Forced silo discharge: Simulation and theory

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We study, through discrete element simulations, the discharge of granular materials through a circular orifice on the base of a cylindrical silo forced by a surcharge. At the beginning of the discharge, for a high granular column, the flow rate Q_{ini} scales as in the Beverloo equation for free discharge. However, we find that the flow rate Q_{end} attained at the end of the forced discharge scales as $\sqrt{\rho_{\mathrm{b}} P} D_{\mathrm{o}}^3 / D_{\mathrm{s}}$, with ρ_{b} the bulk density, P the pressure applied by the overweight, D_{o} the orifice diameter and D_{s} the silo diameter. We use the work–energy theorem to formulate an equation for the flow rate Q_{end} that predicts the scalings only in part. We discuss the new challenges offered by the phenomenology of strongly forced granular flows.

I Introduction

It is well known that granular materials in a silo discharge through an orifice, under the sole action of gravity, at a constant mass flow rate, unlike inviscid fluids [1, 2]. In such free discharge conditions, the mass flow rate Q scales as $D_{\mathrm{o}}^{5/2}$, D_{o} being the diameter of the circular orifice [1, 3]. In addition, the flow rate does not appear to be correlated with the local pressure around the discharge orifice [4, 5]. The Hagen–Beverloo equation for free discharge states

$$Q = C\rho_{\mathrm{b}}\sqrt{g}(D_{\mathrm{o}} - kd)^{5/2}. \qquad (1)$$

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Here, d is the mean particle diameter, g the magnitude of acceleration of gravity, and ρ_{b} the bulk density. The parameters C and k are generally fitted to the experimental data. In general, $k \approx 1.5$ is suitable for spherical grains and $C \approx 0.58$ is a well established universal value with recent theoretical support [6].

It is particularly interesting that Q does not depend on silo diameter D_{s} as long as $D_{\mathrm{s}} > 2D_{\mathrm{o}}$. Furthermore, the material properties of the grains only come into the equation through d and the density ρ of the material the grains are made of. Q is directly proportional to ρ because $\rho_{\mathrm{b}} = \rho\phi$, with ϕ being the packing fraction of the granular column. We note the contrast with liquids, whose mass flow rate depends on $\sqrt{\rho}$. The fact that Q in a free discharge does not depend on other properties of the material, such as the Young modulus, the friction coefficient or the Poison ratio, has been explained in [6, 7].

When the free surface of the granular column in the silo is pressed by an overweight (e.g., a heavy piston), the flow rate accelerates over the final stages of the discharge [7, 8]. The presence of the overweight leads to a pressure distribution in the silo that has recently been experimentally investigated and contrasted with early models [9]. Interestingly, forcing leads to flow rates that depend on material properties other than just ρ [7].

Apart from using a simple overweight to force the

flow, researchers have tested a suspended-wall silo with overweights [10] and the use of a layer of heavier grains on top of the granular column [11]. These two ways of forcing the flow do not show an acceleration of the flow during discharge but, rather, a constant flow rate (over the discharge time) that increases with the applied pressure. Interestingly, while in [10] the flow rate grows linearly with the pressure at the base, in [11] it increases with the square root. Still , there exists a lack of experimental evidence, theoretical models and consensus regarding the phenomenology of forced silo discharges.

In this work, we performed simulations using the Discrete Element Method (DEM) to explore the mass flow rate Q for a forced silo discharge as a function of D_s, D_o, ρ and pressure P exerted by an overweight. We show that there exist two limiting cases: (i) at the beginning of the discharge, when the column of grains is very tall, the flow rate Q_{ini} is similar to that of Beverloo without any effect due to the overweight; (ii) at the end of the discharge, the flow rate Q_{end} depends on the applied pressure and material density in a way similar to that of a liquid. Interestingly, the scaling with orifice diameter and silo diameter is unexpected for Q_{end}. By means of the work–energy theorem, we derive a differential equation for the mass flow rate Q as a function of time during a forced discharge. To this end, we followed an approach similar to that used in [6] for unforced discharges and the stress model experimentally validated in [9] for forced flows. We show that the theory predicts the existence of both limiting cases and part of the scalings for Q_{end}. We discuss some challenges faced in modeling strongly forced granular flows.

