

Critical state analysis of two compacted filtered iron ore tailings with different gradings and mineralogy at different stages of treatment

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ABSTRACT: Slurry tailings storage in large impoundments has been largely used worldwide for a long time, as their cost is very competitive. However, recent disasters have brought to light the need to better comprehend the mechanics of the materials stored and to search for disposal alternatives to overcome the drawbacks. One possibility is the filtered tailings disposal (dry stacking) which requires a better understanding of the material's response in a dewatered (through filtration) and compacted condition. This paper compares two tailings from the same beneficiation (treatment) plant with different gradings and mineralogy, related to the beneficial processes they undergo. A series of triaxial tests comprising isotropic compression without shearing specimens, as well as isotropic compression followed by drained (CID) and undrained (CIU) shearing, and K-compression followed by undrained (CKU) shearing specimens were conducted over a range of confining pressures and initial compaction degrees. The experimental program allowed the evaluation of convergence for Normal Compression Lines (NCLs) and the analysis under the light of critical state soil mechanics for the stress-strain response of the tested materials. The research outcomes show that changes in iron ore tailings gradings due to different production processes and the use of different compaction degrees had an influence on its behavior (compression and shearing) at lower stress levels, while at higher stresses levels, this difference is erased and there is a convergence for unique and parallels NCL and CSL on $v - \ln p'$ plane with a spacing of 2.71. On the $p'-q$ plane both tailings showed a unique and similar CSL.

Keywords: Tailings, iron ore tailings, compacted filtered tailings disposal, dry stacking, critical state soil mechanics, static liquefaction.

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23 **1 Introduction**

24 Brazil is one of the largest iron ore producers in the world, with approximately 338 million
25 metric tons produced in 2020 [1]. Consequently, by-products of the extraction and processing
26 of these materials are generated and must be correctly treated and disposed to minimize
27 environmental and failure risks. Tailings are a byproduct of the mining industry, consisting of
28 the material left after extraction of the valuable fractions from ores.

29 The physical and chemical characteristics of tailings vary a lot due to the different compositions
30 of the parent ores, as well as the different extraction processes they undergo [2]. Commonly,
31 tailings present an aqueous slurry form comprising disintegrated rocks, chemicals, and elevated
32 amounts of water [3, 4]. These characteristics facilitate their disposal in large impoundments
33 designated as tailings dams. Nevertheless, elevated risks are associated to the operation and
34 maintenance of such structures, since the tailings are routinely found saturated and at a loose
35 state [5–7].

36 Worldwide, there have been over 240 major Tailings Storage Facilities (TSF) failures since
37 1960, being at least 5 catastrophic incidents in the last years. Amongst them, were cases such
38 as Mariana and Brumadinho in Brazil, which were upstream tailings dams and represented the
39 biggest disasters in Brazil’s history in terms of environmental pollution and loss of lives,
40 respectively [8]. For these reasons, the Brazilian legislation has been revised, indicating an
41 urgent need to safely decommission upstream tailings dams because these are the most
42 challenging structures to operate in the mining industry. Some of these structures are likely to
43 be subject to piping, collapse, and flow failure and should be monitored carefully [2, 9].

44 Other methods of disposal are being developed and proposed to overcome the risks involved
45 in such structures. In this regard, the filtered tailings disposal by dry stacking appears as an
46 alternative to the traditional disposal of tailings in impoundments and, as well, might be an
47 option for usage in the de-characterization process of an existing dam [10, 11]. Filtering of
48 tailings can take place using pressure or vacuum force. The filtered tailings are transported by
49 conveyor or truck and then placed, spread, and compacted to form an unsaturated, dense, and
50 stable tailings stack requiring no dam for retention [12].

51 In order to safely design the different disposal methods available and prevent failures,
52 Santamarina et al. [13] synthesized key concepts such as enhanced physical understanding,

53 effective engineering and management, and enforcement of regulations. Associated to the
54 design, usually, are the development and use of advanced constitutive models, mostly based on
55 a critical state framework.

56 Studies on liquefaction [14–17] have examined the influence of the fines content of the soil on
57 its behavior in the critical state, especially in relation to the shape and location of the CSL
58 (Critical State Line) in the $v - \ln p'$ plane. However, the findings of different authors seem
59 contradictory because of the lack of research covering different types of soils. Thevanayagam
60 et al. [18] found that the shape of CSL for different gradings remained similar, but as fines
61 were added, the location of the CSL initially moved downwards in the $v - \ln p'$ plane and then
62 moved upwards again. The authors defined as the “threshold fines content” the grading that
63 gave the lowest location on the CSL plane. The studies of Carrera et al. [14], who tested tailings
64 from the Stava disaster, agreed with the findings of Thevanayagam et al. [18]. Torrez-Cruz and
65 Santamarina [19] compiled published data and new experimental results and concluded that
66 the intercept of the CSL on $v - \ln p'$ plane as well as the value of e_{min} (minimum void ratio) are
67 lowest for mixtures with intermediate fines content (FC), typically $FC \approx 30\%$.

68 The amount of non-plastic fines has also been found to influence the compression behavior,
69 with most of the existing research on one-dimensional compression up to conventional stress
70 and some include high-stress levels. Carrera et al. [14] have found that the location of the 1D-
71 NCL moved downwards and reduced its inclination in the $e - \log(\sigma'_v)$ plane as the fine content
72 was increased, and the trend reversed for contents higher than 50–70%. Li and Coop [20]
73 identified unique 1D-NCLs for each of the three iron tailings from Panzihua, China, that they
74 tested. Also, the authors noted that the finest material reached a 1D-NCL much earlier than the
75 coarser materials. Several studies have highlighted the difficulty of achieving unique NCLs
76 (Normal Compression Lines) and CSLs for samples created at different initial conditions for
77 both natural soils and tailings [2, 20–24] which limits the characterization of these materials
78 under the critical state framework, indicating a dependence of the behavior on the initial
79 specific volume termed “transitional behavior”.

80 In this work, detailed laboratory tests were designed to investigate two iron ore tailings (IOT)
81 from the same tailings facility with different gradings, related to the beneficial (treatment)
82 processes they undergo. The two tailings present different proportions of sand and silt, as a
83 result of the different processing methods they were subjected to. Isotropic compression tests

84 were carried out to high pressures (120 MPa) in specimens with different initial densities to
85 study the effect of grading and the initial void ratio on the compressibility of the tailings and
86 the convergence of NCLs over a broad spectrum of pressures. Particle breakage after isotropic
87 compression tests was also evaluated, considering the different tailings and initial void ratios.
88 The shearing response of the materials was assessed through compression drained (CID) and
89 undrained (CIU and CKU) triaxial tests conducted over a wide range of confining pressures
90 (σ'_3 ranging from 50 to 2,400 kPa) and specific volumes ($v \approx 1.50$ to 2.10) to identify the CSLs
91 for each grading, and to examine liquefaction potential relative to the location of these lines.
92 Bender elements were used to analyze the stiffness of tailings with different gradings and
93 degrees of compaction. The effect of the compactness and a possible occurrence of transitional
94 behavior was also investigated on the compression and shearing as the specimens were
95 compacted to different initial specific volume values (v_0).

96 **2 Experimental program**

97 The experimental program comprised the physical characterization of both iron ore tailings and
98 the evaluation of their compression and shear behavior. A total of 50 triaxial compression tests
99 were conducted, with different initial densities, and 3 isotropic compression tests were
100 performed for each material. Also, bender elements were used in 4 tests to evaluate the small
101 strain stiffness of the tailings.

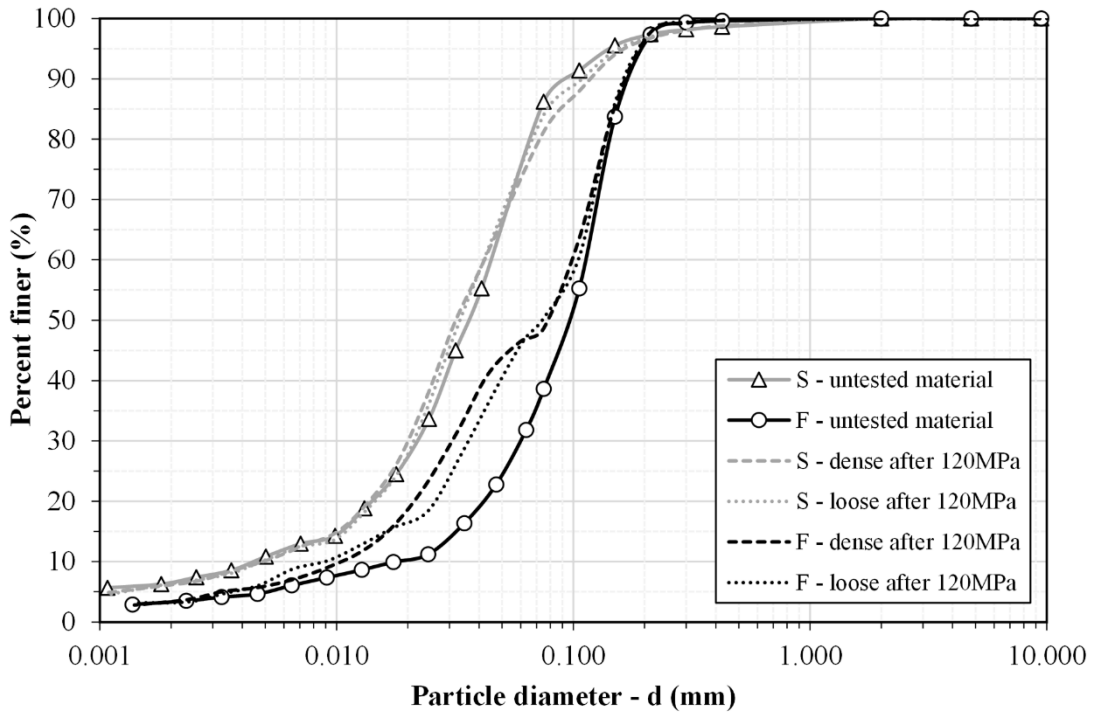
102 **2.1 Materials**

103 The iron ore tailings (IOT) studied herein is from Quadrilátero Ferrífero (QF), in the central
104 region of the state of Minas Gerais in Brazil. This region produces approximately 200 million
105 tons of iron ore per year and is responsible for approximately 65% of all Brazilian iron ore
106 production. Ore processing includes simple crushing and screening methods to more
107 sophisticated processes to upgrade the ore quality. After the screening stage, the fine material,
108 with gradation below the sinter feed (particles diameters from 0.15 to 6.3 mm), moves on to a
109 desliming stage through a sequence of hydro-cyclone batteries to remove the finer material
110 (slime tailings (S)). After this stage, the slimes tailings (S) are stored, and the remaining
111 material moves on to the reverse cationic flotation process, which isolates the fine-grained ore
112 (pellet feed), and results in the flotation tailings (F) that are also stored.

113 The iron ore tailings analyzed in this study are of the slime type (S) and flotation type (F),
114 obtained according to the procedure described above, and were collected in a disturbed state
115 from a mine at QF. The tailings samples were collected directly from the storage facilities. The
116 fine tailings (slimes) were excavated into the dam with the aid of a dragline. In contrast, the
117 tailings from the flotation stage were collected from the pile of filtered dry tailings. X-ray
118 diffraction analysis shows 54.8% of quartz, 42.3% of hematite and 2.9% of kaolinite for the
119 slimes tailings (S) and 98.3% of quartz and 1.7% hematite for the flotation tailings (F). Table
120 1 summarizes the main physical characteristics of both tailings, whereas Fig. 1 portrays the
121 grain size distributions. The particle-size distribution (PSD) was obtained following the ASTM
122 D7928 [25] standard. The Atterberg limits and the specific gravity were evaluated,
123 respectively, according to ASTM D4318 [26] and ASTM D854 [27]. The minimum (e_{min}) and
124 maximum (e_{max}) void ratios were evaluated in agreement with ASTM D4254 [28] and ASTM
125 D4253 [29] respectively. The compaction characteristics were assessed using both the standard
126 and modified efforts in agreement to ASTM D698 [30] and ASTM D1557 [31] and these results
127 are presented in Fig. 2. The slime tailings (S) are classified as low plasticity silt (ML) and the
128 flotation tailings (F) are classified as silty sand (SM) in accordance with the Unified Soil
129 Classification System [32].

