical consumption, water use, and carbon emissions. This work proposes a novel six-step treatment of real, highly saline lithium-rich brine to simultaneously recover lithium carbonate (Li<sub>2</sub>CO<sub>3</sub>), co-produce sodium carbonate (Na<sub>2</sub>CO<sub>3</sub>), and achieve permanent CO<sub>2</sub> storage through mineralization. The strategy integrates five electrochemical steps—employing an anion exchange membrane—with one chemical CO<sub>2</sub> absorption step. Electrolysis initially raises brine pH to eliminate divalent cations without chemical additives. Subsequent CO<sub>2</sub> sparging in alkaline brines induces Li<sub>2</sub>CO<sub>3</sub> and later Na<sub>2</sub>CO<sub>3</sub> precipitation. Results demonstrate 78% Li<sup>+</sup> recovery as impure Li<sub>2</sub>CO<sub>3</sub> and 71.8% Na<sup>+</sup> recovery as Na<sub>2</sub>CO<sub>3</sub>. Notably, 205.8 g of CO<sub>2</sub> per litre of initial brine were absorbed, of which 189.7 g were permanently stored in solid carbonates. The process minimizes chemical inputs, reduces reliance on remote chemical delivery, and leverages high brine salinity to enhance CO<sub>2</sub> capture kinetics and electrochemical efficiency. Though not optimized for energy consumption, this proof-of-concept study reveals a circular approach to lithium extraction, integrating critical material recovery with climaterelevant carbon capture. Future improvements could enable direct air capture integration and Li<sub>2</sub>CO<sub>3</sub> purification. This study introduces an industrially relevant pathway to reduce the environmental impact of lithium brine mining by turning waste brine constituents into valuable, stable products while closing the carbon cycle through mineralization.

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#### 1. Introduction

Carbon capture, utilization and storage has a fundamental role in achieving several of the United Nations Sustainable Development Goals.<sup>1</sup> Electrochemical CO<sub>2</sub> capture technologies are attracting increasing attention due to their flexibility, and their ability to address decentralized emissions.<sup>2,3</sup> Moreover, they are particularly attractive in locations with high potential for renewable energy generation.<sup>2,4</sup> Most commonly, these CO<sub>2</sub> capture methodologies are based on the trivial capability of electrochemical methods to swing the pH of the CO<sub>2</sub> absorbing solution. The hydration and dehydration equilibrium is used firstly to capture CO<sub>2</sub>, and potentially release it more concentrated at a second stage, if required, by re-acidification. Electrochemically the pH of the solution can be adjusted without addition of chemicals<sup>2,5</sup> by electrolysis,<sup>6</sup> bipolar membrane electrodialysis,<sup>7-9</sup> reversible redox reactions<sup>10–12</sup> and capacitive deionization.<sup>13</sup> A non-negligible advantage of electrochemical methods is their ability to couple CO<sub>2</sub> capture and utilization. The main challenge of any CO<sub>2</sub> capture technology is high energy consumption, coupled with the still reduced implementation of tax or legislation incentives for the private sector to voluntary implement such initiatives.<sup>14,15</sup>

The absorption of CO<sub>2</sub> in aqueous solutions and its conversion to bicarbonate anions is favoured thermodynamically. However, at neutral pH this conversion rate is kinetically slow. <sup>16–18</sup> The kinetics can be considerably accelerated by increasing the pH, the ionic strength of the solution or the temperature. Increasing temperature will however diminish the total amount of inorganic carbon that will be solubilized in aqueous solution. Conversely, increasing the pH will not only increase the kinetics of absorption and conversion, but will also radically augment the amount of dissolved total inorganic carbon by over 2 orders of magnitude per pH unit. <sup>2,19</sup> Finally, high salinity brines increase the kinetics of absorption and conversion, are naturally abundant, and aid the implementation of electrochemical technologies, by lowering costs when reducing the electrical resistance of a given media. H<sub>2</sub> production during water electrolysis can partially counterplay the cost of CO<sub>2</sub> capture. <sup>2</sup>

CO<sub>2</sub> capture is only the beginning of the pathway towards sustainability. In order to conversion of carbon cycle, CO<sub>2</sub> needs to be alternatively utilized or permanently stored. Chemical conversion of CO<sub>2</sub> to products such as alcohols, formic acid or different organics is possible via different routes.<sup>7,20,21</sup> The other option is to permanently store CO<sub>2</sub>. Gaseous CO<sub>2</sub> can be stored at high pressure in deep geological layers<sup>22</sup> or alternatively by mineralization as solid carbonates.<sup>23–26</sup> The latter is subjected to higher social acceptance as compared to the former, and the storage capacity is potentially unlimited, since it is less restricted by geological conditions in deep layers.<sup>25</sup>

Today, lithium mining from natural brines is performed through a methodology known as the evaporitic technology.<sup>27–29</sup> Briefly, brines are pumped from underground deposits and poured into large, water-proof linen covered, open-air evaporation ponds where they are let to reside for as long as required until a suitable lithium cations concentration is reached for further processing. This process takes place exclusively in extremely arid regions, such as the Lithium Triangle in South America, the Nevada Dessert or southwestern China, where the weather is sufficiently dry. In addition, large extensions of flat inexpensive land are required to build ponds. After in between 10-24 months, a large proportion of the more concentrated salts sequentially crystallize in the ponds. At the moment lithium concentration is high enough for successful crystallization of lithium carbonate, sodium chloride concentration is close to saturation in the concentrated brine.<sup>29</sup>

In the current evaporitic technology, the concentrated brine is firstly purified to remove borates, magnesium and calcium. Boron is most commonly removed with liquid-liquid extraction in a mixture of kerosene and different alcohols.<sup>29</sup> Magnesium is removed by increasing the pH via addition of hydrated lime, while calcium is removed via addition of small amounts of soda ash (sodium carbonate).<sup>27,29</sup> In one facility, these three impurities are removed instead by re-dissolving the primary lithium carbonate via carbonation, and treating the LiHCO<sub>3</sub> solution in ion exchange resins that remove said impurities.<sup>30</sup> Lithium carbonate is crystallize via addition of soda ash at a temperature close to 90 °C, to take advantage of its inverse solubility.<sup>27,31</sup> Overall, while most concerns about current lithium

mining from brines is focused on water issues,<sup>27,32</sup> the large consumption of chemicals in the articles processing stage is not to be ignored. In addition to costs and the environmental footprint of the large consumption of chemicals, there are logistical issues associated with the required constant delivery of said chemicals. Lithium brine mining facilities are exclusively in very remote desertic locations, often on gravel winding roads up the mountains. Railway infrastructure is often not in place in those remotes areas, and lorry transportation of chemicals is sometimes disrupted due to extreme weather conditions, with non-negligible economic impacts.<sup>28</sup>