II Simulations

We consider a cylindrical silo (diameter $D_s = 2R_s$) with a flat base, like the one sketched in Fig. 1. The base has an orifice of diameter D_o (radius R_o) in its center. The silo is filled with an initial mass M_{ini} of a granular material (each grain has mass m and diameter d) and a piston of mass M_p is placed as an overweight on top of the free surface. The piston is slightly narrower than the silo so that grains cannot fit into the gap between the piston and the silo walls.

We run DEM simulations of the forced silo discharge. We utilized LIGGGHTS [12] with a particle–particle Hertz interaction and Coulomb criterion using a Young modulus $Y = 70$ MPa, Poisson ratio $\nu = 0.25$, restitution coefficient $e = 0.95$ and friction coefficient
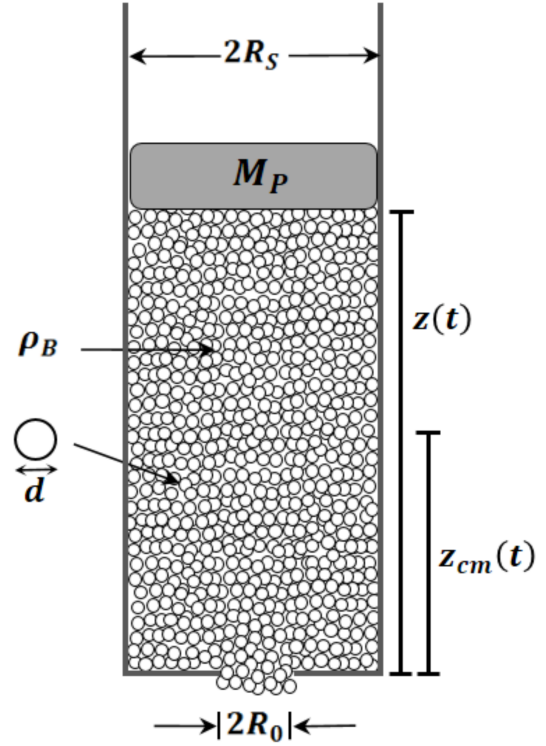

Figure 1: Sketch of the axial cross section of a cylindrical silo with granular material and overweight.

$\mu = 0.5$. Unless otherwise stated, the same interaction applies for the particle–walls contacts. In terms of dimensional units, the particle diameter is set to $d = 1.0$ mm. Particles (up to 3×10^5 grains, depending on the silo diameter) are poured into the silo with the orifice initially blocked by a plug. Then, a piston of desired weight M_p is placed on top of the granular column. The piston has no friction with the silo walls. We let the grains rest in the silo, waiting until the kinetic energy per particle falls below 10^{-10} J. Then, we remove the plug and allow the material to discharge through the orifice. The magnitude of the acceleration of gravity is set to $g = 9.81$ m/s^2 and the integration time step is $\Delta t = 5 \times 10^{-6}$ s. Different silo diameters, grain material densities, orifice diameters, and piston weights are tested. During the entire discharge, we record the particle positions, velocities, accelerations, and particle stress tensors.
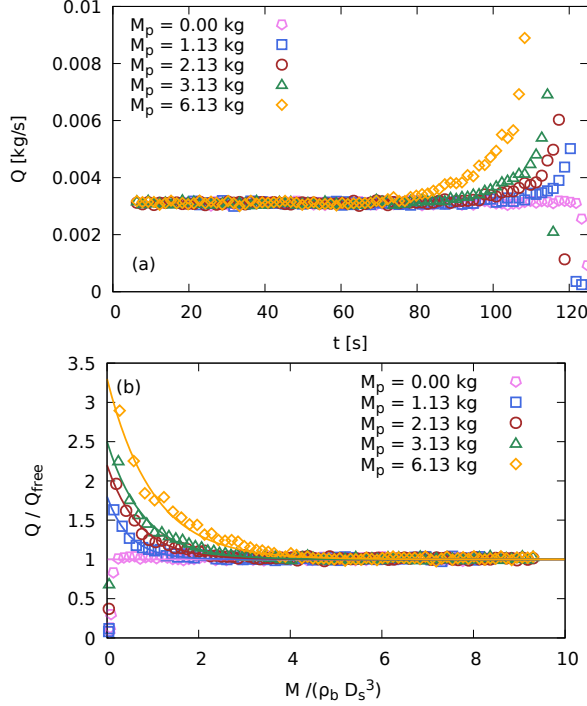
Figure 2: Mass flow rate Q as a function of time (a) and as a function of the instantaneous mass inside the silo (b) for various overweight masses M_p. In part (b) Q is scaled by the mass flow rate Q_{free} of a free discharge through the same orifice. Simulations correspond to $D_s = 30d$, $D_o = 6d$ and $\rho = 2500$ kg/m^3. The solid lines correspond to fitting the function $Q(M) = (Q_{end} - Q_{ini}) \exp[-M/M_{decay}] + Q_{ini}$.