130 Table 1 - Physical properties of iron ore tailings.

Parameters	S	F
Specific gravity – G_s	4.02	2.83
Gravel (%)	0.00	0.00
Coarse sand (%)	0.00	0.00
Medium sand (%)	1.39	0.40
Fine sand (%)	12.42	61.00
Silt (%)	79.93	35.12
Clay (%)	6.26	3.48
w_L (%)	-	-
PI (%)	Nonplastic	Nonplastic
Coefficient of uniformity	9.0	5.75
Coefficient of curvature	2.15	1.57
ASTM-USCS Classification	ML	SM
Maximum void ratio – e_{min}	0.565	0.503
Minimum void ratio – e_{max}	1.087	1.078
Optimum water content at standard effort – w_{opt} (%)	12.20	14.20
Maximum dry density at standard effort – γ_{dmax} (kN/m ³)	25.09	17.28
Optimum water content at modified effort – w_{opt} (%)	10.00	12.80
Maximum dry density at modified effort – γ_{dmax} (kN/m ³)	26.55	18.23

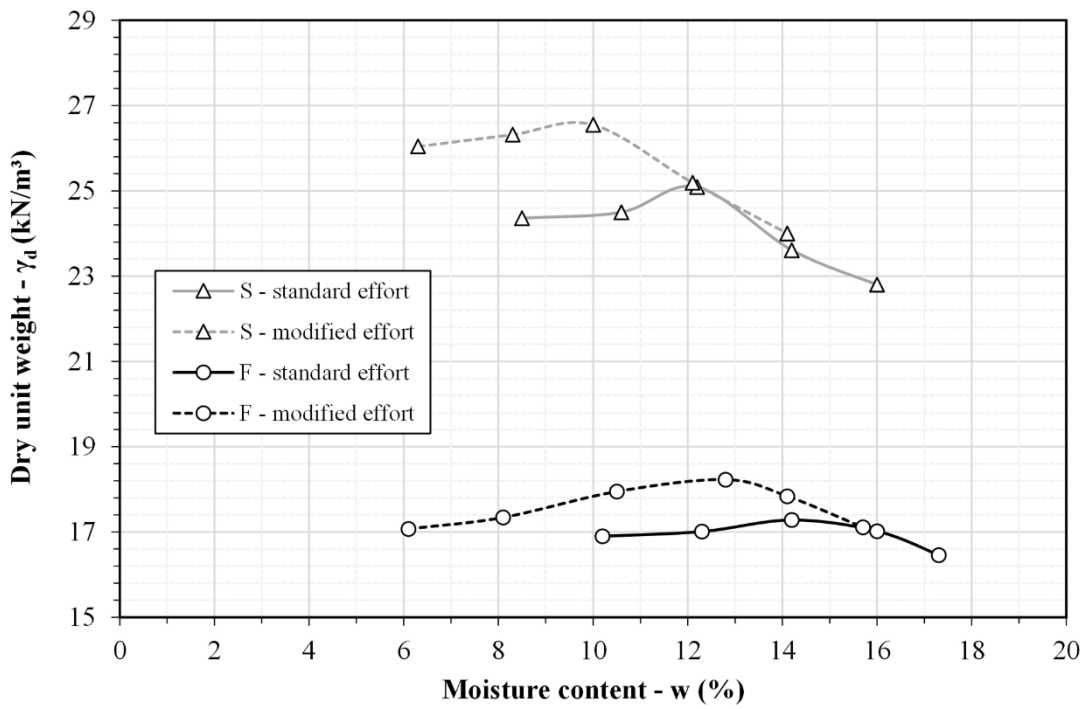


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Figure 1 – Particle size distribution of the iron ore tailings.

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Figure 2 – Compaction characteristics of the iron ore tailings.

136 As described above, differences in the physical properties of the two tailings are associated
137 with their production process. The slime tailings have finer particles (86.2% passing sieve
138 0.075mm) and considerably higher specific gravity due to the presence of particles with high
139 iron content. These tailings are retrieved at the initial stage of treatment, before the flotation
140 process. Thus, flotation tailings are coarser (38.6% passing sieve 0.075mm) because they are
141 obtained after desliming and have a lower specific gravity due to the recovery of pellet feed at
142 this stage. These characteristics also influence the compaction behavior of the materials since
143 the dry unit weight is much higher for the slime tailings due to its high specific gravity.

144 **2.2 Methods**

145 A total of 25 triaxial tests were carried out for flotation tailings and 25 for slime tailings. For
146 both materials, different stress paths were conducted, which allowed for assessing the stress-
147 strain behavior and liquefaction potential, as well as analyzing the results under the critical
148 state framework.

149 *2.2.1 Isotropic Compression Testing*

150 Isotropic compression tests were completed for loose and dense states of each material,
151 totalizing six tests. The specimens were prepared by moist tamping at three different molding
152 specific volumes for each tailings ($v = 1.65, 1.73$ and 1.88 for flotation specimens and $v = 1.67,$
153 1.73 and 1.79 for slimes specimens). For the isotropic compression tests, specimens with 50
154 mm in diameter and 100 mm in height were used. The maximum isotropic stress applied to the
155 specimens was about 120 MPa.

156 *2.2.2 Triaxial Compression Testing*

157 Triaxial tests were carried out following the determinations of ASTM D7181 [33] for drained
158 tests and ASTM D4767[34] for undrained tests. Cylindrical specimens having 70mm in
159 diameter and 140mm in height were molded through moist tamping [35, 36]. The specimens
160 were compacted to four degrees of compaction relative to the standard effort (i.e., 75% - 450
161 kN-m/m³ (75N), 85% - 510 kN-m/m³ (85N), 95% - 570 kN-m/m³ (95N), and 100% - 600 kN-
162 m/m³(100N)) and one degree relative to the modified effort (i.e., 100% - 2,700 kN-m/m³
163 (100M)). This spectrum was defined aiming to reproduce feasible levels of compactness for
164 use in dry stacking and to investigate the occurrence of “transitional behavior”. Such
165 occurrence would imply the dependence of the location of both the normal compression line
166 and the critical state line on the initial specific volume [21, 23]. In other words, a transitional

167 geomaterial would present multiple NCLs or CSLs, depending on the initial degree of
 168 compaction that would result in fabric differences that would not be fully erased during
 169 compression and/or shear. Based on the compaction curve results (Fig. 2) moisture contents
 170 (w) between 13% and 15% were chosen for flotation tailings (F) and 10% and 13% to slime
 171 tailings (S). Steel split molds were used and manual compaction in six layers of equivalent
 172 height was completed.

173 For the CID and CIU triaxial tests, the specimens were isotropically consolidated to a range of
 174 initial effective confining pressures ($\sigma'_3 = 50$ kPa to $\sigma'_3 = 2,400$ kPa). For the CKU tests, the
 175 consolidation stresses were gradually applied by servo-controlled adjustment of deviatoric load
 176 and cell pressure to the target coefficient of lateral pressure ($K = \sigma'_h/\sigma'_v$) of 0.70. Oversized
 177 end platens and lubricated “free-ends” were used to minimize the effects of end friction during
 178 the triaxial tests. After consolidation, the specimens were sheared either drained or undrained
 179 at a constant rate of 1.5% per hour for a total axial strain of approximately 30%. Pore-pressures
 180 or volume change, vertical load, and axial displacement were recorded continuously throughout
 181 the tests. At the end of the tests, the specimens were frozen, allowing the determination of the
 182 final water content and the accurate measurement of the initial and final void ratios. Table 2
 183 summarizes the data relative to the triaxial tests conducted on the flotation tailings and Table
 184 3 summarizes the data relative to the triaxial tests conducted on slime tailings. The notation
 185 used to identify each test is also presented in the tables.

186 Table 2 – Summary of triaxial compression tests data for flotation tailings.

Identification	Soil	Type	Compaction energy	ν at molding	ν at shear	p'_o (kPa)
1F	F	CID	75N	2.09	1.93	150
2F	F	CID	75N	2.08	1.90	300
3F	F	CID	75N	2.10	1.86	600
4F	F	CID	75N	2.11	1.83	1200
5F	F	CID	75N	2.10	1.80	2400
6F	F	CID	85N	1.90	1.89	100
7F	F	CID	85N	1.90	1.86	300
8F	F	CID	95N	1.72	1.74	100
9F	F	CID	95N	1.72	1.73	300
10F	F	CID	100N	1.65	1.67	50
11F	F	CID	100N	1.63	1.66	100
12F	F	CID	100M	1.57	1.59	200
13F	F	CID	100M	1.57	1.61	400
14F	F	CID	100M	1.57	1.58	1200
15F	F	CIU	75N	2.13	1.98	100
16F	F	CIU	75N	2.12	1.94	200

17F	F	CIU	75N	2.10	1.91	400
18F	F	CIU	75N	2.11	1.84	800
19F	F	CIU	75N	2.07	1.81	1600
20F	F	CKU	75N	2.08	1.89	400
21F	F	CKU	75N	2.09	1.86	800
22F	F	CKU	75N	2.09	1.80	1600
23F	F	CKU	85N	1.91	1.85	800
24F	F	CKU	95N	1.71	1.72	800
25F	F	CKU	95N	1.71	1.71	1600

187

188

Table 3 – Summary of triaxial compression tests data for slime tailings.

Identification	Soil	Type	Compaction energy	ν at molding	ν at shear	p'_o (kPa)
1S	S	CID	75N	2.09	1.82	150
2S	S	CID	75N	2.09	1.81	300
3S	S	CID	75N	2.11	1.78	600
4S	S	CID	75N	2.09	1.74	1200
5S	S	CID	75N	2.11	1.70	2400
6S	S	CID	85N	1.88	1.82	100
7S	S	CID	85N	1.88	1.81	300
8S	S	CID	95N	1.68	1.7	100
9S	S	CID	95N	1.67	1.69	300
10S	S	CID	100N	1.61	1.64	50
11S	S	CID	100N	1.61	1.63	100
12S	S	CID	100M	1.54	1.56	200
13S	S	CID	100M	1.53	1.55	400
14S	S	CID	100M	1.53	1.54	1200
15S	S	CIU	75N	2.05	1.86	100
16S	S	CIU	75N	2.10	1.83	200
17S	S	CIU	75N	2.11	1.79	400
18S	S	CIU	75N	2.10	1.75	800
19S	S	CIU	75N	2.07	1.70	1600
20S	S	CKU	75N	2.11	1.80	400
21S	S	CKU	75N	2.11	1.76	800
22S	S	CKU	75N	2.09	1.71	1600
23S	S	CKU	85N	1.86	1.77	800
24S	S	CKU	95N	1.68	1.69	800
25S	S	CKU	95N	1.67	1.65	1600

189

190 2.2.3 Bender Elements

191 Four tests (4F, 14F, 5S, and 14S) were isotropically consolidated incrementally in the following
 192 stages approximately: 50 kPa, 150 kPa, 300 kPa, 600 kPa, and 1,200 kPa. In each stage bender
 193 elements (BE) tests were completed. The specimens were then unloaded incrementally to
 194 approximately 600 kPa, 300 kPa, and 150 kPa and the BE tests were repeated in each unloading

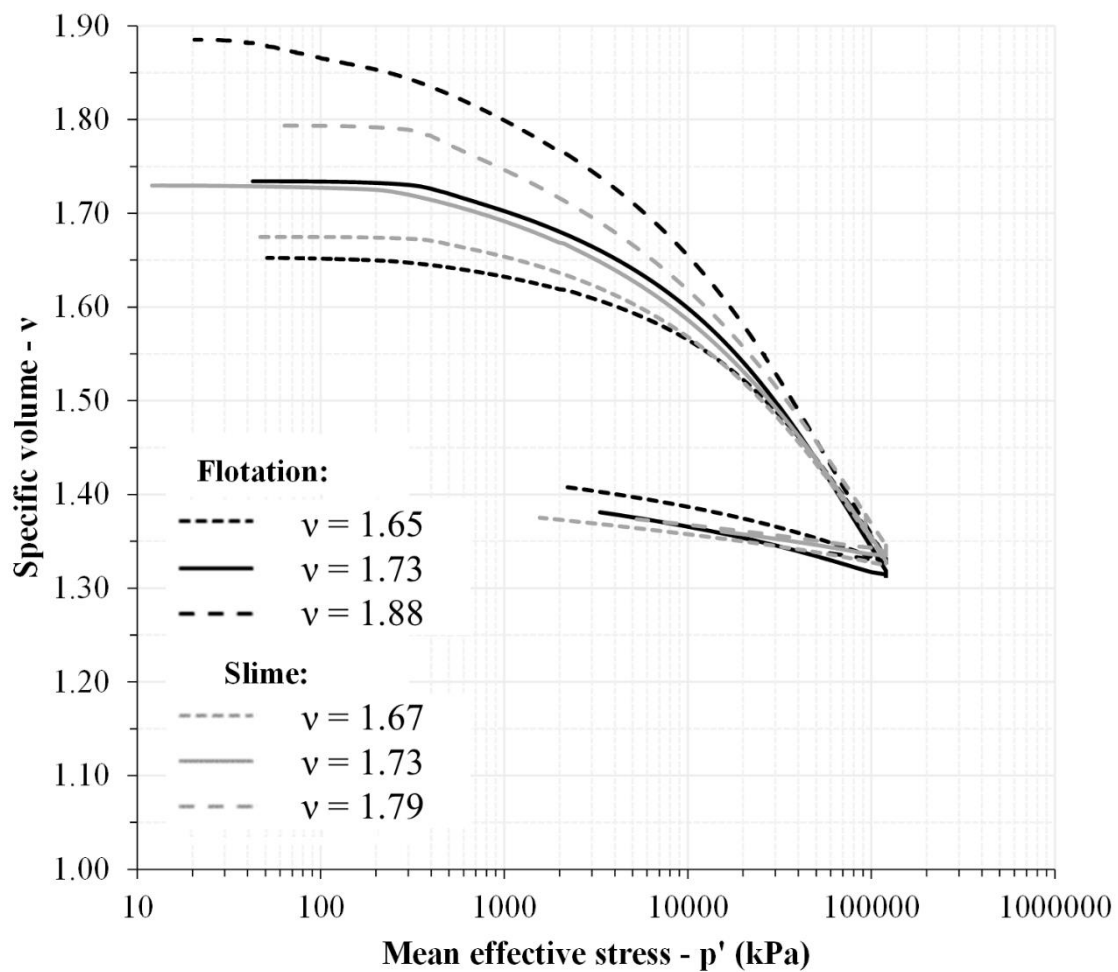
195 stage, after which the specimens were reloaded to the target confining pressures. The remaining
196 specimens were all isotropically consolidated in a single stage to the target pressures.