Here we report on a new integrated and intensified chemical process at proof-of-concept level to simultaneously crystallize lithium carbonate, Li<sub>2</sub>CO<sub>3</sub>, with the co-production of large amounts of Na<sub>2</sub>CO<sub>3</sub>, capture CO<sub>2</sub> and permanently store it in said mineralized products. A 6-step processing strategy was developed comprising 5 electrolytic and 1 chemical processing step. The electrolytic steps all comprise a water electrolyser fitted with an anion exchange membrane. Brine is always fed to the cathodic compartment, which is coupled in 4 of 5 electrolytic steps to a side-decanter where CO<sub>2</sub> is continuously sparged. A qualitative-quantitative analysis of all solids and intermediate brines was carried out. The first treatment was successful in the full abatement of divalent cations. The second and third treatment successfully crystallized 78 % of the original Li<sup>+</sup> content as primary lithium carbonate (70 % purity). The fourth, fifth and sixth step achieved the crystallization of 71.8 % of the original Na+ content as sodium carbonate. Despite the large depleting of salts, the effluent brine was only reduced 10 % in total dissolved solids content, because the original brine volume was also reduced to 40 % of its original volume and the remaining salts were then contained in a much more reduced solution volume. In the overall technology proposal a total of 205.8  $g_{CO_2}L_{brine}^{-1}$  were captured in both brine and solids, 189.7  $g_{CO_2}L_{brine}^{-1}$  of which are considered to have been also permanently stored in highly stable mineralized products: Li<sub>2</sub>CO<sub>3</sub> and Na<sub>2</sub>CO<sub>3</sub>.

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#### 2. Materials and methods

# 2.1. Experimental setup

A schematic representation of the 6 step brine treatment proposed here is shown in Figure 1. This is comprised of 5 electrochemical steps, and one pure chemical step (step 4). The first brine processing step, named ELECTROLYSIS E0, was an electrochemical process, carried out to eliminate divalent cations (see Figure 1, flux diagram in left, coloured in black). The electrolysis was carried out during 2 h at 10 mA cm<sup>-2</sup>, according to the initial composition of the brine, for a successful elimination of divalent cations. The second and third brine processing steps, named ELECTROLYSIS E1.1, and E1.2, were accomplished to alkalinize the brine, to favour absorption of CO<sub>2</sub>, and trigger the crystallization of lithium as lithium carbonate (Figure 1, flux diagram on the left, coloured in blue). The electrolysis was carried out at alternatively 300 and 350 A cm<sup>-2</sup>. ELECTROLYSIS E1.1 was carried out until the large amount of solids accumulated in the reactor made solution recirculation difficult. ELECTROLYSIS E1.2 was carried out until the amount of solid did not seem to increase any further (see Figure S1 in Supporting information).

The fourth brine processing step, named CHEMICAL C2, is a chemical absorption step of CO<sub>2</sub>. Technical grade CO<sub>2</sub> was sparged in the brine in an open vessel for 14.0 minutes at a flow rate of 176 mL min<sup>-1</sup>. This step, triggered the crystallization of sodium carbonate (Figure 1, flux diagram on the left, in purple). Finally, the fifth and sixth brine processing steps were methodologically similar to the second and third processing steps, although sodium carbonate was the main compound crystallized.

At the end of each of the 6 steps, the suspension in the cathodic compartment and connected crystallization vessel was filtered and the solids were dried at 100 °C for chemical analysis. An aliquot of the obtained supernatant was separated for analysis, and the remaining brine was further processed as indicated in Figure 1. Experimental conditions for all 6 steps are detailed in Table 1.

**Table 1.** Detail experimental conditions for the six process steps.

<u>Table 1.</u> Detail experimental conditions for the six process steps.  View Article On DOI: 10.1030/DSSL1005						View Article Onlir	
Step	Total circulated charge (C)	Average current applied (A)	Maximum current applied (A)	Experiment duration (h)	Mass of CO <sub>2</sub> bubbled (Kg)*	Initial brine Volume (L)	Final brine Volume (L)
E0	26910	2.10	4.26	3.55	0	2.00	2.00
E1.1	274095	2.99	3.51	25.48	0.448	2.00	1.75
E1.2	266730	3.96	4.00	18.72	0.330	1.75	1.55
<b>C2</b>	0	0.00	0.00	8.33	0.147	1.55	1.40
E3.1	136714	2.02	2.02	18.80	0.331	1.40	1.00
E3.2	126048	2.02	2.02	17.33	0.305	1.00	0.80

<sup>\*</sup> Calculated at 20°C and 0.92 bar.

### 2.1.1. Electrochemical setup

A two-compartment electrochemical reactor manufactured in-house with similar specifications as described in Desloover et al. was used in all electrochemical experiments.<sup>33</sup> Two acrylic frames with internal dimensions of  $5 \times 20 \times 2$  cm, separated by an anion exchange membrane (AEM, model AMI-7001CR, Membrane International Inc., USA), were assembled together and fixed, yielding an internal volume of 200 mL for both anodic and cathodic compartments. The electrodes were separated about 2 mm with a plastic mesh to avoid direct contact with the membrane. The anode was a titanium mesh covered with a mixture of iridium and titanium oxides, (IrO<sub>2</sub>/TiO<sub>2</sub>; 65/35 %; dimensions 4.8 cm × 19.8 cm × 1 mm, Magneto Special Anodes), and the cathode was a 316 stainless-steel mesh. Experiments were run at a constant current by using a UNI-T Instrument DC/DC regulated power supply. Figure S2 shows a picture of this setup. Properties and characteristics of anionic membrane are listed in Table S1. Membranes were previously immersed in NaCl 5% overnight to allow their hydration and expansion. Before usage, they were washed with ultra-pure water and kept in NaCl 5% solution when not in use.

The anodic compartment was connected to a 2.0 L plastic drum, whilst the cathodic compartment was connected to a 2.0 L Schott bottle serving as a decanter. Initially, the volume of brine treated was 2.00 L (for Electrolysis E0 and E1.1). For Electrolysis E1.2, Chemical C2, Electrolysis E3.1, and E3.2, initial volumes were 1.75 L, 1.55 L, 1.40 L, and 1.00 L respectively. Final

volume of brine B4 of the overall process was 0.80 L. Brines and anolyte were continuously recirculated in the electrochemical steps with the aid of a peristaltic pump (6 L/h, PC28 APEMA, Argentina) with the aim of forcing mass transport. For the chemical step, also bubbling facilitates the stirring phenomenon. To avoid chlorine production in the anodic compartment, a carbonate-bicarbonate buffer was used as anolyte (2 M, pH 10, industrial grade).<sup>34</sup>

# 2.2. Initial brine

The departing brine sample, termed Brine B0, was a generous sample from a lithium mining plant of in the Northwest of Argentina, de-identified site. The composition of this brine is listed in Table 2. This is not a native brine sample. The native brine contained a much reduced Li<sup>+</sup> content, in the order of 600-900 ppm Li<sup>+</sup>. Brine B0 was obtained after the native brine was left to rest for approximately a year in a series of solar evaporation ponds, which allowed for the loss of water from the original brine by natural evaporation, producing an increase in Li<sup>+</sup> concentration, and the crystallization of a portion of the most concentrated salts, such as NaCl and KCl. Hydrated lime was added at some point before collection to eliminate a large portion of Mg<sup>2+</sup> from the concentrated brine (which explains the high pH of Brine B0). This type of concentrated brine is representative of those generated in various lithium mining operations in countries such as Chile, Argentina, the United States, and China.