III Results

In Fig. 2, we present some sample discharge flow rate Q as a function of time and as a function of the mass M of grains inside the silo (a proxy for the column height). As in previous studies [7], we observe a constant flow rate at the beginning of the discharge followed by a rapid increase when the column height decreases. At the very end of the discharge, a sharp drop to zero flow is observed due to the fact that the silo runs out of particles. In Fig. 2(b), we have normalized Q by the one obtained in a free discharge (i.e., for $M_p = 0$). It is clear that at the beginning of the discharge $Q = Q_{free}$ and the forcing due to M_p does not matter.

To characterize the two limiting flow rate values, we extract the initial flow rate Q_{ini} at very tall columns and the final flow rate Q_{end} at zero column height (at the end of the acceleration process, just before the final drop to zero). To this end, we fit the simula-
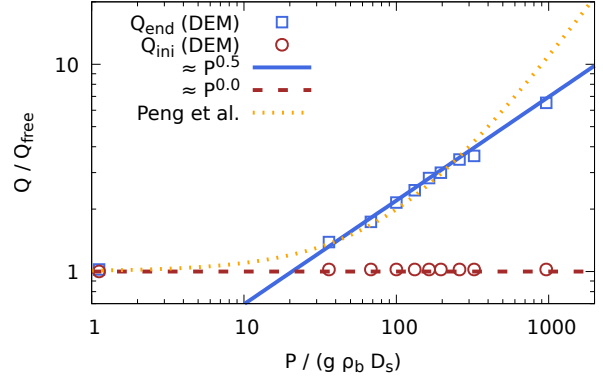

Figure 3: Mass flow rate for a tall column Q_{ini} and at the end of the discharge Q_{end} scaled by that of a free discharge Q_{free} as a function of the pressure P applied by the overweight. Simulations correspond to $D_o = 6d$, $D_s = 30d$ and $\rho_b = 2500$ kg/m^3. The yellow dotted line corresponds to Peng et al. model [10]. Thick lines correspond to different power laws.

tions curves Q vs. M using the exponential function $Q(M) = (Q_{end} - Q_{ini}) \exp[-M/M_{decay}] + Q_{ini}$ and extract the two limits $Q_{ini} = Q(M \to \infty)$ and $Q_{end} = Q(M \to 0)$ (see solid lines in Fig. 2(b)). We ran simulations for various D_o, D_s, ρ_b, and M_p and extracted Q_{ini} and Q_{end} in each case.