197 3 Results and discussion

198 3.1 Compression Behavior

199 3.1.1 Isotropic compression

200 The compression curves obtained for the two tailings are shown in Fig. 3. The convergence of
201 the compression curves from different initial v_0 is very slow, so that even at the maximum stress
202 level of about 120 MPa the curves have still not quite converged to a unique NCL. This suggests
203 that even for a very large tailings stack of 300 m height, the compression even at the base would
204 not cause a NCL to be reached.



205

206

Figure 3 – Isotropic compression results.

207 This is demonstrated in Figure 3, where all the specimens show the tendency to converge
208 towards the same specific volume at very high pressures (120 MPa). In this regard, a slight
209 change in the response can be roughly identified on the curves at lower stress levels, but a clear
210 yield point could be not identified due to scale reasons. Moreover, the response during the
211 unloading was much stiffer, indicating the occurrence of large plastic deformations. The
212 convergence to the NCL is slower for lower initial specific volumes because the soil
213 experiences much smaller volumetric strains and much higher stresses are therefore needed to
214 reach the NCL. When the initial specific volume is high, the compression curves tend to
215 converge more easily as the volumetric strains are larger.

216 It is also noted that the two tailings with different gradings tended to converge to exactly the
217 same NCL. For low stresses, the compression behavior of both tailings is also very similar,
218 with the finer material being slightly more compressible, but as the stresses increase, the
219 compressibility of the samples seems to be independent of fines content and mineralogy. As
220 will be discussed later, the flotation tailings underwent particle breakage during compression
221 but the slimes did not, which suggests that the breakage that occurred in the flotation tailings
222 did not control yielding or convergence to the NCL.

223 Both studied tailings can be called well-graded, their coefficients of uniformity of 9.0 for slime
224 tailings and 5.75 for flotation tailings are larger and their coefficients of curvature of 2.15 for
225 slime tailings and 1.57 for flotation tailings are in between 1 and 3 [37]. Several studies have
226 been reported in the literature on the compressive behavior of granular soils, the majority being
227 carried out in uniformly graded sands. Authors such as Nakata et al. [38] and McDowell [39]
228 performed one-dimensional compression tests on uniformly graded samples of silica sand and
229 showed that for a given void ratio, the yield stress depends on the particle size and increases
230 with the decrease in particle size. They also noted that the compression index after the yield
231 was independent of particle size.

232 Looking at the previous literature [2, 14, 20, 40] it is also possible to find other granular
233 materials with different fines content converging to a unique NCL (although in some cases it
234 has not been fully discussed), especially other tailings and sand-silt mixtures. Altuhafi and
235 Coop [40] indicated that uniformity has an important role in compressibility and the location
236 of the yield point in compression curves. The authors studied sands with distinct mineralogy in
237 different gradings (uniform to well-graded) in one-dimensional compression. The sands tested
238 up to 107 MPa showed different compression indexes for different initial gradings; however,

239 it is possible to notice a tendency of convergence to the same specific volume at higher stresses
240 for the two materials tested herein.

241 Carrera et al. [14] performed oedometer tests on the Stava sand and silt mixed at different
242 percentages with a maximum applied load of approximately 14 MPa. The authors found that
243 the 1D-NCL moves downwards, and the compression index decreases as the silt content
244 increases, with an inversion in the tendency at larger silt contents. The stresses achieved by the
245 authors were significantly lower than those of the present work and from the tendency of the
246 curves is possible that mixtures with different gradings could converge to the same NCL for
247 higher stresses.

248 Li and Coop [20] found that the compression index decreases with the increasing fines content,
249 which was attributed to a better filling of the voids by the finer particles. The authors identified
250 a unique one-dimensional normal compression line for each iron tailings grading studied,
251 considering a maximum vertical stress of 20 MPa. The convergence of the NCLs for the three
252 gradings studied was not observed for the tested stress level, although two of them show some
253 tendency to converge at very high stresses.

254 Li et al. [2] studied a silty gold tailings and compared it with the coarser gold tailings studied
255 by Bedin et al. [41]. By comparing the one-dimensional compression behavior of both tailings
256 until 20 MPa, it can be seen that the normal compression lines were in very similar locations,
257 despite the distinct initial gradings.

258 *3.1.2 Particle Breakage*

259 It is well known that granular materials may experience particle breakage during isotropic
260 compression to high mean effective stresses. The evolution of particle breakage during the
261 compression of granular soils and its influence on soil behavior has long been investigated [39,
262 40, 42–48]. In granular materials, particle breakage has been usually associated with yielding
263 [43]. To investigate the occurrence of particle breakage, the grain size distribution was
264 evaluated for two specimens of each tailings type (loose and dense) after being tested under
265 isotropic stress at 120 MPa. The aim was to examine the effects of particle packing on the
266 tailings' behavior, considering different initial gradings and densities.

267 The results are shown in Fig. 1 compared to the original grain size distributions. Also, the
268 relative breakage B_r defined by Hardin [49] was calculated. It was 0.16 for both densities of
269 flotation tailings and zero for both densities of slime tailings. Observing the curves and the

270 calculated values, there was no significant difference in the particle breakage of the flotation
271 tailings due to differences in the initial densities. This agrees with the findings of Coop and
272 Lee [43] and suggests that a unique relationship between particle breakage and effective stress
273 might exist independently of the initial density. Altuhafi and Coop [40], however, found that
274 there was more breakage for the samples with a higher initial void ratio after compression,
275 although the effect of the initial void ratio on the amount of breakage was less pronounced for
276 some soils.

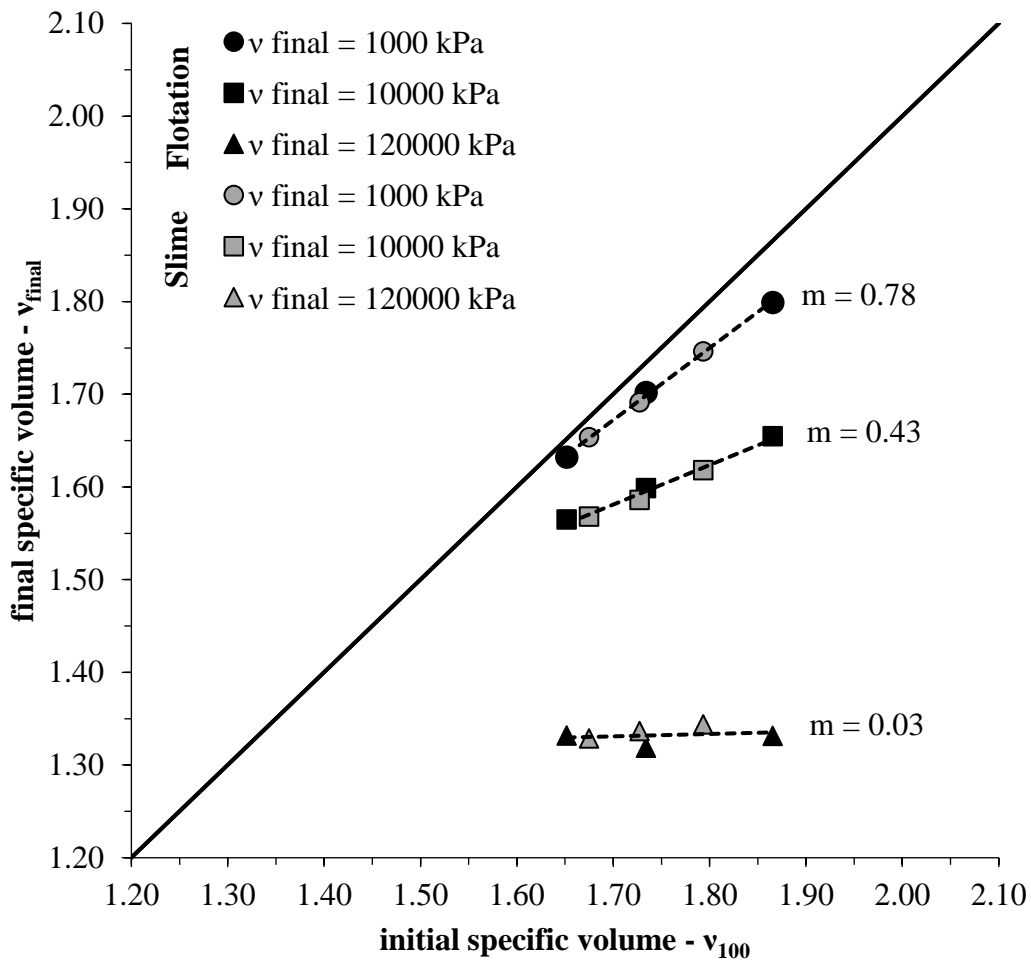
277 It is possibly the higher fines content of the slimes that prevents breakage. Similar behavior
278 was found by Li et al. [2], where the coarser tailings of Bedin et al. [41] was subject to particle
279 breakage while the finer was not. The particle size distribution is related to the packing of the
280 specimen. Muir Wood [24, 50] demonstrated by the discrete element method (DEM) that soils
281 with a wider particle size range (well-graded) have greater packing efficiency, resulting in a
282 higher coordination number for larger particles and thus reducing the probability of breakage.
283 Altuhafi and Coop [40] also demonstrated that the packing efficiency, and thus the coordination
284 number, can be increased by increasing the fines content of the sample. By adding fines and
285 changing the grading of different sands, particle breakage decreased until it ceased. In this case,
286 it is unusual, and perhaps a coincidence, that a tailings with breakage and one without converge
287 to exactly the same NCL.

288 *3.1.3 Quantification of convergence*

289 Ponzoni et al. [51] proposed a method to quantify the convergence of the compression curves
290 using an m value. The m value is based on the relation between an initial specific volume ($v_{initial}$)
291 and the specific volume at maximum stresses reached (v_{final}) and is analyzed with at least two
292 initial densities. The authors define that, for soils with fully convergent compression curves, m
293 is equal to 0 (same v_{final} for different $v_{initial}$). In other words, soils existing at initially different
294 v values, but that reach the same (or very similar) v value at the end of compression, present a
295 convergent compressive response and, thus, a single NCL. For soils with perfectly parallel
296 compression curves, m equals 1. This would indicate the existence of a family of parallel
297 compression curves, each one related to the initial specific volume.

298 Figure 4 shows m values for the two tailings based on the initial specific volume at 100 kPa of
299 mean effective stress (v_{100}) and at the maximum stresses reached (v_{120000}) at 120,000 kPa. Also,
300 m values are presented based on two intermediate stresses, 1,000 kPa (v_{1000}) and 10,000 kPa
301 (v_{10000}). These intermediate stresses were defined as, respectively, a usual limiting stress level

302 in the geotechnical practice and a possible stress level for application in dry stacks. Straight
 303 lines were assumed between v_{100} and the different final specific volumes and the m values are
 304 between 0 and 1 for both tailings. The results in Fig. 4 indicate again that the compression
 305 curves of the two materials are very similar, as the differences between both are smaller than
 306 the likely accuracy of the void ratios measured. The m values obtained for the three pressures
 307 ranges were very similar for both tailings regardless of the gradings, mineralogy and particle
 308 breakage. For this reason, the points from the compression curves of the different tailings were
 309 fed by a unique m value.