**Table 2**. Composition of the departing brine sample.

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Brine	$\mathbf{Li}^{\scriptscriptstyle +}$	Na <sup>+</sup>	K <sup>+</sup>	Ca <sup>2+</sup>	Mg <sup>2+</sup>	Cŀ	SO <sub>4</sub> -2	В	CO <sub>3</sub> -2
	g/L								
R0	$6.0 \pm 0.3$	$112 \pm 4$	$59 \pm 2$	$0.043 \pm 0.003$	$0.069 \pm 0.002$	$190 \pm 7$	$0.55 \pm 0.07$	$3.47 \pm 0.15$	$16.8 \pm 0.5$

# 2.3. Analysis

X-ray diffraction (XRD) patterns were measured to characterize chemical structures. An X-ray powder Rigaku Miniflex diffractometer (45KV and 30 mA) with a copper anode and curved graphite monochromator was employed. XRD patterns were recorded over a 2θ range from 4 to 90° with a step

of 0.02°. Results were compared with different lithium reference patterns to assign the main chemical structure and impurities.

Scanning electron micrographs were determined to evaluate particle size, morphology, and surface appearance at 3 kV acceleration voltage, working distance between 9 - 10 mm, with 40 nm of gold covering, and using a secondary electron detector (SEM, Carl Zeiss, Jena, Germany, EVO MA10 Model, equipped with a W filament).

Measurements to assess the chemical composition of brines and solids were conducted using an Inductive coupled plasma-optical emission spectrometer (ICP-OES) 5800 by Agilent. Each analysis was repeated three times, and informed values correspond to the mean value. Solids were dissolved in 5 % HNO<sub>3</sub> to assess composition. Chloride and carbonates anions were determined by volumetric titration.

Conductivity was determined in a dilution sample of 1:500 of the solution of interest with a conductivity probe (HI763063 probe and HI99301 controller, HANNA). Brine density was determined by measuring ten times the weight of 1.000 mL of sample and averaging the results. The pH was determined by acid-basic titration. 0.10 mL samples were taken to measure total dissolved solids (TDS).

# 2.4. Calculations

Volume changes in Figure 3 refer to the change in one single processing step (e.g. not vs. the initial brine volume entering Electrolysis E0).

The amount of sequestered CO<sub>2</sub> in these experiments was calculated assuming that the difference in Li<sup>+</sup> and Na<sup>+</sup> content between the different brines and B0 corresponds to crystallized Li<sub>2</sub>CO<sub>3</sub> and Na<sub>2</sub>CO<sub>3</sub>, respectively. K<sub>2</sub>CO<sub>3</sub> was not considered since it was not observed in the diffractograms (see below). The missing amounts of Li<sup>+</sup> and Na<sup>+</sup> were converted to moles. The amounts of the corresponding carbonates (in moles) is half the amount of moles of the corresponding cations. Finally, the mass of carbonate was converted to CO<sub>2</sub> (factor 60/44) and divided by 2 so that

the final result would correspond to 1 L of initial brine fed to the first electrolytics of the first electrolytics (ELECTROLYSIS E0).

CO<sub>2</sub> emissions considering different energy sources were calculated using data from Life Cycle Impact Assessment (LCIA) of Electricity Generation Options<sup>35</sup> summarized in table 3:

Table 3. LCIA results for climate change indicator for selected energy sources

Energy source	CO <sub>2</sub> equivalent emissions per kWh /
	g CO <sub>2</sub> equivalent
Natural gas (without CO <sub>2</sub> capture)	486
Oil (without CO <sub>2</sub> capture)	840
Concentrated solar power (tower and trough)	28
Photovoltaics (thin film and crystallized Si)	43
Wind	13

#### 3. Results and discussion

# 3.1. Proposed brine processing

# 3.1.1. First set of electrolytic steps

The first processing step (B0 is the initial composition and B1.1 is the final composition) was devised to increase the brine pH with the aim of producing the abatement of the small amount of divalent cations present in the as-received brine (see Figure 1, top left). Electrolysis E0 effectively produces the full abatement of Mg<sup>2+</sup> and Ca<sup>2+</sup>. After filtering the obtained suspension, both Mg<sup>2+</sup> and Ca<sup>2+</sup> were below ICP detection limit in the supernatant. Cations and anions composition, together with physical-chemical parameters of brines after the different processing steps are shown in Figures 2 and 3. The other brine components remain almost constant after Electrolysis E0. The slight increase in concentration is explained by water splitting at the electrode surface.

In the second processing step (B1.1 is the initial composition and B2.1 is the final composition), water electrolysis is carried out with concomitant CO<sub>2</sub> sparging to favour CO<sub>2</sub> absorption. The sequestration of CO<sub>2</sub> in alkaline solutions is favoured due to the equilibria system depending on the pH of the solution. At high pH, the equilibrium is displaced to carbonate formation (Eq. 1 to 3). <sup>16–18</sup> If lithium and/or sodium concentrations are high enough, precipitation of the corresponding carbonates

e.g. absorption of larger absolute amounts of CO<sub>2</sub>. <sup>19,36,37</sup> At intermediate pH values, bicarbonate is the main C species, and provided that Na<sup>+</sup> concentration is high enough NaHCO<sub>3</sub> can be obtained (Eq. 6). <sup>9,37</sup>

$$2H_2O + 2e^- \rightleftharpoons 2OH^- + H_2$$
 (Eq. 1)

$$OH^- + CO_2 \rightleftharpoons HCO_3^-$$
 (Eq. 2)

$$OH^- + HCO_3^- \rightleftharpoons CO_3^{2-} + H_2O$$
 (Eq. 3)

$$2Li^{+} + CO_3^{2-} \rightleftharpoons Li_2CO_{3\downarrow}$$
 (Eq. 4)

$$2Na^{+} + CO_3^{2-} \rightleftharpoons Na_2CO_{3\downarrow}$$
 (Eq. 5)

$$HCO_3^- + Na^+ \rightleftharpoons NaHCO_{3\downarrow}$$
 (Eq. 6)

Figure 2 indicates that Electrolysis E1 produces a marked decrease in the C/C<sub>0</sub> ratio for Li<sup>+</sup>, falling to 0.321 times its initial value. This behaviour indicates that Electrolysis E1.1 effectively operates as a lithium extraction process, in which Li<sup>+</sup> removal occurs within the decanter attached to the cathodic compartment. Water electrolysis generates OH<sup>-</sup>, raising the pH above 13-14 and allowing the formation of carbonate anions from dissolved CO<sub>2</sub>. These react *in situ* with Li<sup>+</sup> to form lithium carbonate (Li<sub>2</sub>CO<sub>3</sub>). The anionic membrane (AEM in Figure 1) serves to separate the anolyte and catholyte, avoiding H<sup>+</sup> from the anode to reach the cathode and allowing the control of the pH. The AEM is not selective to different anions, and chloride, sulphates, borates and hydroxyl species can all migrate to maintain electroneutrality in both compartments. Figure 3C suggests that chloride species are the main charge carriers. In contrast, during Electrolysis E1, Na<sup>+</sup> and K<sup>+</sup> cations maintain C/C<sub>0</sub> values close to 1, indicating that they mostly remain in solution (see discussion on solid compositions below). After filtration of solid S1.1., Electrolysis E1.2 was started following the same procedure as before. A new solid, S1.2. was obtained. Figure 2 shows that the Li<sup>+</sup> content can be further decrease, although at this step, only 10% of the original Li<sup>+</sup> content is removed from the brine (vs. 67% during Electrolysis E1.1.).