First, we consider the effect of the applied pressure $P = 4M_p/\pi D_s^2$ as shown in Fig. 3 for given D_o, D_s and ρ_b. As we mentioned above, the initial flow rate Q_{ini} is independent of P. However, pressure induces an increase in the final flow rate Q_{end}, as seen by others [8,10,11]. In Fig. 3 we have used log–log scale to highlight the power law at high pressures, which is compatible with $\sqrt{P}$. This is in accordance with the discussion presented by Horabik et al. [11] but contrasts with the linear dependency suggested by Peng et al. [10]. A remarkable feature is that this square root scaling with P is akin to the one observed in viscous fluids, while a free discharge presents a flow rate independent of pressure [4,13]. It is worth mentioning that at low forcing pressures, Q_{end} tends to Q_{free}, as observed in experiments [8,10], and the square root scaling ceases. Peng et al. proposed a base flow rate at zero pressure which corresponds to the free discharge corrected by a linear term αP. This empirical equation works well at low pressures (see yellow dotted line in Fig. 3) but fails at large P. We will show in Section IV that the Work-Energy theorem can predict the observed square root dependency for high P.
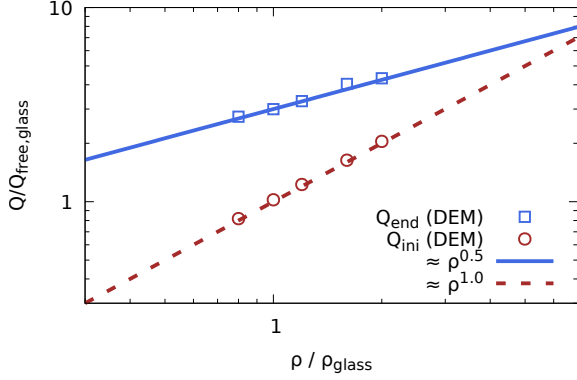
Figure 4: Q_{end} and Q_{ini} scaled by Q_{free} of glass as a function of the material density of the particles ρ scaled by the density of glass ($\rho_{\mathrm{glass}} = 2500$ kg/m^3). Simulations correspond to $D_{\mathrm{o}} = 6d$, $D_{\mathrm{s}} = 30d$ and $P = 85.0$ kPa. Lines correspond to different power laws.

Figure 5: Q_{end} and Q_{ini} scaled by Q_{free} at $D_{\mathrm{o}} = 6d$ as a function of the empty annulus corrected orifice diameter $D_{\mathrm{o}} - kd$ scaled by d. Simulations correspond to $\rho = 2500$ kg/m^3, $D_{\mathrm{s}} = 30d$ and $P = 85.0$ kPa. Lines correspond to different power laws.

In view of the previous result, which indicates that Q_{end} resembles a viscous flow, we consider the effect of density. We recall that, in a free discharge, the mass flow rate is proportional to the density of the granular material [3], whereas in a viscous flow it is proportional to $\sqrt{\rho}$. Figure 4 shows Q_{end} and Q_{ini} as a function of ρ on a logarithmic scale. As we can see, the scaling is also compatible with a viscous flow with $Q_{\mathrm{end}} \propto \sqrt{\rho}$ while $Q_{\mathrm{ini}} \propto \rho$.

Having observed a viscous-like behavior for Q_{end} with P and ρ, we explore the effect of the orifice diameter that we would expect to present a quadratic form as in a fluid. Figure 5 shows Q_{ini} and Q_{end} as a function of $D_{\mathrm{o}} - kd$. We have introduced the empty annulus correction kd, as done by Beverloo [3]. Surprisingly, the scaling is compatible with $Q_{\mathrm{end}} \propto (D_{\mathrm{o}} - kd)^3$. This is neither the expected quadratic dependency observed in fluids, nor the $5/2$ power observed in a free granular discharge. It is hard to distinguish power laws with exponents 3 and $5/2$. However, our data seems to be better fitted by $Q_{\mathrm{end}} \propto (D_{\mathrm{o}} - kd)^3$. As a reference, we also include in Fig. 5 the flow rate for tall columns Q_{ini}. In this case, the forcing pressure does not affect the flow, which remains the same as for a free discharge, and the observed $5/2$ scaling is compatible with the Beverloo scaling. In the next section, we will propose a plausible explanation for the origin of this unexpected scaling.