310

311 Figure 4 – Convergence lines in isotropic compression.

312 The m value that is not 0 indicates that the initial fabric is still not quite erased for the maximum
 313 stress applied. So, it is possible to conclude that at the usual pressures to 1 MPa the compression
 314 curves are very different and suggest a form of transitional behavior (m is close to 1). A similar
 315 tendency is noted in the pressure of 10 MPa, but a greater convergence tendency is confirmed

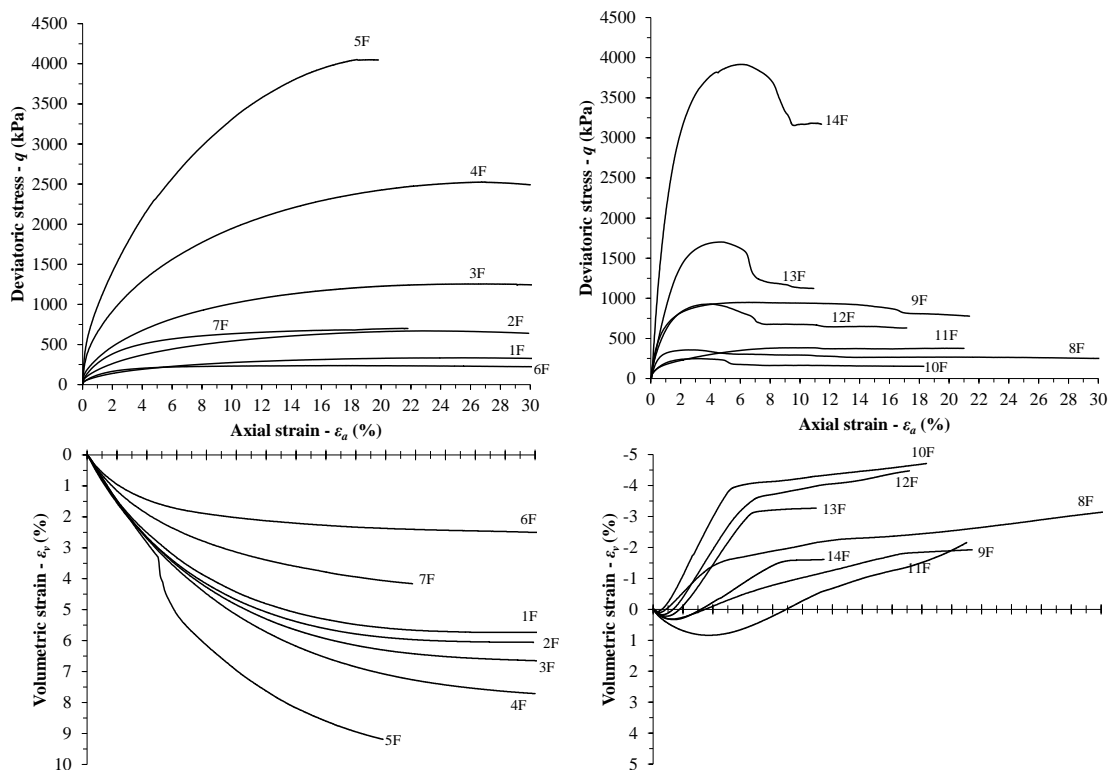
316 due to the decrease in m (m is between 0 and 1). Finally, at the highest stress of 120 MPa, all
 317 isotropic curves almost converge (m is close to 0).

318 These results demonstrate the importance of evaluating convergence by considering a stress
 319 level compatible with the material tested and its application. The m values found for the usual
 320 range of stresses applied in dry stacks indicate that the mode of behavior expected in these
 321 situations is still dependent on the initial fabric, where the deposition density significantly
 322 affects the in-situ volume and compression behavior of the material.

323 3.2 Shearing Behavior

324 3.2.1 Stress-strain data

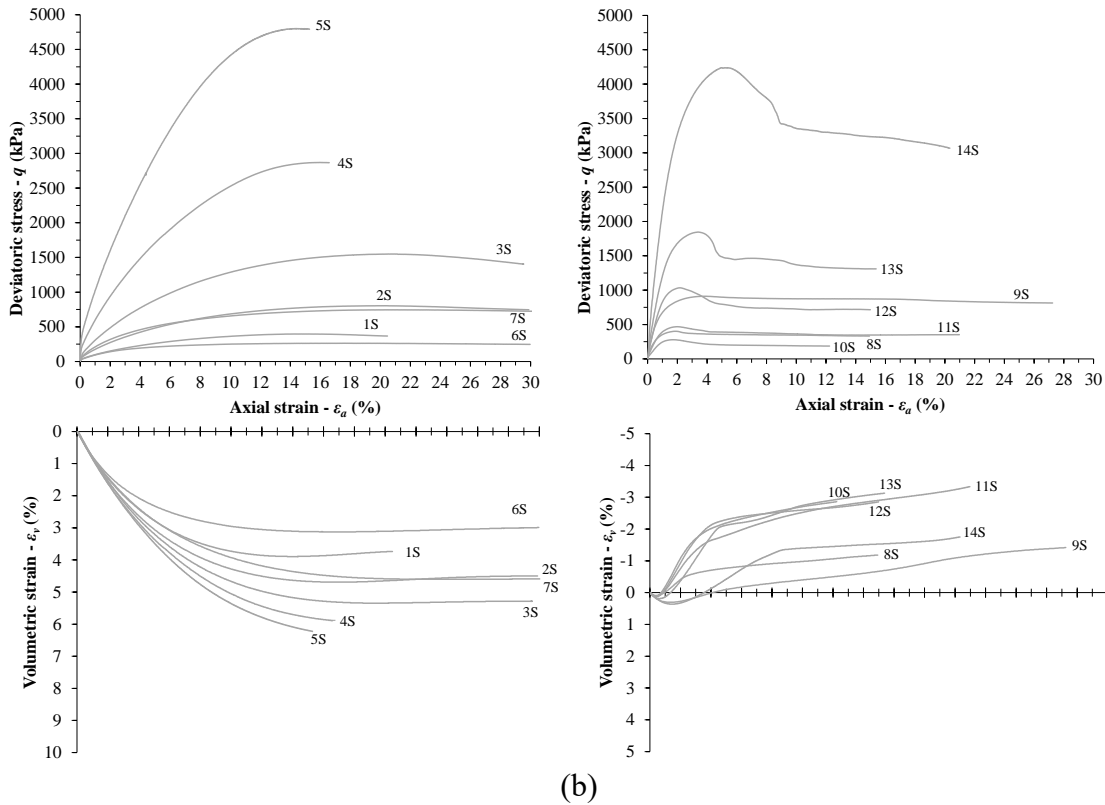
325 Figure 5 presents the stress-strain responses of the IOT under CID (consolidated isotropically
 326 drained) shearing tests. It is observed that the behavior of the two types of tailings was very
 327 similar. The loosest specimen types (1F/1S to 7F/7S) show a ductile response accompanied by
 328 a fully contractive behavior. On the other hand, specimens molded at the highest compaction
 329 degrees (8F/8S to 14F/14S) have a peak strength associated with an initial contractive response
 330 followed by a dilatant behavior.



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 332
 333

(a)

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339 Figure 5 – Stress-strain results of CID tests for: (a) flotation tailings and (b) slime tailings.

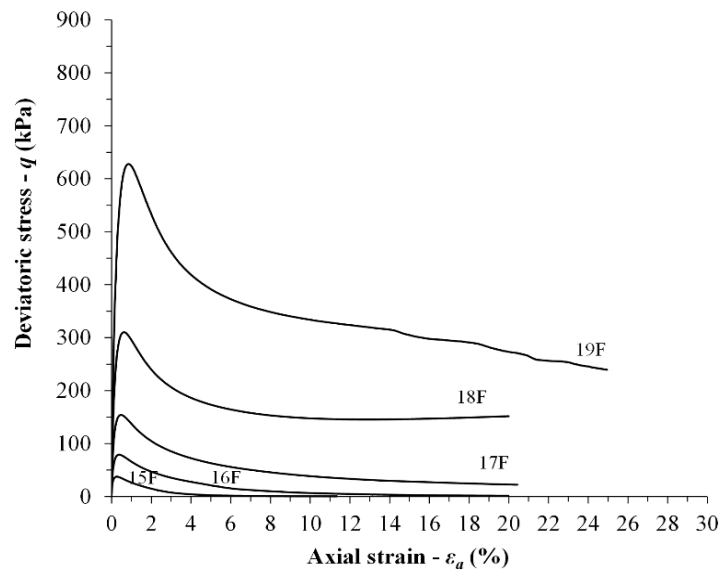
340 The area correction for the samples was carried out according to La Rochelle et al. [52].
341 However, it is difficult to obtain real strength values, especially for denser samples due to the
342 localization of strains. In this way, the dense tests were carefully assessed to avoid false
343 conclusions, so dense samples were not used to define the critical state line and only the
344 samples that showed no variation in both volumetric strains and deviatoric stress at large strain
345 levels were considered in the critical state analysis ($\delta\epsilon_v/\delta\epsilon_s = \delta\epsilon_q/\delta\epsilon_s = 0$).

346 Figure 6 presents the stress-strain responses of the iron ore tailings isotropically consolidated
347 and sheared under undrained conditions (CIU tests). Unlike the drained tests, the undrained
348 behavior of the two types of tailings presents significant differences, especially concerning the
349 stress-strain response. As all samples were molded in a loose state, a positive change of pore-
350 pressure was generated during shearing. For flotation tailings specimens (Fig. 6a), a substantial
351 loss of strength, accompanied by a strain softening behavior, was observed corresponding to
352 the positive pore-pressure increments registered during shearing. Hence, this has led to static
353 liquefaction or, at least, to susceptibility to it, especially for the specimens sheared at low levels
354 of initial effective confining pressure.

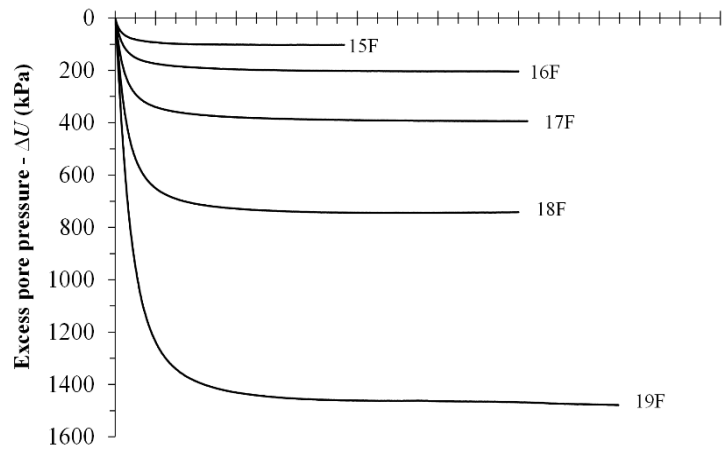
355 The slime tailings specimens (Fig. 6b) do not show the complete loss of strength, or true
356 liquefaction, although specimens 15S and 16S presented a peak strength followed by strain
357 softening behavior. Tests 17S, 18S, and 19S, which were sheared at higher confining pressures,
358 showed contractive behavior with peak strengths at up to approximately 1% strain followed by
359 a quasi-steady state which lasts for a short range of strains, after which the deviatoric stress
360 increases again with a tendency to dilatant behavior. Li et al. [2] and Li and Coop [20] found
361 similar behavior in undrained shear. The slime tailings were more compressible in the
362 consolidation than the flotation tailings, resulting in greater proximity in relation to the CSL
363 (see Fig. 12) at the beginning of the shearing and, consequently, smaller values of the state
364 parameter [53–55]. As a reason, the loosely compacted slime tailings samples (lying above the
365 CSL) were less contractive during shearing than the corresponding flotation tailings specimens,
366 implying a lesser amount of positive pore-pressure generation during undrained shear [56].