Figure 4 shows the diffractograms of all solids obtained throughout processing. The Model of MaCl and KCl. While no other diffraction peaks are observed, elemental analysis also show the presence of Li<sup>+</sup>, Mg<sup>2+</sup> and Ca<sup>2+</sup>, determined at 0.0114  $\pm$  0.0007, 0.024  $\pm$  0.001, 0.048  $\pm$  0.003 g per gram of solid, respectively, *e.g.* with considerably lower compositions as compared to both Na<sup>+</sup> and K<sup>+</sup> (see figure 5). The abatement of Mg<sup>2+</sup> and Ca<sup>2+</sup> cations is associated to the *in situ* production of hydroxyl anions producing the precipitation of Mg(OH)<sub>2</sub>, Ca(OH)<sub>2</sub> and CaCO<sub>3</sub> following Eq. 7 to 9.

$$Mg^{2+} + 2OH^{-} \rightarrow Mg(OH)_{2} \downarrow$$
 (Eq. 7)

$$Ca^{2+} + 2OH^{-} \rightarrow Ca(OH)_{2} \downarrow$$
 (Eq. 8)

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$$Ca^{2+} + CO_3^{2-} \rightarrow CaCO_3 \downarrow$$
 (Eq. 9)

Mg(OH)<sub>2</sub> is well known for precipitating with very small particles,<sup>38,39</sup> below 500 nm average diameter, and thus adsorbing other ions and entrapping relatively large brine amounts, which upon drying, explain NaCl and KCl diffraction peaks in S0.

The X-ray diffractogram of S1.1 shows distinctive peaks identified as Li<sub>2</sub>CO<sub>3</sub>, with peaks of lower intensity assigned to NaCl and KCl. The corresponding diffractogram of S1.2 also shows peaks for Li<sub>2</sub>CO<sub>3</sub>, with lower intensity peaks assigned to NaCl and Na<sub>2</sub>CO<sub>3</sub>. The solid purity values were calculated as 71 % and 36 %, respectively in Li<sub>2</sub>CO<sub>3</sub> for S1.1 and S1.2. The quantitative solid analysis shown in Figure 5 is in good agreement with the X-ray diffractograms. Solid purification was beyond the scope of this work. Impurities identified just above are highly water soluble (NaCl, KCl and Na<sub>2</sub>CO<sub>3</sub>). Li<sub>2</sub>CO<sub>3</sub> shows an inverse solubility with temperature,<sup>31</sup> hence solid washing with pure water at approximately 95°C is suggested as the best alternative to purify both S.1.1 and S1.2., while minimizing Li<sub>2</sub>CO<sub>3</sub> loss. Counter current washing should minimize freshwater requirements for solid washing.

This proposal replaces the cost and the logistical hurdles of shipping large amounts of chemical reagents to remote mining locations, while CO<sub>2</sub> is captured and permanently stored (see Section 3.5).

The percentage of Li<sup>+</sup> recovery from the brine originally fed to the system is similar to current practice  $^{\text{line}}$  via crystallization with soda ash, the amount of Li<sup>+</sup> remaining in brine is due to the relatively large solubility of Li<sub>2</sub>CO<sub>3</sub>,  $^{29}$  at 13.0  $g_{kg_{H_2O}}$  at 25°C.

# 3.1.2. Chemical processing

In the fourth processing step (departing from B2 to reach B3.1) the system passes from an electrochemical to a chemical step where sodium is partially removed from the brine. During the chemical treatment, a decrease in both Na<sup>+</sup> and K<sup>+</sup> content in the brine (Figure 2A) is observed, accompanied by a plateau in Cl<sup>-</sup> levels and a drop in  $CO_3^{2-}$ . This behaviour is interpreted as a consequence of CO<sub>2</sub> absorption in the brine, which, under alkaline conditions, favours the formation of carbonate species (Eq. 2 and 3) that react with Na<sup>+</sup> to precipitate as Na<sub>2</sub>CO<sub>3</sub> (Eq. 5). The maximum  $CO_3^{2-}$  is reached at the end of the third electrolytic step. The subsequent chemical step produces a drop in  $CO_3^{2-}$  concentration that agrees with the decrease in Na<sup>+</sup> concentration, suggesting effective recovery of Na<sup>+</sup> in the form of solid salt. The crystallization of Na<sub>2</sub>CO<sub>3</sub> seems to take place only when Li<sup>+</sup> concentration has been largely depleted, in agreement with the much higher solubility value of Na<sub>2</sub>CO<sub>3</sub>, as compared to Li<sub>2</sub>CO (2.9 vs. 0.18  $mol kg_{H_2O}$ , respectively, in pure water at 25 °C).<sup>31</sup> Qualitatively, the kinetics of crystallization of Li<sub>2</sub>CO<sub>3</sub> seem to be faster than those of Na<sub>2</sub>CO<sub>3</sub>.

The collected solid S2 shows sharp peaks of Na<sub>2</sub>CO<sub>3</sub> according to XRD analysis, with concurrent presence, albeit at lower intensity, of Li<sub>2</sub>CO<sub>3</sub>, KCl and NaCl. The XRD data is again in good agreement with the composition analysis which indicates that Na<sup>+</sup> and CO<sub>3</sub><sup>2-</sup> are the major species, with a calculated solid purity in Na<sub>2</sub>CO<sub>3</sub> of 68% (previous to any purification step). A pH of 9.5 had been reached at the moment when solid S2 was filtered (Figure 3B). Knowing that continuing with CO<sub>2</sub> sparging would result in a further decrease in pH and that this would favour the presence of bicarbonate anions over carbonate (Eq. 2) the decision to further increase the pH electrochemically was taken.

At 1.23  $mol\ kg_{H_2O}$  vs. 2.9  $mol\ kg_{H_2O}$ , the solubility in pure water of NaHCO<sub>3</sub> is slightly fower than that of Na<sub>2</sub>CO<sub>3</sub>. Targeting the crystallization of NaHCO<sub>3</sub> instead of Na<sub>2</sub>CO<sub>3</sub> is yet a plausible alternative to be tested. From the perspective of CO<sub>2</sub> capture and storage, NaHCO<sub>3</sub> can store twice the amount of CO<sub>2</sub> as compared to Na<sub>2</sub>CO<sub>3</sub>. However, from the perspective of brine desalination, the crystallization of Na<sub>2</sub>CO<sub>3</sub> deprives the brine from twice the amount of Na<sup>+</sup>, as compared to NaHCO<sub>3</sub>.

### 3.1.3. Second set of electrolysis

Electrolysis E3.1. was started to further rise brine pH and increase CO<sub>2</sub> fixation. For the second time during processing a large proportion of Na<sup>+</sup> content in the brine is crystallized (B3.1 to B3.2), evidenced by a drop in both Na<sup>+</sup> and K<sup>+</sup> contents, with content drops of 22.1 and 13.1% of their original content, and a minor drop in Li<sup>+</sup> concentration of 4.4% of its original content (Figure 2A). The X-ray diffractogram shows that solid S3.1 is mostly composed of Na<sub>2</sub>CO<sub>3</sub>, with lower intensity peaks corresponding to Li<sub>2</sub>CO<sub>3</sub>, KCl and NaCl, in agreement with data in Figure 5. The continuous observed precipitation of Li<sub>2</sub>CO<sub>3</sub> is attributed to the brine volume decrease which allows for Li<sub>2</sub>CO<sub>3</sub> to reach saturation again triggering its crystallization. Conversely, the presence of both KCl and NaCl is attributed to ion adsorption in the large amount of crystallized solid, since both species are still far from their solubility values. In this proof-of-concept experimental setup, Electrolysis E3.1 had to be stopped because of the accumulation of considerable solid amounts within the electrochemical reactor (no in-line filtering was used).