Finally, we consider the effect of the silo diameter D_{s}. Although in a free discharge D_{s} is not relevant except for silos in which D_{o} approaches D_{s} [14], we expect some effect in forced silos based on previous

studies [10]. In Fig. 6, we present Q_{ini} and Q_{end} as a function of D_{s}. As we expected, Q_{ini} is independent of D_{s}. However, Q_{end} falls as D_{s}^{-1}. Peng et al. found an exponential decay, for a suspended-wall forced silo, rather than a power law [10]. We will show that the work–energy theorem predicts a power-law dependency on D_{s}, but with a different exponent. However, we note that this power law decay has to break down at very large D_{s} since Q_{end} must necessarily converge to a constant limiting value Q_{free} set by a free discharge.

To summarize, in a forced discharge, while Q_{ini} conserves all the scalings known for a free discharge, the behavior of Q_{end} seems to be compatible with

$$ Q_{\mathrm{end}} = \chi \sqrt{\rho_{\mathrm{b}} P} \frac{(D_{\mathrm{o}} - kd)^3}{D_{\mathrm{s}}}. \qquad (2) $$

This scaling works for $D_{\mathrm{s}} < 10 D_{\mathrm{o}}$ and $P > 10 g \rho_{\mathrm{b}} D_{\mathrm{s}}$. According to our simulation results, $\chi \approx 0.25$. Figure 7 collects the Q_{end} values for the range of control parameters investigated and compares them against Eq. (2).

IV Work-energy theorem

Consider the silo sketched in Fig. 1. Let D_{s} be its diameter, $R_{\mathrm{s}} = D_{\mathrm{s}}/2$ its radius, and $A_{\mathrm{s}} = \pi R_{\mathrm{s}}^2$ its cross section. The base has an orifice of diameter D_{o} (radius $R_{\mathrm{o}} = D_{\mathrm{o}}/2$ and cross section $A_{\mathrm{o}} = \pi R_{\mathrm{o}}^2$) in its center. The silo is filled with an initial mass M_{ini} of a granular material (each grain has mass m) and a piston of mass M_{p} is placed as an overweight on top of the free
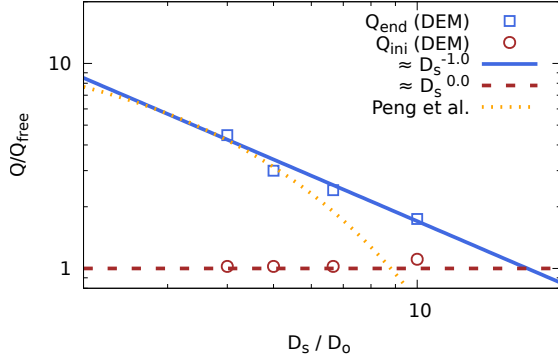
Figure 6: Q_{end} and Q_{ini} scaled by Q_{free} as a function of the silo diameter D_{s} scaled by D_{o}. Simulations correspond to $\rho = 2500$ kg/m^3, $D_{\text{o}} = 6d$ and $P = 85.0$ kPa. The yellow dotted line corresponds to Peng et al. [10]. The thick lines correspond to different power laws.
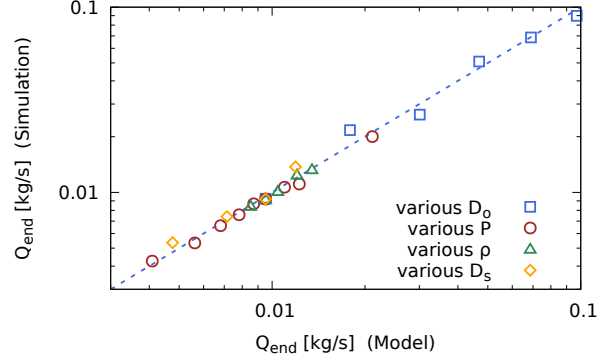

Figure 7: Q_{end} from simulation versus the model scaling of Eq. (2) with $\chi = 0.25$ for a range of material densities, orifice diameter, silo diameter and applied overweight pressure.

surface. The grains cannot fit into the gap between the piston and the silo walls. As an approximation, we assume that the bulk density ρ_{b} of the granular material is homogeneous.