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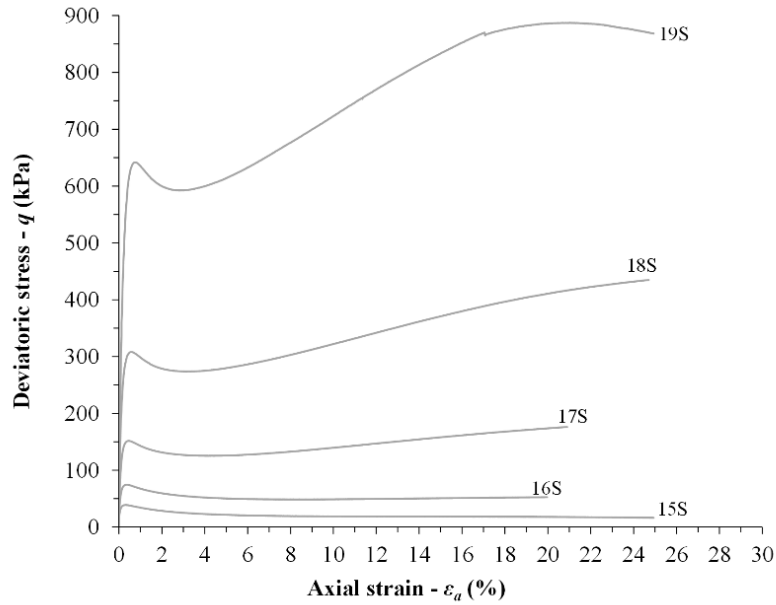


370

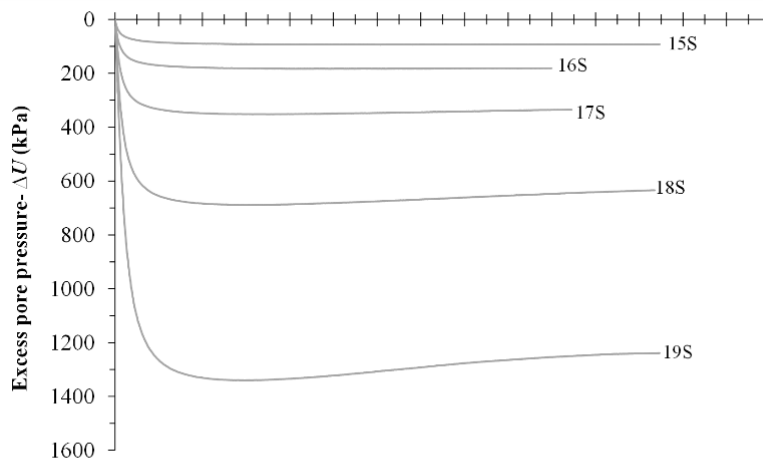
371

(a)

372



373



374

375

(b)

376

Figure 6 – Stress-strain results of CIU tests for: (a) flotation tailings and (b) slime tailings.

377

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379

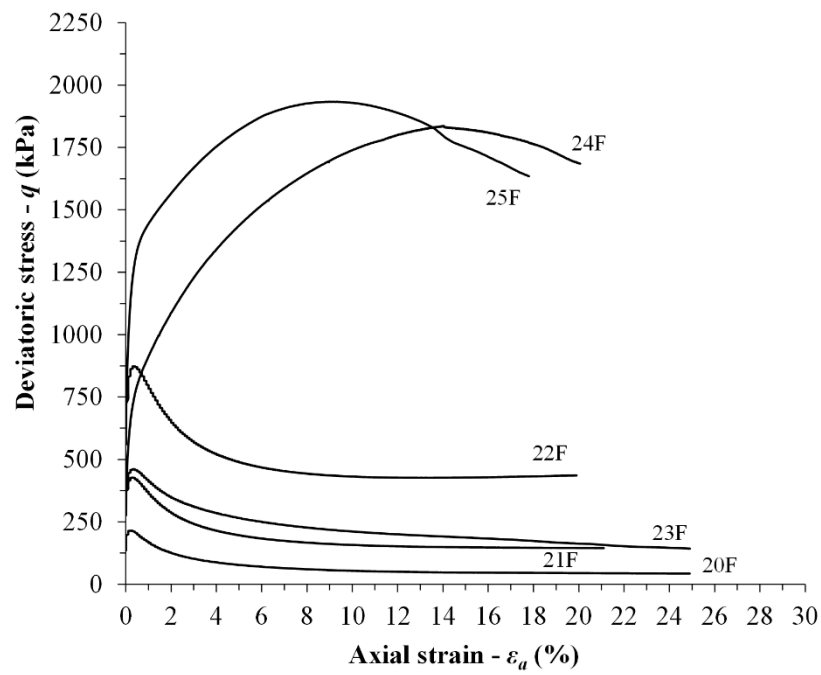
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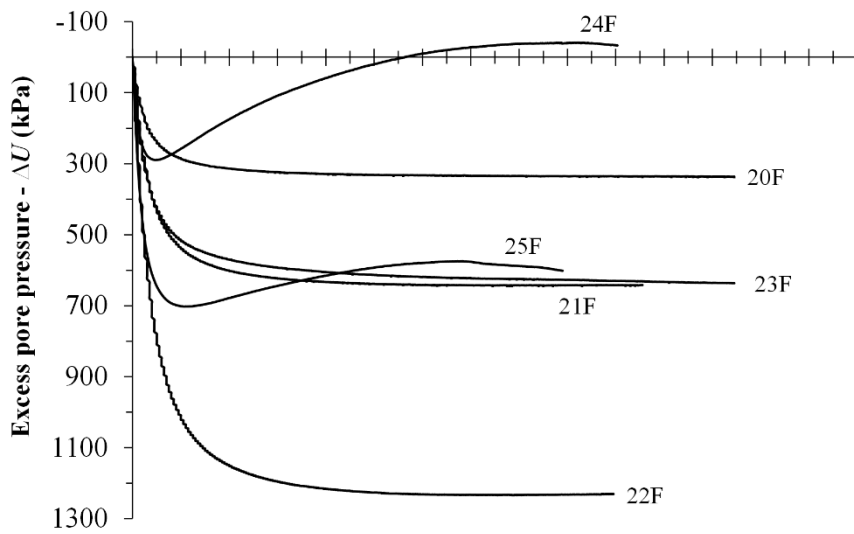
382

Figure 7 presents the stress-strain responses of the iron ore tailings K-consolidated and sheared under undrained conditions (CKU tests). It is noted that the K-consolidated specimens' response was similar to the isotropically consolidated specimens. The denser samples have an initial positive change of pore-pressure, followed by a negative pore-pressure generation trend accompanied by a slight peak strength. In general, the looser specimens have a loss of strength, accompanied by a strain-softening behavior, and positive pore-pressure increments were

383 observed during shearing. Strength loss was larger for the flotation tailings specimens, while it
384 was either slight or there was a strain hardening trend for some slime tailings specimens.



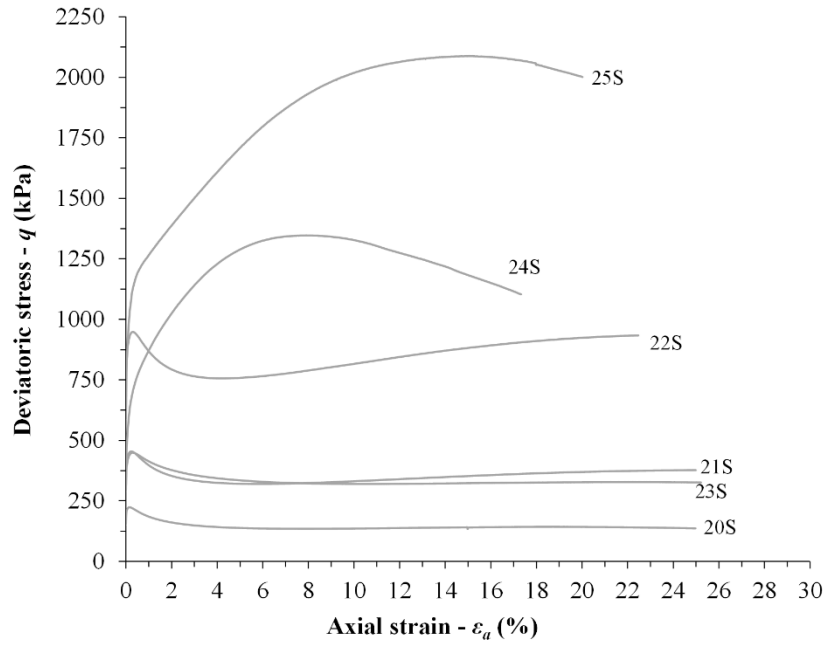
385



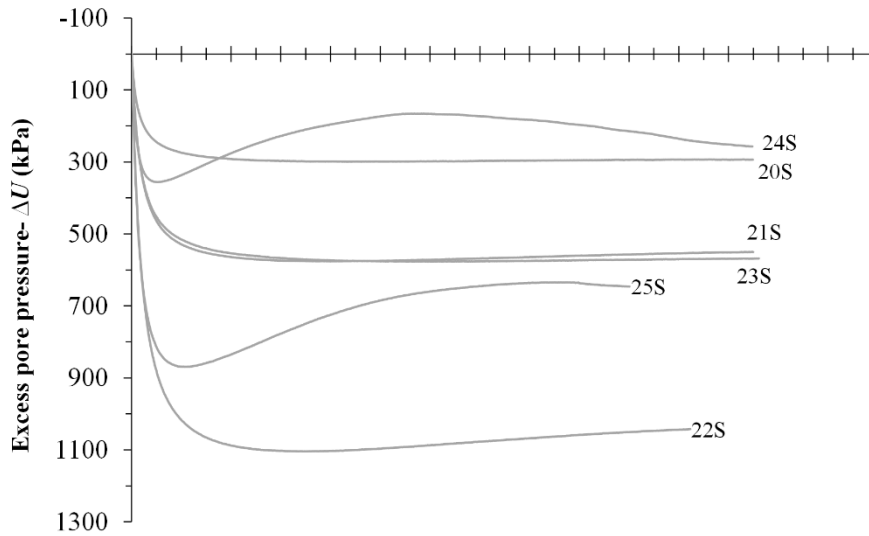
386

387

(a)



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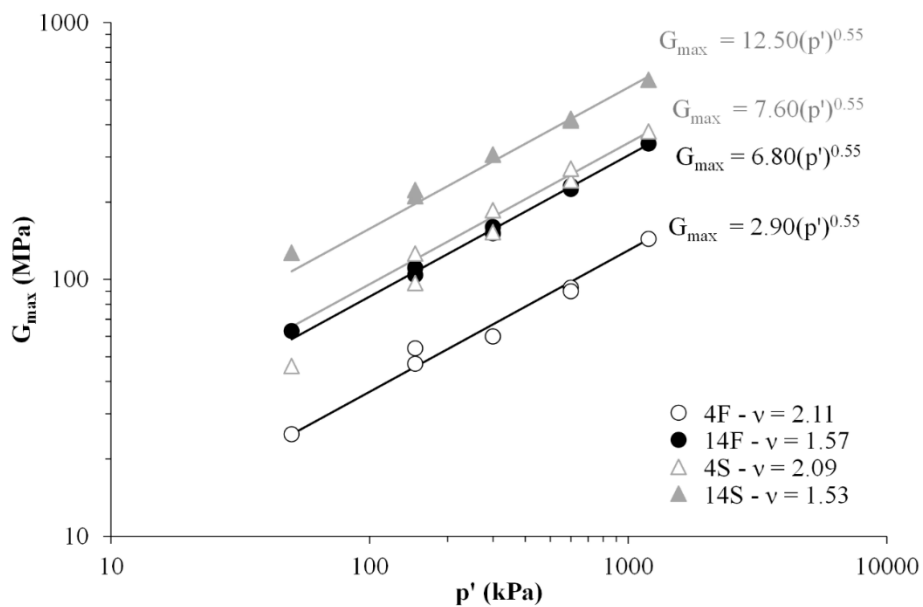
(b)

391 Figure 7 - Stress-strain results of CKU tests for: (a) flotation tailings and (b) slime tailings.

392 *3.2.2 Elastic shear modulus*

393 Figure 8 presents the elastic shear modulus (G_{max}) determined with BE testing to both tailings
 394 studied at different initial states. It is observed that the values of small strain shear modulus are
 395 influenced by the effects of initial density and the state of stress. The results agree with the

396 Hertzian contact theory which, ultimately, states that the small strain stiffness in a porous media
 397 is dependent upon the contact area between the constituent particles [57–59]. Hence, G_{\max}
 398 increases with the decrement in ν which, in turn, decreases owing to the augment of the mean
 399 effective stress (p'). In this regard, both results fit well a power law with the same exponent
 400 equal to 0.55. This suggests that the scalar accounts for the fabric existing differences between
 401 the slime and flotation tailings as well as within the samples molded using the same tailings
 402 but at different densities (loose and dense). Regarding the two types of tailings, the slime
 403 tailings were stiffer, possibly due to the higher fines content and better packing of the
 404 specimens, which guarantees a greater contact area between the particles, as well as because of
 405 the differences in mineralogy.