After filtration of the suspension, a further electrolytic treatment coupled to CO<sub>2</sub> sparging was started, following exactly the same procedure, this final treatment was named Electrolysis E3.2. Both Na<sup>+</sup> and K<sup>+</sup> content continue to decrease, now with content drops of 20.9, 11.5, and 1.2% of the respective original content of Na<sup>+</sup>, K<sup>+</sup> and Li<sup>+</sup> (Figure 2A). The X-ray diffraction pattern of solid S3.2. (Figure 5G) is quite similar to that of the solid obtained during the previous step, Electrolysis E3.1., except that no peaks are observed for Li<sub>2</sub>CO<sub>3</sub>, while those of NaCl are of much lower intensity as in solid S3.1. The elemental content of solid S3.2 (Figure 4) is also in good agreement with its diffraction

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pattern. Overall, ELECTROLYSIS E3.1 and E3.2 can be considered the same processing attended to brine composition and physico-chemical parameters follow the same trends, while the two obtained solids are very similar. In a hypothetical larger scale application with on-line filtering of solids, there would be no need to interrupt the processing, as performed here.

Overall, after the 6-step processing strategy, the net diminution in these cations in brines were 1.514, 6.994, and 1.882 moles, corresponding to 87.6%, 71.9% and 63.4% of the initial Li<sup>+</sup>, Na<sup>+</sup> and K<sup>+</sup> content, respectively, in addition to 100% removal of Mg<sup>2+</sup> and Ca<sup>2+</sup>. These changes in cation content come along with the production of Li<sub>2</sub>CO<sub>3</sub> and Na<sub>2</sub>CO<sub>3</sub>, that is why we considered the proposed strategy as a circular economy approach, since it produces two different products, minimizes liquid waste, in addition to capturing and storing CO<sub>2</sub>.

# 3.2. Further brine analysis

Because of the extremely high salinity, density values for all these brines are above 1.25 g cm<sup>3</sup>. Brine conductivity was also very high and therefore only the conductivity values of 1:500 dilutions could be measured. Brine density, conductivity, pH, volume changes, TDS, chloride and carbonate content are depicted in Figure 3. Brine density remains unchanged in the short ELECTROLYSIS 0 step, while its conductivity increases. Both brine density and conductivity increase during the much longer ELECTROLYSIS E1.1 and 1.2. The increase in conductivity is undoubtedly explained by the increase in pH, due to the in-situ formation of hydroxyl anions, which molar conductivity is known to be higher than that of the chloride anions that are replaced during electrolysis.<sup>40</sup> The maximum value reached for brine density in B2 is correlated to the concentration of Na<sup>+</sup> in brine, in agreement with its behaviour showed in Figure 2. While Na<sup>+</sup> content is almost the same as that in the starting brine, the concentration increases due to the brine volume decrease (explained below). In the chemical step (B2 to B3.1) a marked drop of brine density is observed, explained by the sharp decrease in pH. Following the same reasoning as above, the decrease in brine density is associated to the large precipitation of sodium salts. In the last two electrochemical step (B3.1 to B4), the conductivity and density of brine

decreases due to the precipitation of large salt amounts, mostly sodium carbonate, as shown if Figure 12.

It is observed that the brine volume is reduced during ELECTROLYSIS E1.1., 1.2, 3.1 and 3.2. This is explained to a large extent by electro-osmotic effects (anions are mostly transferred across the ionic exchange membrane in a hydrated form),<sup>34</sup> by water electrolysis at the cathode (about 250 g H<sub>2</sub>O in total), and likely due to poor filtration of large amounts of solids. The latter could be reduced at laboratory scale with vacuum filtration and/or centrifugation, although it was preferred to stick to a simple filtration steps to resemble more closely a potential industrial application. The duration of the electrolytic steps was far from being optimized. Electrolysis and CO<sub>2</sub> absorption were let to run while the amount of crystallized solids was observed to increase, although this was only a qualitative appreciation. Shorter electrolytic treatments would have resulted in lesser volume changes. Indeed, only in the short ELECTROLYSIS 0, is the change in the volume negligible.

Anions' concentrations are shown in Fig 3C. Chloride composition falls in all electrolytic steps due to the migration of these anions from the catholyte compartment to the anolyte through the anion exchange membrane. While chloride concentration falls only to half of the initial concentration, since brine volume was reduced by over half of the initial value, the total content of chlorides was reduced in 8.44 moles Cl<sup>-</sup> migrated, corresponding to 78.7% from the total initial chloride content in B0. Carbonate anions' concentration increases during ELECTROLYSIS 0, 1.1. and 1.2. While no CO<sub>2</sub> was sparged during ELECTROLYSIS 0, the increase in carbonate concentration is explained by direct absorption from CO<sub>2</sub> in air (the decanter was open to the atmosphere). This is interesting because it opens up the possibility to studying direct CO<sub>2</sub> capture from air, although this is beyond the scope of this work. However, this is most interesting possibility both in terms of reducing operation costs and sustainability. For the successive steps, with large Li<sup>+</sup> and Na<sup>+</sup> concentrations to be crystallized, direct capture from air would entail longer residence times, since the kinetics of CO<sub>2</sub> absorption would be slower considering the reduced CO<sub>2</sub> concentration in air (at about 400 ppm, more reduced at high

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altitude due to lower atmospheric pressure). The energy consumption would also be reduced atthough a consumption would also be reduced at the consumption would at the consumption would be reduced at the consumption would b considering 5 electrolytic steps are part of the overall process, the reduction in energy consumption could be rather small considering that electrolytic steps often take the largest share of energy consumption in an overall technology (see Section 3.4).<sup>40</sup>

The increase in carbonate concentration was even faster when CO<sub>2</sub> was sparged (it should also be considered that part of the captured CO<sub>2</sub> was removed as the solids were separated from the suspension). Carbonate decreases sharply after the chemical step, which is easily explained by the decrease in pH to about 9.0 shifting the equilibrium to bicarbonate species (Eq. 2). Carbonate concentration never reached again the maximum value, despite continuous electrolysis and sparging since during ELECTROLYSIS E3.1 and 3.2 was kept at lower values than before (never higher than 11, as compared to higher than 13 in ELECTROLYSIS E1.1 and 1.2).

Figure 3C shows that the TDS value remains at extreme salinity values close to 400 g L<sup>-1</sup> during all processing, falling at 370 g L<sup>-1</sup> by the end of the brine processing. At first glance, this might seem a very poor performance in terms of brine desalination. A closer look indicates that the initial brine volume was reduced by 60% (Figure 3B), e.g. while the TDS value was only reduced in about 10%, the amount of salts that were not removed from the initial brine is now present in a much reduced brine volume. In turn, it should also be noted that hydroxyl anions, present in brine B0 at close to 0.5 M decrease in concentration to about 10<sup>-4</sup> M after the first three electrolytic steps. Hydroxyl anions are mostly replaced by carbonate anions to balance charge, and these add a higher value to the TDS (34 g for 2 moles of hydroxyl vs. 60 g for 1 mol of carbonate, accounting for the same amount of negative charges).