During the discharge, the mass $M(t)$ inside the silo at time t can be written as

$$M(t) = \rho_{\text{b}} A_{\text{s}} z(t) = 2\rho_{\text{b}} A_{\text{s}} z_{\text{cm}}(t), \qquad (3)$$

where $z(t)$ is the height of the column of grains, and $z_{\text{cm}}(t) = z(t)/2$ is the vertical position of the center of mass of the granular column at time t. Therefore, the mass flow rate $Q(t)$ is

$$Q(t) = -\dot{M}(t) = 2\rho_{\text{b}} A_{\text{s}} v_{\text{cm}}(t), \qquad (4)$$

where $v_{\text{cm}}(t) = |\dot{z}_{\text{cm}}(t)|$ is the speed of the center of mass of the granular column. We note that since $M(t)$ decreases with time during the discharge, $\dot{M}$ and $\dot{z}_{\text{cm}}$ are negative.

Let us consider the system composed of the grains inside the silo at any given time. The work–energy theorem states that the change in kinetic energy $\dot{K}_{\text{in}}$ of this system is

$$\dot{K}_{\text{in}} = W_{\text{g}} + W_{\text{p}} - W_{\text{out}} - W_{\text{el}} - W_{\text{d}}, \qquad (5)$$

where W_{g} is the power injected by the force of gravity acting on the grains, W_{p} is the power injected by the weight of the piston, W_{out} is the power loss due to the grains that leave the silo through the orifice at a velocity v_{out}, W_{el} is the "elastic power", i.e., the rate of change of the elastic energy of the grain–grain contacts, and

W_{d} is the dissipated power due to the non–conservative interactions (friction and inelastic collisions between the grains and between the grains and the walls). In the following subsections we will consider in detail each of the terms of Eq. (5).

i   Internal kinetic energy ($\dot{K}_{\text{in}}$)

Figure 8(a) displays different power contributions extracted from a DEM simulation during a forced silo discharge. The fluctuations in kinetic energy K_{in} and elastic energy E_{el} are large relative to the mean, which makes time derivatives to calculate the power not very reliable. Therefore, to estimate $\dot{K}_{\text{in}}$ and W_{el} we plot $K_{in}(t)$ and $E_{\text{el}}(t)$ as functions of time in Figure 8(b) and use the slope of a fitting straight line as a mean time derivative. As observed in Fig. 8(b), neither K_{in} nor E_{el} are linear in time at the end of the discharge, but we fit only the first half of the discharge time to have a representative slope. Simulations using other silo diameters, overweight and material density display the same trends as observed for free discharge in [6]. As we can see in Fig. 8, the contribution of $\dot{K}_{\text{in}}$ is orders of magnitude smaller than any other contribution. Based on this, we will neglect $\dot{K}_{\text{in}}$ in Eq. (5). To provide a partial explanation for this observation, consider the kinetic energy of the grains inside the silo

$$K_{\text{in}}(t) = \frac{1}{2} \sum_{i=1} m \boldsymbol{v}_i^2(t). \qquad (6)$$

The sum runs over all particles inside the silo at time t and $\boldsymbol{v}_i(t)$ is the velocity of particle i. This can be expressed in terms of the center of mass velocity v_{cm}
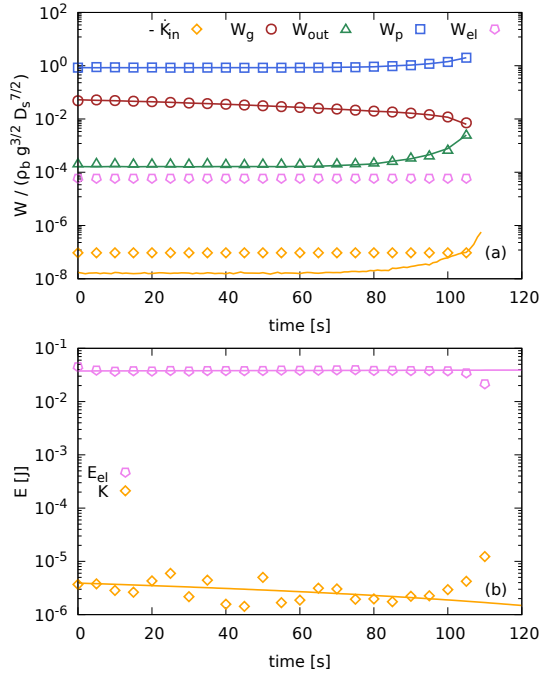
Figure 8: (a) Power contributions during the discharge of a forced silo with $M_{\mathrm{p}} = 6.0$ kg as a function of time (DEM simulations). (b) Smoothed kinetic energy $K_{\mathrm{in}}(t)$ and elastic energy $E_{\mathrm{el}}(t)$ with linear fits to estimate $\dot{K}_{\mathrm{in}}$ and W_{el} from the slope, shown as constant values in part (a). The solid lines in part (a) correspond to the theoretical estimates for: $W_{\mathrm{p}}(t)$ [Eq. (16), blue]; $W_{\mathrm{g}}(t)$ [Eq. (12), maroon]; $W_{\mathrm{out}}(t)$ [Eq. (18), green] and $\dot{K}_{\mathrm{in}}(t)$ [second term of Eq. (10), yellow].