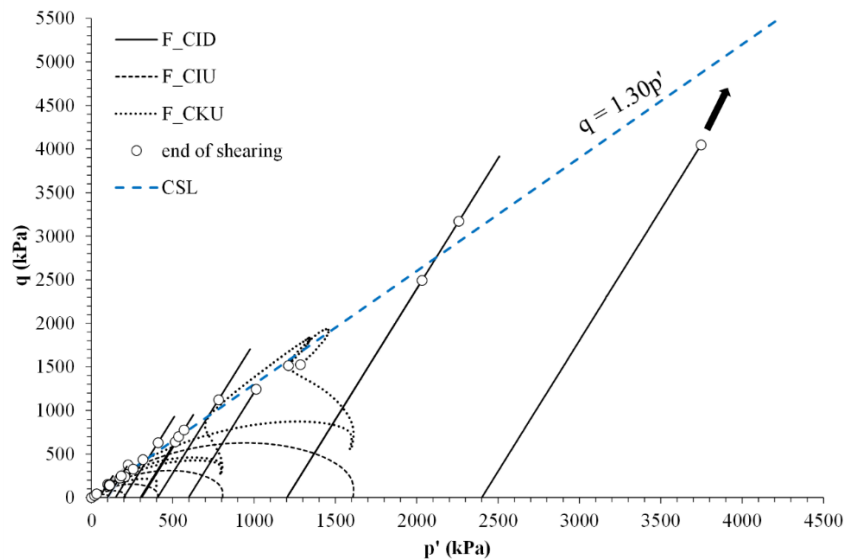


406
 407 Figure 8 - Small-strain shear modulus against the mean effective confining pressure for
 408 flotation and slime tailings in loose and dense conditions.

409 3.2.3 Stress paths and critical states

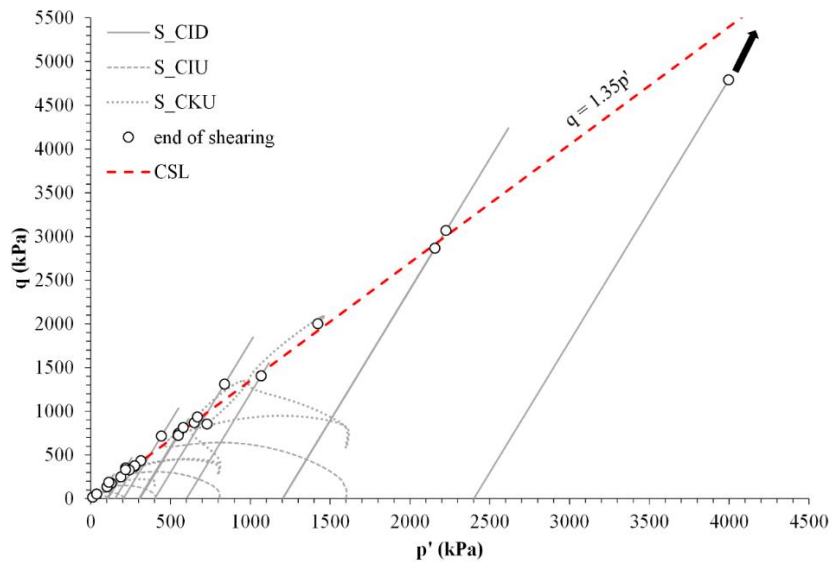
410 The critical state is defined by the absence of variations of volume, deviatoric stress, and mean
 411 normal effective stress during the shearing of a soil/tailings. This state is generally achieved at
 412 large strain values [60]. From the results previously discussed, most specimens underwent
 413 volumetric compression during shearing and reached or were close to reaching this condition
 414 at the end of the shearing.

415 The stress paths of the iron ore tailings studied are presented in Fig. 9. The flotation tailings
 416 are shown in Fig. 9a and it can be seen that the endpoints of these triaxial tests define a unique
 417 CSL in $q - p'$ space and the gradient of this line M_{tc} is 1.30 so the critical state angle of shearing
 418 resistance, ϕ'_{cs} is 32.2° . The stress paths of the slime tailings are shown in Fig. 9b and a unique
 419 CSL with a gradient M_{tc} of 1.35 is identified ($\phi'_{cs} = 33.4^\circ$).



420
 421

(a)



422
 423

(b)

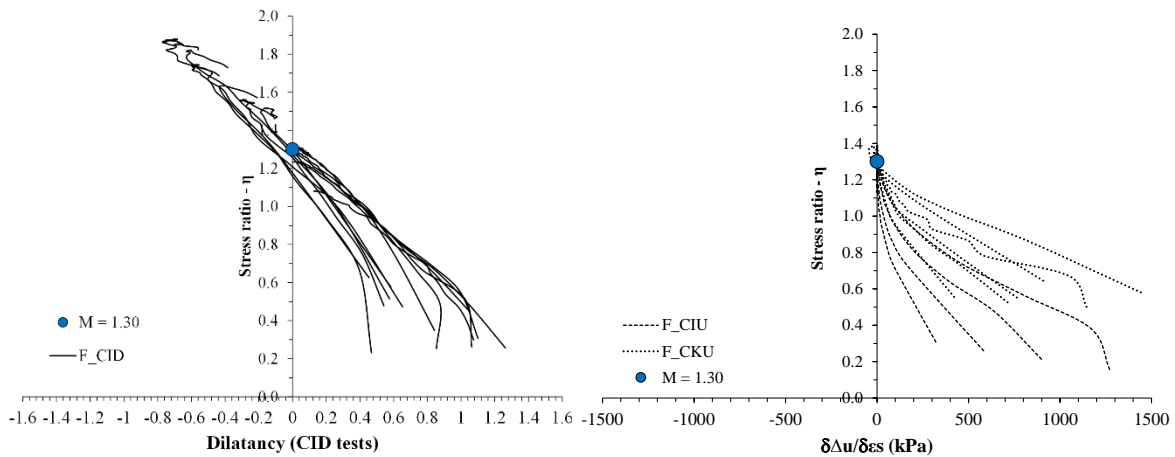
424 Figure 9 – Stress paths for triaxial tests for: (a) flotation tailings and (b) slime tailings.

425 The angle of shearing resistance at the critical state is principally controlled by grading, particle
426 shape, and inter-particle sliding friction. Li et al. [2] compiled several types of tailings and
427 concluded that all the ϕ'_{cs} values for different tailings fall into a narrow range at about $33^\circ \pm$
428 2° , which indicates that the particle sizes and mineralogy do not affect much ϕ'_{cs} . The values
429 obtained from the tailings of the present research are within the range indicated by Li et al. [2].

430 The difference between the critical state angle values for the different materials is minimal.
431 These results are similar to the findings by Carrera et al. [14]. The authors found that the critical
432 state angle varies only slightly for different sand-silt mixtures of the same tailings, probably
433 due to the common origin of the materials. Several studies [61–64] performed DEM
434 simulations to evaluate changes in critical states due to changing grading, with no reports of
435 significant differences in the q - p' plane.

436 However, the small increase in ϕ'_{cs} with the increase in fines content might suggest a slightly
437 greater strength for the finer material, possibly due to the interlocking portion as the fines
438 provide a better packing or possibly the differences in mineralogy, and pure friction can be
439 more related to their mineralogical origin. Some authors have also observed this difference in
440 the critical state angle with grading changes. Chang et al. [65] reported critical state angles
441 varying between 38.2° and 43.3° according to the grading variation of the material studied.
442 Bedin et al.[41], testing Brazilian gold tailings, found a critical state angle of 33° , while Li et
443 al. [2] found 34.8° for the same material with a finer grading.

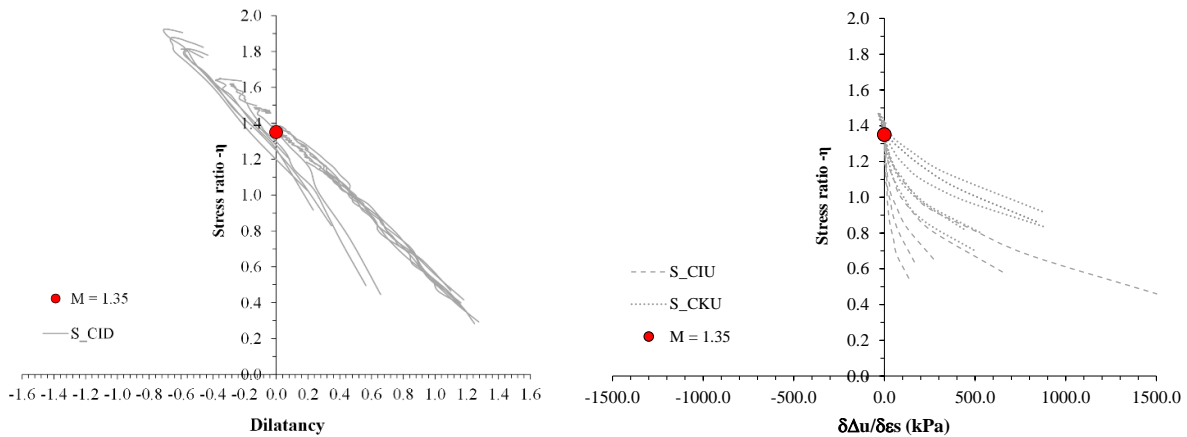
444 Figure 10 is helpful to predict and confirm the chosen value of stress ratio at the critical state
445 ($\eta = M_{tc}$) and verify which samples have effectively reached critical states. This is obtained by
446 plotting the stress ratio (η) against the dilatancy rate for the drained tests or the stress ratio
447 against the rate of change of excess pore-pressure ($\delta\Delta u/\delta\varepsilon_s$) for undrained tests. Following
448 Cuccovillo and Coop [66] and Coop and Wilson [67], the dilatancy was calculated considering
449 the total (plastic + elastic) volumetric and shear strain components. This is an adequate
450 approach considering that the recoverable strain parcels are only relevant at the very beginning
451 of the shear phase. Then, the M_{tc} value is observed at the intersection of the tests with the zero
452 dilatancy (or pore pressure change) axis and the tests that reached this point had attained the
453 critical state. Figure 10a presents the tests for flotation tailings with the M_{tc} of 1.30 and the Fig.
454 10b shows the tests for slime tailings with the M_{tc} of 1.35, which are in agreement with the
455 values defined in the q - p' plane. The loose tests reached the defined values of M_{tc} while the
456 dense ones are still tending towards this value.



457

458

(a)



459

460

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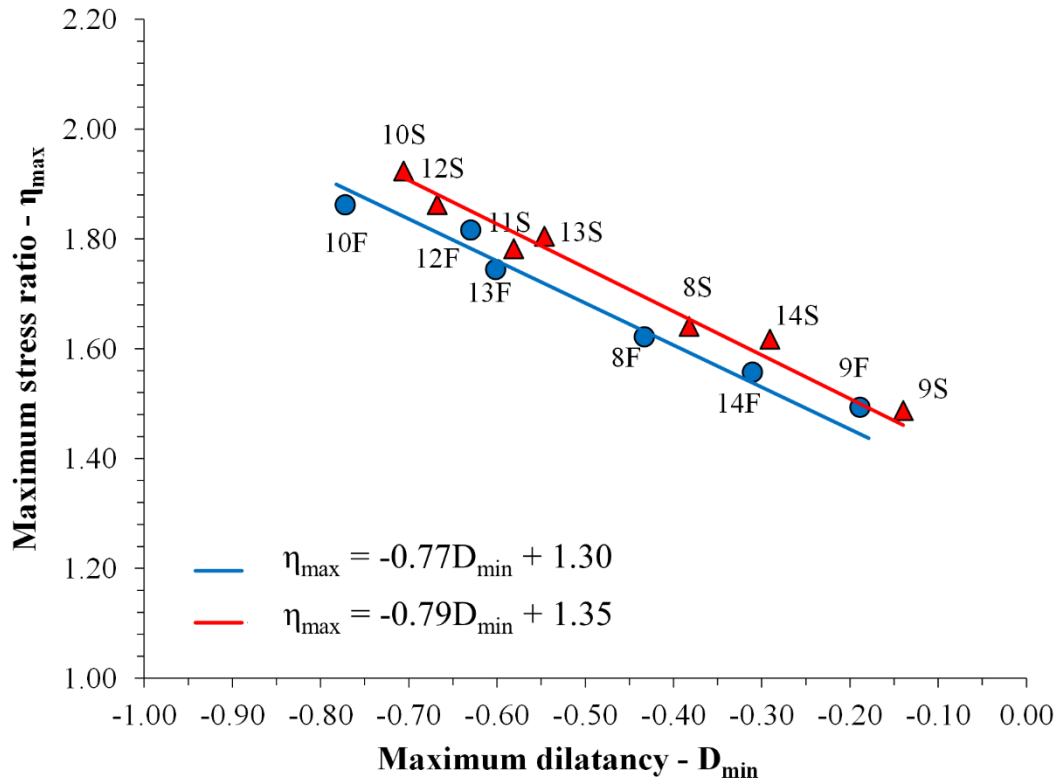
(b)

462 Figure 10 – Stress-dilatancy and stress-excess pore-pressure variation analysis for: (a)

463 flotation tailings and (b) slime tailings.