The water contained in the initial brine is not irreversibly lost entirely. H<sub>2</sub> and O<sub>2</sub> produced at the electrodes should certainly be captured, particularly the former, and re-used as fuel, which would regenerate water. Water migration across the anion exchange membrane could be reduced firstly by reducing the duration of the electrolysis, of utmost importance also in terms of energy consumption.

In the case of S2, Na<sub>2</sub>CO<sub>3</sub> is the key product of the chemical step. In the same way, it was observed a non-negligible coprecipitation on Li<sub>2</sub>CO<sub>3</sub> for S2 and adsorption of brine in as shown with the peak for NaCl at 20 = 45°. The attenuation of the Li<sub>2</sub>CO<sub>3</sub> peaks in these samples reinforces the idea that the system allows for sequential and selective recovery of carbonates, first precipitating Li<sub>2</sub>CO<sub>3</sub> and then Na<sub>2</sub>CO<sub>3</sub> under controlled conditions. These results suggest that, even without purification steps, it is possible to guide the process toward obtaining differentiated products of interest, which constitutes a significant advantage for the development of direct lithium extraction (DLE) technologies with stepwise cation separation. For the solids obtained from the last electrochemical step, S3.1 and S3.2, they are similar to S2. It is important to note that none of the solids were washed after recovery, implying that the crystallographic profiles include both precipitation products and species possibly adsorbed from the residual brine.

# 3.3. Scanning electron microscopy analysis

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In Figure 6, scanning electron micrographs show the shape and size of the obtained solids. For comparison, the micrographs for analytical grade commercial samples of Li<sub>2</sub>CO<sub>3</sub> and Na<sub>2</sub>CO<sub>3</sub> are also shown. Figure S3 in Supporting Information shows the micrographs of analytical grade commercial samples of NaHCO<sub>3</sub>, NaCl and KCl. Solids S1.1 and S1.2 were previously identified via ICP-OES and X-ray diffraction analysis as composed mostly of Li<sub>2</sub>CO<sub>3</sub>. The latter shows crystals with homogeneous prismatic shapes, while the former also depict prismatic shapes. These shapes are related to the commercial sample, however, they are much thinner than the commercial samples. The crystals obtained here are also of a more reduced size, in agreement with much lower purity as compared to an

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None of the three other solids which were identified as being composed mostly of Na<sub>2</sub>CO<sub>3</sub> bare a close resemblance to the Na<sub>2</sub>CO<sub>3</sub> commercial sample analysed here. The latter consist of agglomerated small crystals, a mixture of prismatic and platelet shapes. In turn, solids S2 and S3.1 both are mostly prismatic. Surprisingly the size of the crystals obtained for S2 and S3.1 are considerably larger than those of commercial Na<sub>2</sub>CO<sub>3</sub>, with the sample obtained during the ELECTROLYSIS E3.1 being even larger than that obtained during CHEMICAL C2. It could actually be argued that these crystals have a mixed shape between Na<sub>2</sub>CO<sub>3</sub> and KCl, which is in agreement with the chemical composition (Figure 4). Finally, the last solid, S3.2 show the smaller size crystals, not resembling much any of the pure samples analysed here.

# 3.4. Electrolytic energy consumption

Considering that electrolysis is an energy intensive process, the energy consumption was calculated only for the electrolytic steps. Fluid pumping and  $CO_2$  sparging were not considered in the energy calculation. The electrolytic energy consumption,  $U_{EC}$ , was calculated as:

$$U_{EC} = \int_0^t E \cdot i \cdot dt \tag{Eq. 10}$$

where E is the voltage, i the applied current, and t the duration of a given electrolytic step.

Figure 7A presents the energy consumption, expressed in Wh L<sup>-1</sup> of treated brine. Electrolysis E0 requires the lowest energy consumption due to a low composition in divalent cations. Electrolysis E1.1, E1.2 show the highest energy consumption above 200 Wh L<sup>-1</sup>, which reflects the intensity of electrolysis in these steps. In turn, Electrolysis 3.1 and 3.2 show a more moderate energy consumption (below 150 Wh L<sup>-1</sup>), which suggests an improvement in the CO<sub>2</sub> fixation process efficiency. The chemical step C2, without applied current, showed zero energy consumption, as expected. This proposal was studied here at proof-of-concept level, and the energy consumption was not optimized.

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It is believed that the energy consumption could be considerably decreased from the values of the considerably decreased from the considerably d

# 3.5. CO<sub>2</sub> capture and storage

CO<sub>2</sub> was absorbed both in brine and solids. Figure 8A shows the amount of CO<sub>2</sub> absorbed at each of the 6 steps of the process in both brine and solids. The amount of CO<sub>2</sub> absorbed in brine is actually a cumulative value including the amounts absorbed in previous steps. Conversely, since solids are filtrated at the end of each step, the CO<sub>2</sub> absorbed amount corresponds solely to the indicated step. For ELECTROLYSIS 0 it should be recalled that CO<sub>2</sub> absorbed is directly captured from air. For ELECTROLYSIS E1.1. and 1.2, the solids retain lower quantities of CO<sub>2</sub> than brine. However, CO<sub>2</sub> captured in solids is known to be permanently stored, while the amount captured in brines could potentially be released depending on brine conditioning, storage or re-injection. During CHEMICAL C2, ELECTROLYSIS 1 and ELECTROLYSIS 2, the amount of CO<sub>2</sub> captured in solids increases considerably, which is ascribed to the start of the massive crystallization of Na<sub>2</sub>CO<sub>3</sub>, and the original larger content of Na<sup>+</sup> in brine as compared to Li<sup>+</sup>.

Figure 8B shows the total amount of  $CO_2$  absorbed in the two solid species,  $Li_2CO_3$ ,  $Na_2CO_3$ , and brine. It is interesting to note that if the processing would be interrupted without completion of 6 steps, the amount of absorbed  $CO_2$  in the solids would be lower than indicated in Figure 8B, since a lesser amount of solid would have been recovered. However, Figure 8A indicates that a higher amount of  $CO_2$  would have been captured in brine. This is explained by both the decreased  $CO_3^{2-}$  concentration in brines, as well as the decreasing brine volume as the processing proceeds (see above).  $CO_2$  permanent storage is a more sustainable solution than an intermediate solution where the fate of capture  $CO_2$  is yet to be determined (capture in brine). In addition, while  $Na_2CO_3$  has a considerably lower market value as compared to  $Li_2CO_3$ , it is still a commodity with extremely versatile applications that will find buyers, increasing the revenue of the process.