and the "granular temperature" as $K_{\mathrm{in}}(t) = K_{\mathrm{in}}^{\mathrm{cm}}(t) + K_{\mathrm{in}}^{\mathrm{temp}}(t)$, being

$$K_{\mathrm{in}}^{\mathrm{cm}}(t) = \frac{1}{2} M(t) v_{\mathrm{cm}}^2(t), \qquad (7)$$

and

$$K_{\mathrm{in}}^{\mathrm{temp}}(t) = \frac{1}{2} \sum_{i=1}^{N(t)} m_i [\boldsymbol{v}_i(t) - \boldsymbol{v}_{\mathrm{cm}}(t)]^2. \qquad (8)$$

We will disregard the term "temperature" and focus on the contribution of the center of mass. We assume here that the x and y components of $\boldsymbol{v}_{\mathrm{cm}}$ are null and, therefore, $|\boldsymbol{v}_{\mathrm{cm}}| = v_{\mathrm{cm}} = -\dot{z}_{\mathrm{cm}}$.

$$\dot{K}_{\mathrm{in}}(t) \approx \dot{K}_{\mathrm{in}}^{\mathrm{cm}}(t)$$
$$= M(t) v_{\mathrm{cm}}(t) \dot{v}_{\mathrm{cm}}(t) + \frac{1}{2} v_{\mathrm{cm}}^2(t) \dot{M}(t), \quad (9)$$

which can be written, using Eqs. (3) and (4), as

$$\dot{K}_{\mathrm{in}}(t) \approx \frac{1}{4\rho_{\mathrm{b}}^2 A_{\mathrm{s}}^2} \left[M(t) \dot{M}(t) \ddot{M}(t) + \frac{1}{2} \dot{M}^3(t) \right]. \qquad (10)$$

As we can see, $\dot{K}_{\mathrm{in}}$ decays as A_{s}^{-2}. Therefore, $\dot{K}_{\mathrm{in}}$ will be small for wide silos. We will see below that other terms in the energy balance fall more smoothly with A_{s}. Hence, $\dot{K}_{\mathrm{in}}$ can be neglected as a first approximation. In practice, we can neglect the first term in the brackets of Eq. (10) and keep only the second term. This contribution is still small for large D_{s} and it can be proved that it corresponds to the known correction factor of the flow rate of an orifice-plate configuration for fluids given by $1/\sqrt{1 - (D_{\mathrm{s}}/D_{\mathrm{o}})^4}$ [15].