464 Figure 11 plots the maximum stress ratio ($\eta = q/p'$) versus the minimal dilation rate ($D_{min} =$
 465 $d\varepsilon_v/d\varepsilon_s$) for the dense triaxial tests. Higher η values were reached for more dilatant specimens
 466 (denser specimens at lower confining pressures), while $\eta = M_{tc}$ is observed for specimens at
 467 critical state and with no dilatancy ($D_{min} = 0$). Despite some scatter, the points lie close to a
 468 linear fitting for both tailings, and the lines fitted are almost parallel with slight translation
 469 upwards for the tailings with higher fines content (S). This similarity again is due to their

470 common origin and suggests a low influence of the grading and mineralogy on dilatancy and
 471 stress ratio, as previously observed.



472
 473
 474

Figure 11 – Stress-dilatancy correlation.

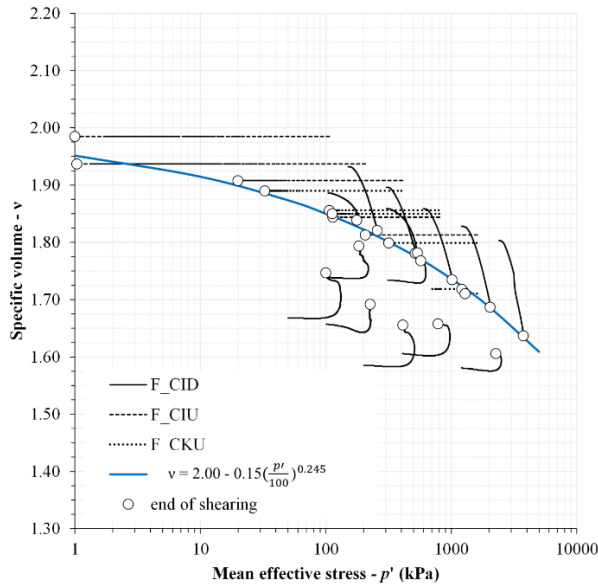
475 The critical state line (CSL) in the $v - \ln p'$ plane was determined by the values of v and p'
 476 corresponding to the end of the tests which were identified to reach critical state. For both
 477 tailings tested, dense specimens could not be used in CSL fitting as they had not yet achieved
 478 constant volume and stress states and almost certainly suffered from significant strain
 479 localization. The curved CSLs fitted were obtained through a power-law shown in Equation
 480 (1) [68] which attempts to include all the stress range tested.

481
$$v = a - b \cdot \ln\left(\frac{p'}{p'_{ref}}\right)^c \quad (1)$$

482 where a refers to the curve intercept, b refers to the initial curve slope, c is the exponent that
 483 controls the curvature and p'_{ref} was chosen as 100 kPa.

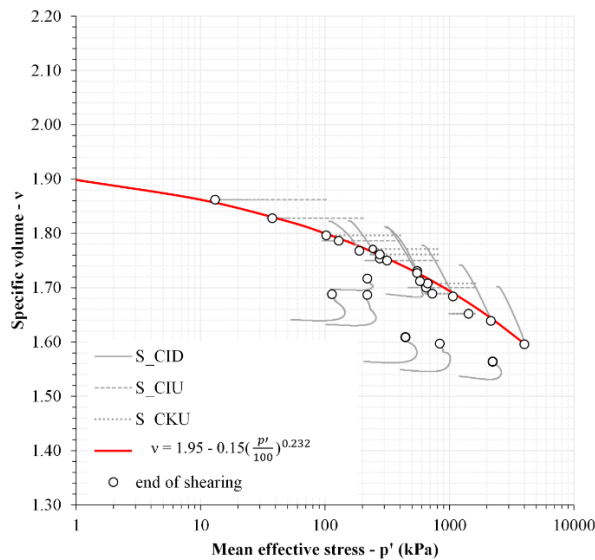
484 This idealization allows representation of the curvature in the CSL, that is clear for both
 485 flotation (Fig. 12a) and slime tailings (Fig. 12b). Over the stress levels tested it was possible to

486 achieve a unique CSL for both tailings, for the wide range of initial void ratios tested indicating
 487 the absence of a transitional mode of behavior on shearing. For three of the four initial densities
 488 tested herein, there is convergence to a unique CSL. However, Only the densest samples stay
 489 far with an upward movement towards the defined CSL, but this is more an artifact of
 490 incomplete testing and/or strain localization.



491
 492
 493

(a)



494
 495

(b)

496 Figure 12 – Critical state line in the $v - \ln p'$ plane for: (a) flotation tailings and (b) slime
 497 tailings.

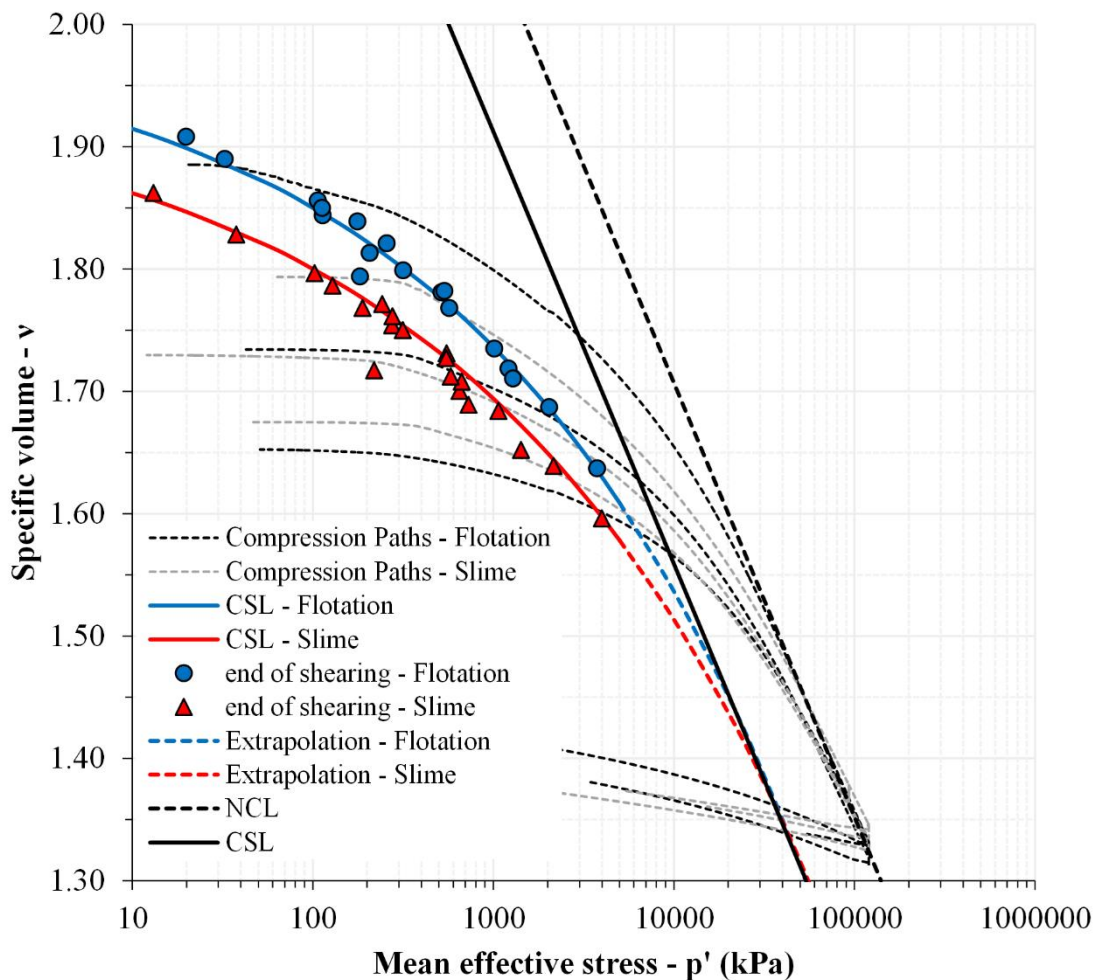
498 The CSL defined for the slime tailings has the parameters $a = 1.95$, $b = 0.15$ and $c = 0.232$,
499 while the adjustment to the CSL for flotation tailings gave $a = 2.00$, $b = 0.15$ and $c = 0.245$.
500 The curves obtained have identical slopes, different exponents, and different intercepts, so the
501 CSLs determined are almost parallel at low pressures with the intercept shifting downwards
502 owing to the fabrics-related differences that end up affecting the intergranular stress
503 distribution. Physically, non-plastic fines tend to fill the voids existing between larger grains
504 and reduce the void ratio up to a point termed threshold fines content (TFC) after which the
505 void ratio starts to increase because the increasing fines content tend to separate the larger
506 particles [18, 69] . This ends up impacting the attainable maximum and minimum void ratio
507 values and, thus, the location of the CSL. These findings agree with the behavior usually
508 reported in the literature. Different authors [14, 18, 70] have also observed the translation of
509 the curve with a change in fines content.

510 Bandini and Coop [71] suggest that the CSL movement with increasing fines is a result of both
511 vertical translation and rotation. These effects can also be noted for the materials tested. As the
512 stress level increases, the coarser material has a greater shift in gradient, while for the finer
513 there is only a slight difference in CSL gradient. Thus, the final gradients of each tailings vary
514 according to the fines content with a trend of convergence at higher pressures.

515 As also noted by Carrera et al. [14] although the different gradings have similar CSLs in the q -
516 p' plane, in the $v - \ln p'$ plane they are significantly different at low pressures. The tendency of
517 behavior was also explored at high pressure in this paper. Figure 13 shows the projection of
518 the CSL for each tailing type compared with the unique isotropic NCL obtained. The endpoints
519 of the shearing tests that reached the critical state are highlighted for each CSL. The two CSLs
520 tend to converge at stresses around 30 MPa to a unique CSL parallel to the limiting NCL and
521 with a spacing ratio (r) of 2.71. This value is similar to clays and also other non-plastic
522 materials such as iron tailings (2.7 [72]), gold tailings (2.2 [2]) and copper tailings (3.3[72])
523 and is the quotient between the normal consolidation pressure (p'_o) and the corresponding
524 pressure at the critical state line (p'_c) considering the same specific volume [73].

525 The behavior of both tailings with different gradings was found to converge at higher pressures
526 to a unique NCL and CSL. This trend of convergence can also be noted in other experimental
527 results found in the literature [2, 14], although the authors have not discussed it. These results
528 suggest that changing the grading and mineralogy might not necessarily influence the critical
529 state of the materials at very high pressures. At lower stress levels these two tailings are not

530 strictly transitional, or at least it is a different class of transitional behavior. The compression
 531 curves are each non-unique, but the critical state is unique so that the strains imposed during
 532 compression cannot have been sufficient to reach a unique fabric but the shear and volumetric
 533 strains occurring during shearing to failure were. This contrasts with the soils investigated by
 534 Todisco and Coop [74] for which no volumetric or shear strains brought about unique volumes
 535 (and hence fabric) states in either compression or shear.



536
 537 Figure 13 – CSLs and compression paths converging into unique CSL and NCL.