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The partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness than the partial cumulative plot illustrates the sequestration pathway of CO<sub>2</sub> across two vietness to the pathway of CO<sub>2</sub> across the pathway of CO<sub>3</sub> across two vietness two vietness than the pathway of CO<sub>3</sub> across two vietness than the pathway of CO<sub>3</sub> across two vietness than the pathway of CO<sub>3</sub> across two vietness two vietnes phases: dissolution in brines and mineralization in solid carbonates. A fraction of unreacted CO<sub>2</sub> dissolves into brines, accumulating as aqueous carbonate species. In the initial process steps (ELECTROLYSIS 1 and 2), lithium carbonate precipitation acts as the dominant mechanism for CO<sub>2</sub> storage. Here, CO2 is rapidly incorporated into crystalline lithium carbonate solid. During these lithium-recovery focused steps, a maximum CO<sub>2</sub> is accumulated in the brine. Beyond these initial steps, sodium carbonate emerges as the primary mineralization product. The continued precipitation of sodium carbonate sustains, and accounts for 86.8% of the total amount of mineralized CO<sub>2</sub>. The formation of Li<sub>2</sub>CO<sub>3</sub> occurs to a lesser extent, although its presence is technologically of utmost importance for the fabrication of rechargeable lithium ion batteries. The first two processing steps demonstrate that selective lithium capture is feasible without the need for additional chemical reagents, solely through electrochemical control of the environment, nucleating conditions, pH and  $CO_3^{2-}$ concentration, the last two easily achievable by an automatic control of CO<sub>2</sub> sparging rate and/or current density. The fact that a considerable fraction of CO2 remains in the brine suggests that there are opportunities for process optimization, which opens new possibilities both in terms of CO<sub>2</sub> absorption efficiency and mineralized product recovery.

Figure 8C shows CO<sub>2</sub> emissions associated to different energy sources that could likely be installed close to lithium brine mines in the Lithium Triangle, in South America. This region is rich in natural gas, with pipelines at relatively short distances from where extensions could be easily constructed.<sup>30,41</sup> The other non-renewable energy source would be oil.<sup>41</sup> In the absence of carbon capture and storage technology, natural gas has been associated to lower emissions than oil (Table 3). There are also ample solar and wind resources, with already installed infrastructure, 42,43 which explain our choices of renewable energy options. As expected, emissions associated to electricity production from oil and natural gas are the largest. Production from oil emits 1.73 times more CO<sub>2</sub> than production from natural gas, while both options emit over one order of magnitude more CO<sub>2</sub> than all other 3

renewable sources considered. Figure 8D shows a comparison of the total emissions (all 5 electrolyces) steps) vs. the total capture capacity of the 6 steps-technology. If either of the 3 renewable energy options was chosen, this proposal could be classified as carbon negative, considering that a net capture (total capture minus emissions) of 172.3; 184.0 and 195.7  $g_{CO_2}L_{brine}^{-1}$  is the balance for photovoltaics, concentrating solar power and wind, respectively. Conversely, the technology would still be carbon positive if either natural gas or oil is used.

# 4. Conclusions

Here a new strategy has been developed to promote a more sustainable lithium production from brines. The processing starts with an electrolytic brine polishing. The lithium carbonation process is promoted via CO<sub>2</sub> absorption and conversion to carbonate anions in the highly alkaline brine. Alkalinity is produced via a continuous electrolytic. These are zero-chemical treatments, as opposed to the current technology that requires lime (CaO) and soda ash (Na<sub>2</sub>CO<sub>3</sub>) to remove divalent cations and crystallize Li<sub>2</sub>CO<sub>3</sub>. The chemical input is replaced here by CO<sub>2</sub> absorption and electricity. 78 % of the original Li<sup>+</sup> content was recovered during two electrolytic processes coupled to CO<sub>2</sub> absorption as an impure Li<sub>2</sub>CO<sub>3</sub> primary solid. A further 11 % of the original Li<sup>+</sup> content was recovered in a solid that was composed mostly of Na<sub>2</sub>CO<sub>3</sub>, but for which a recovery strategy could be sought.

Beyond Li<sub>2</sub>CO<sub>3</sub> recovery, a chemical step was implemented to take advantage of the high brine pH to secure a larger CO<sub>2</sub> absorption. In the absence of water reduction at the cathode, the pH dropped after a few minutes of CO<sub>2</sub> absorption and Na<sub>2</sub>CO<sub>3</sub> crystallization. Thus, two new electrolytic steps were introduced to increase the amount of solid product formation and CO<sub>2</sub> mineralization. Overall, 71.8 % of the original Na<sup>+</sup> content was removed from the brine and recovered mostly as Na<sub>2</sub>CO<sub>3</sub>. All tests performed here were performed on real samples, pumped from South American lithium rich brine deposits and pre-processed by an active lithium mining company. The nature of the tested samples further validates the proposed technology due to the high brine complexity.

In this work, focus was made on the production of solid products and the maximization of SOLO absorption A total of 205.8  $g_{CO_2}L_{brine}^{-1}$  were absorbed in both brine and solids. Out of that amount, 189.7  $g_{CO_2}L_{brine}^{-1}$  are considered to have been permanently stored in highly stable mineralized products: Li<sub>2</sub>CO<sub>3</sub> and Na<sub>2</sub>CO<sub>3</sub>. CO<sub>2</sub> mineralization at ambient pressure and temperature in highly saline brines is still a field much to be explored. A large amount of work remains in terms of operative costs, where there is yet a lot of room for improvement. The lithium carbonation steps deliver a concentrated pulse of CO<sub>2</sub> storage in brines but the contribution to CO<sub>2</sub> mineralization is minimal as compared to accumulation in Na<sub>2</sub>CO<sub>3</sub>. In contrast, the sodium-dominated steps achieve high-yield mineralization in the last steps while permitting passive CO<sub>2</sub> dissolution in brines.

# **Supporting Information**

Current and potential plots for electrolytic steps. Picture of the experimental setup used for all electrochemical experiments. Anion exchange membrane specifications. Quantitative compositions of all solids. SEM micrographs for analytical grade commercial samples of NaCl and NaHCO<sub>3</sub>.

# **Authors contributions**

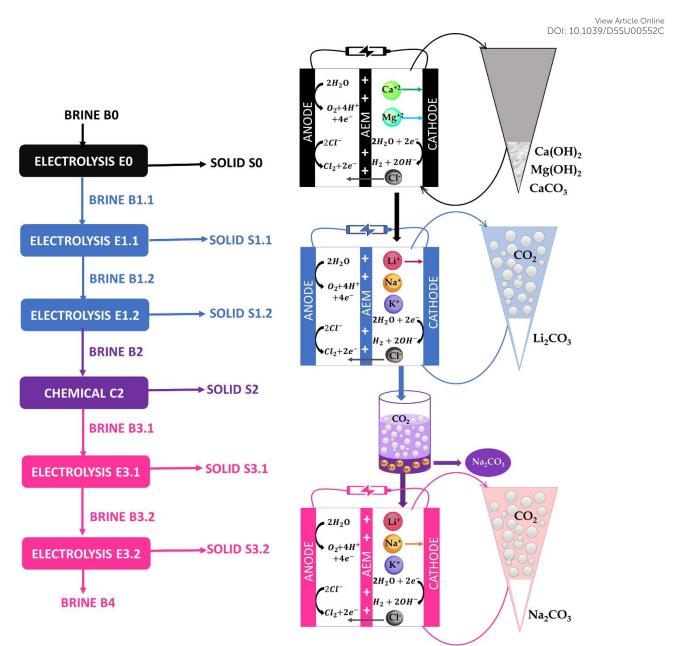
NCZ: Investigation, Data Curation, Formal Analysis. WRT: Conceptualization, Data Curation, Formal Analysis, Investigation, Methodology. Writing-original draft, Funding acquisition. VF: Conceptualization, Methodology, Formal Analysis, Writing-original draft, Writing-review & editing, Funding acquisition.

#### **Conflicts of interest**

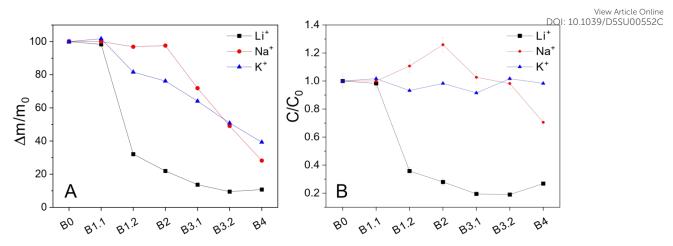
There are no conflicts of interest to declare.

#### Acknowledgments

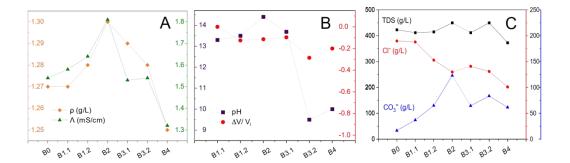
WRT, and VF are CONICET permanent research fellows. This work was supported by core funding from Universidad Nacional de Jujuy. Sistema Nacional de Microscopías and Sistema Nacional de Rayos X (AR) are acknowledged for the SEM and XRD determinations.



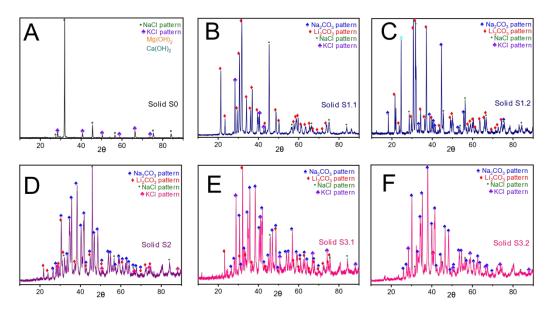
**Figure 1**. Representative scheme of methodology applied in the anhydrous carbon absorption from brines. In black, the process not-involving  $CO_2$  absorption. In blue, the methodology to co-obtain lithium salts. In purple, the chemical process for obtaining sodium crystals. In pink, the electrochemical process to co-obtain sodium salts.



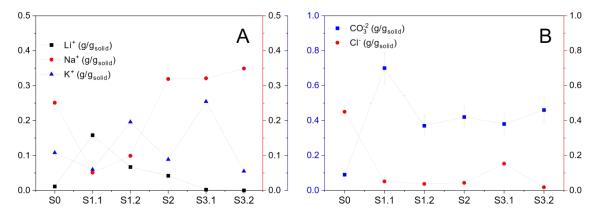
**Figure 2**. **A.** Percentage of the initial amount of the respective cations remaining in the processed brines. **B.** Concentrations (C) relative to the initial value  $(C_0)$  of  $Li^+$ ,  $Na^+$ , and  $K^+$  in brines at the different steps.



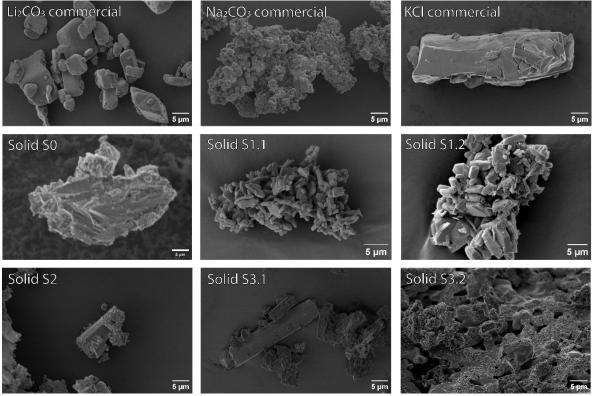
**Figure 3**. Physical and chemical parameters for the different brines. **A.** Density (direct measurement) and conductivity (1:500 dilution). **B.** pH and relative volume decrease, **C.** Total dissolved solids (TDS), chloride and carbonate concentrations.



**Figure 4**. X-ray diffractograms of the different solids obtained. Different diffraction peaks are possible obtained obtained of the different solids obtained. Different diffraction peaks are possible of the different solids obtained. Different diffraction peaks are possible of the different solids obtained. Different diffraction peaks are possible of the different solids obtained. Different diffraction peaks are possible of the different solids obtained. Different diffraction peaks are possible of the different solids obtained. Different diffraction peaks are possible of the different solids obtained.

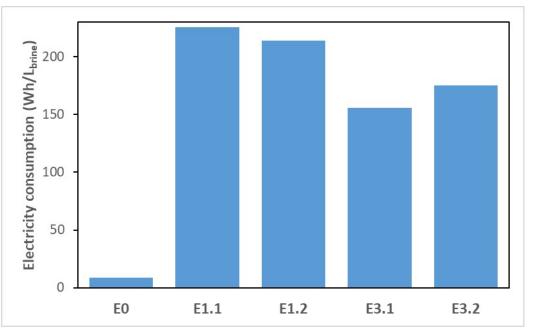


**Figure 5**. Composition of solids. **A.** Li<sup>+</sup>, Na<sup>+</sup>, and K<sup>+</sup>. B. Carbonate and chloride.

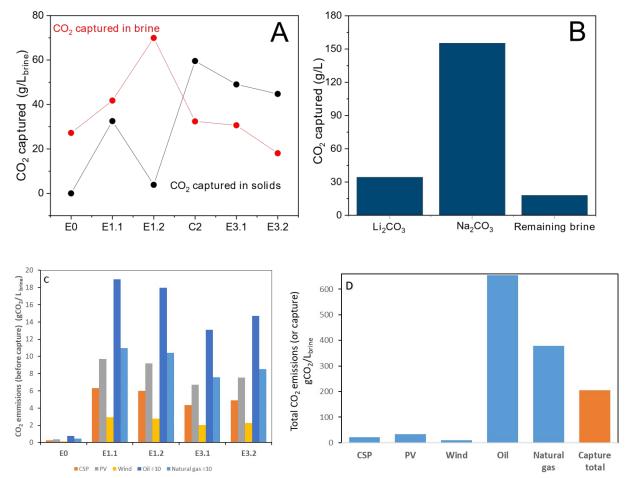


**Figure 6**. SEM micrographs of the different obtained solids and commercial pure samples of Li<sub>2</sub>CO<sub>3</sub>, Na<sub>2</sub>CO<sub>3</sub> and KCl, as indicated in each panel.

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**Figure 7**. Energy consumption for electrolytic steps.



**Figure 8. A.**  $CO_2$  captured at the different processing steps in brines and solids. **B.** Cumulative  $CO_2$  captured for the different solids and remaining brine (B4). **C.** Estimation of  $CO_2$  emissions for each electrolytic step for different possible energy sources. **D.** Comparison of total  $CO_2$  emissions and  $CO_2$  capture for the overall processing strategy.

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The data supporting this article have been included as part of the Supplementary Niew Article Online Information. No personal software code was used and no new crystallographic data was generated.