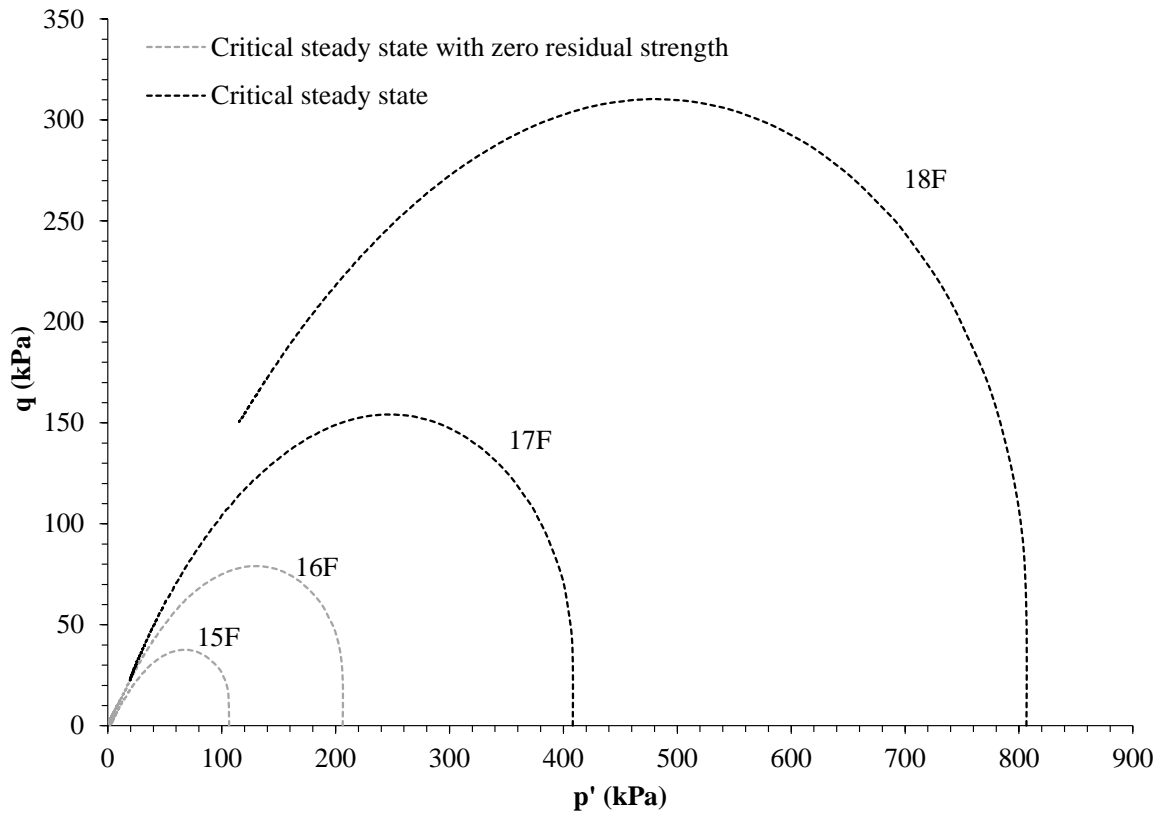
538 *3.2.4 Liquefaction*

539 The concept of liquefaction is approached in different ways in the literature, and it is often hard
 540 to understand which definition each author adopts. Herein, liquefaction is referred to the
 541 transformation of granular material from a solid state into a liquefied state due to increased

542 pore-water pressure up to the achievement of a zero effective stress state, characterized by a
543 complete loss of strength. Analogously, such a state can be termed a critical steady state with
544 zero residual strength [75]. In this regard, the curved critical state line (flatter at lower stresses
545 and parallel to the NCL at higher stresses) gives rise to a changing susceptibility to liquefaction
546 as the stress level increases, as also noted by Li et al. [2] and Carrera et al. [14]. Bedin et al.
547 [41] have defined different regions for curved CSL, indicating the changing susceptibility to
548 liquefaction as stress level increases. The regions indicated are: (1) true liquefaction, which
549 corresponds to low stresses and high specific volumes for which there is a complete loss of
550 strength; (2) flow instability, which ranges from low to moderate stress with typical strain
551 softening behavior and positive excess pore-pressure generation, which significantly reduces
552 deviatoric stresses, but still, achieve steady states with a finite residual shear strength; (3) stable
553 condition, drained and undrained critical states coincide and a stable slope essentially parallel
554 to the NCL; and, (4) particle breakage, when the stresses reached are greater than those at the
555 onset of shear-induced grain crushing.

556 Figure 14 shows examples of stress paths for different responses obtained during this research
557 for flotation tailings (Fig. 14a) and slime tailings (Fig. 14b). Regarding flotation tailings, it is
558 observed that tests 15F and 16F suffered a severe strain softening which has led to the complete
559 loss of strength, characterizing a steady state with no residual strength (p' and q tends to zero).
560 Tests 17F and 18F have also shown strain softening but reached an ultimate steady state in
561 which the residual strength was different from zero. This is an indication that the complete loss
562 of strength in an undrained response is not reached amongst all specimens whose states are
563 lying above the CSL, as well as the susceptibility to it depends more on the initial void ratio
564 than on the stress state of the soil.

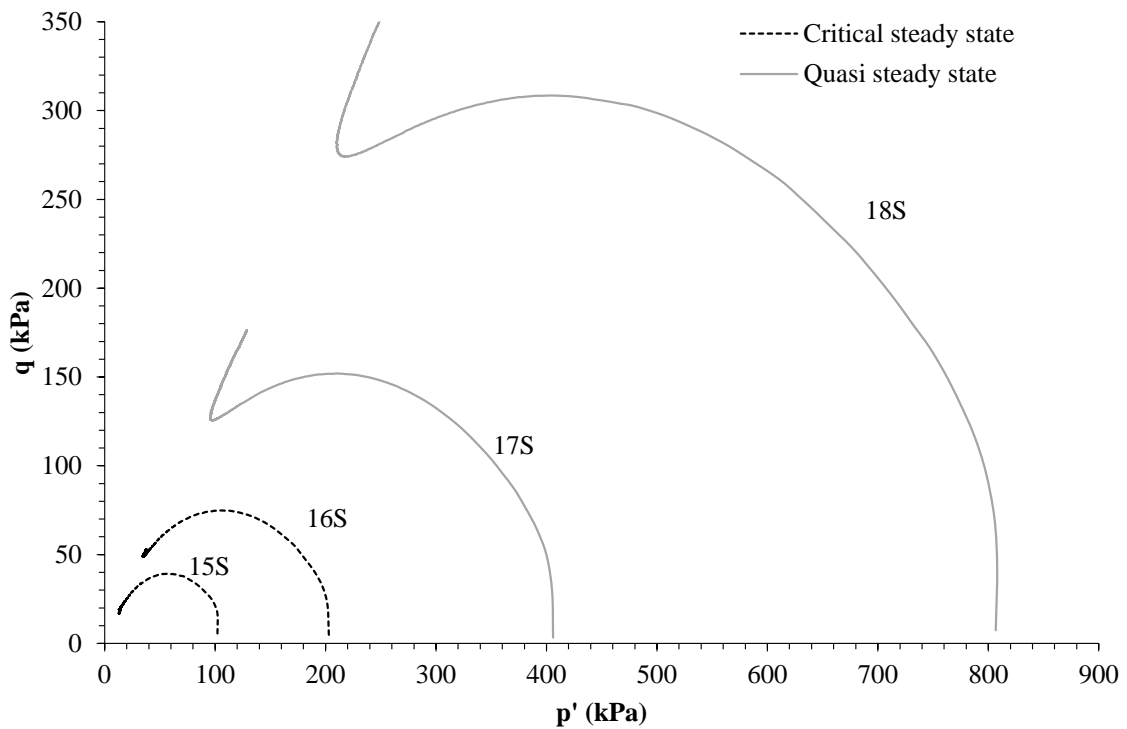
565



566

567

(a)



568

569

(b)

570

Figure 14 – Liquefaction analysis for: (a) flotation tailings and (b) slime tailings.

571 Considering the same tests on slime tailings, none has shown a complete loss of strength. Tests
572 15S and 16S have exhibited strain-softening that has led to a critical steady state with residual
573 strength different from zero. At higher pressures, the tests (17S and 18S) presented a
574 contractive behavior until a minimum mean effective stress or phase transformation was
575 reached, with the pore water pressures increasing quickly, and then the behavior changed to a
576 dilative trend. In other words, a quasi-steady state followed the phase-transformation point with
577 subsequent strain-hardening. This trend has been reported by Li and Coop [20] in other tailings.

578 The complete loss of strength in undrained loading occurs when the initial state lies above the
579 asymptote that exists for any CSL at lower stress levels, as is clearer for the flotation tailings
580 (Fig. 12a). Nonetheless, Figure 13 shows that this asymptote is higher for the flotation tailings
581 than the slimes, and given the e_{max} is similar for both tailings (Table 1) the slimes should
582 therefore be more susceptible to liquefy. The fact that, it is the flotation tailings that are more
583 prone to complete strength loss is due to the much higher initial specific volumes that were
584 achieved for this material (Table 2) than for the slimes (Table 3) despite using similar
585 preparation techniques. As well, the slime tailings lying above the CSL presented lower state
586 parameter values, indicating a lower contractive trend than the flotation tailings.

587 **4 Concluding remarks**

588 The present research evaluated the mechanical response of two compacted filtered iron ore
589 tailings from same origin but different gradings by means of drained and undrained triaxial
590 tests. The following conclusions can be drawn:

- 591 - A unique Normal Compression Line (NCL) was determined for both tailings with different
592 gradings and different initial densities when tested to stresses up to 120 MPa. However,
593 there were almost parallel compression curves at lower stress levels, with compression
594 paths that must be associated with differences in fabric corresponding to their initial void
595 ratios. These results demonstrate that the deposition density affects the in-situ volume and
596 compression behavior of the material, indicating a slowly convergent behavior for the
597 typical stress levels used in dry stacks while a total convergence in shearing and
598 compression was noticed only at very high stress levels;
- 599 - The grain size distribution was evaluated for looser and denser samples of each tailing type
600 after being tested in isotropic stress to 120 MPa. It was found that there was no significant
601 difference in the particle breakage for the flotation tailings due to differences in initial

602 density. Also, the slime tailings showed no particle breakage after the test which is
603 probably related to the increased packing efficiency and coordination number resulting
604 from the greater fines content of the sample compared to the flotation tailings that showed
605 particle breakage;

- 606 - Both tailings showed a unique CSL in the $p'-q$ plane. The flotation tailings have an M_{tc} of
607 1.30 with critical state friction angle ϕ'_{cs} of 32.2° , while the slime tailings had an M_{tc} of
608 1.35 and $\phi'_{cs} = 33.4^\circ$. The similar values indicate that the critical state angle depends
609 mainly on the origin of the material and an increase in fines content changes this value
610 only slightly. The slime tailings also had slightly larger values for the maximum stress
611 ratios for the same dilatancy. This behavior may be related to better interlocking on slime
612 tailings due to the presence of more fines between the sand grains;
- 613 - The critical state line (CSL) in the $v - \ln p'$ plane was fitted with a power-law with the
614 parameters $a = 1.95$, $b = 0.15$ and $c = 0.232$ for the slime tailings and $a = 2.00$, $b = 0.15$
615 and $c = 0.245$ for the flotation tailings. The wide range of initial void ratios tested indicates
616 the absence of transitional mode of behavior on shearing, even if the compression curves
617 at lower stress levels are not unique. The CSLs of the two tailings have identical slopes,
618 different exponents and different intercepts. They are almost parallel at low pressures with
619 the intercept shifting downwards. As the stress level increases, the coarser material has a
620 greater change in gradient, while for the finer there is a smaller difference in the CSL
621 gradient.
- 622 - The projection of the NCLs and CSLs were considered up to high pressures. The curves
623 for the CSLs showed tendency of convergence at high stresses to a unique CSL for both
624 materials parallel to the limiting NCL and with spacing ratio of 2.71. This indicates that
625 for these tailings the behavior at very high stresses is insensitive to the initial grading,
626 density and mineralogy, while at lower stresses those factors affect the behavior
627 significantly.
- 628 - Finally, the results reinforce the need for a broader investigation of different tailings at
629 different initial states for application in dry stacks. Both the fines content and initial void
630 ratio influenced the mechanical response of the tailings for usual pressures in engineering
631 applications, being required extremely high pressures to achieve the classical critical state
632 behavior for the materials studied. A better understanding of tailings behavior in its full
633 range of possible states is therefore required to predict the performance of the projected
634 Tailings Storage Facility.

635 **Data availability statement:** Some or all data, models, or code that support the findings of
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639 **Declarations**

640 Conflict of interest: The authors declare that they have no known competing financial interests
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642 **NOTATION**

643	CSL	critical state line
644	CIU	isotropically consolidated undrained triaxial compression test
645	CID	isotropically consolidated drained triaxial compression test
646	CKU	K-consolidated undrained triaxial compression test
647	DEM	discrete element method
648	FC	fines content
649	F	flotation tailings
650	IOT	iron ore tailings
651	NCL	normal compression line
652	PI	plasticity index
653	QF	region of Cuadrilátero Ferrífero
654	S	slime tailings
655	TFC	threshold fines content
656	D	dilation rate = $d\varepsilon_v/d\varepsilon_s$
657	D_{min}	maximum dilation rate
658	e	void ratio
659	e_0	void ratio prior to consolidation
660	e_{min}	minimum void ratio
661	e_{max}	maximum void ratio
662	G	elastic shear modulus
663	m	slope of convergence lines
664	M	critical state stress ratio
665	M_{tc}	critical state stress ratio at triaxial compression
666	ϕ_{cs}'	critical state friction angle
667	ν	specific volume = $1 + e$
668	η	Stress ratio = q/p'
669	η_{max}	Top stress ratio = q/p'
670	p'	mean effective stress
671	p_0'	mean effective stress at the beginning of the shearing phase
672	q	deviatoric stress ($\sigma_1 - \sigma_3$)
673	σ'_1, σ'_3	principal effective stresses

674	σ'_h	horizontal effective stresses
675	σ'_v	vertical effective stresses
676	u	pore water pressure
677	ε_a	axial strain
678	ε_s	shear strain
679	ε_v	volumetric strain
680	w	water content
681	w_L	liquid limit
682		

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