In Fig. 8(a), we test the prediction of the second term in Eq. (10). To plot this expression, we use in Eq. (10) the instant value of $\dot{M}(t)$ measured in the simulation. As we can see, the prediction is far from perfect, but it gives the correct order of magnitude and allows us to justify neglecting $\dot{K}_{\mathrm{in}}$ in the energy balance.

ii Gravitational energy (W_{g})

The gravitational potential energy of the particles inside the silo is

$$U_{\mathrm{g}}(t) = M(t) g z_{\mathrm{cm}}(t) = \frac{g M^2(t)}{2\rho_{\mathrm{b}} A_{\mathrm{s}}}, \qquad (11)$$

wherein we have used Eq. (3). Therefore, the power injected by the action of gravity is

$$W_{\mathrm{g}}(t) = -\dot{U}_{\mathrm{g}}(t) = -\frac{g}{\rho_{\mathrm{b}} A_{\mathrm{s}}} M(t) \dot{M}(t). \qquad (12)$$

As we can see, W_{g} scales with A_{s}^{-1} in contrast to the faster decay displayed by $\dot{K}_{\mathrm{in}}$. We recall here that $\dot{M}(t)$ is negative; therefore, $W_{\mathrm{g}}(t)$ is positive. In Fig. 8(a), we show that Eq. (12) is consistent by using the instant values of $M(t)$ and $\dot{M}(t)$ measured in the simulation.

iii Overweight energy (W_{p})

The force applied by the piston on the granular column is

$$F_{\mathrm{p}} = M_{\mathrm{p}}(g - a_{\mathrm{p}}(t)) = M_{\mathrm{p}}(g - \ddot{z}(t)). \qquad (13)$$

Here, a_{p} is the acceleration experienced by the piston during discharge. The power injected can be calculated as

$$W_{\mathrm{p}}(t) = -F_{\mathrm{p}} \dot{z}(t) = M_{\mathrm{p}}(g - \ddot{z}(t)) \dot{z}(t). \qquad (14)$$

Using Eq. (3), the power injected by the piston is

$$W_{\mathrm{p}}(t) = -\frac{gM_{\mathrm{p}}}{\rho_{\mathrm{b}}A_{\mathrm{s}}}\dot{M}(t) + \frac{M_{\mathrm{p}}}{\rho_{\mathrm{b}}^2 A_{\mathrm{s}}^2}\dot{M}(t)\ddot{M}(t). \quad (15)$$

We have tested in simulations that the second term in Eq. (15) is small enough and can be neglected as a first-order approximation. Hence,

$$W_{\mathrm{p}}(t) \approx -\frac{gM_{\mathrm{p}}}{\rho_{\mathrm{b}}A_{\mathrm{s}}}\dot{M}(t). \quad (16)$$

As shown in Fig. 8(a), the expression in Eq. (16) is an excellent approximation.

iv Discharge energy (W_{out})

While the system discharges, the particles that leave the system take away some energy since their own kinetic energy is no longer part of the internal energy of the silo. If during a time interval dt the system discharges a mass $dM = -\dot{M}(t)dt$ at velocity v_{out}, then the kinetic energy removed per unit time is

$$\dot{K}_{\mathrm{out}}(t) = -\frac{1}{2}v_{\mathrm{out}}^2\dot{M}(t). \quad (17)$$

By definition, $-\dot{M}(t) = \rho_{\mathrm{o}}A_{\mathrm{o}}v_{\mathrm{out}}$. Therefore,

$$W_{\mathrm{out}} = \dot{K}_{\mathrm{out}}(t) = -\frac{\dot{M}^3(t)}{2\rho_{\mathrm{o}}^2 A_{\mathrm{o}}^2}. \quad (18)$$

The discharge power does not depend on the silo cross section, but on the orifice cross section A_{o}. Then again, W_{out} is positive since $\dot{M}(t)$ is negative. In Fig. 8(a), we show that Eq. (18) is consistent by using the instant value of $\dot{M}(t)$ measured in the simulation.

v Elastic energy (W_{el})

If the grains are stiff, the variation in the elastic energy at the contacts is expected to be small. W_{el} corresponds to the rate of change of the conservative component of the contact forces $\dot{E}_{\mathrm{el}}$. In Fig. 8(a), we show W_{el} during a forced silo discharge in a DEM simulation. As we can see, this term is small in comparison with the other energy contributions. Hence, we will neglect W_{el} as a first approximation. Of course, this may be inadequate for very soft grains. For our purposes, we simply disregard this contribution in the analysis. However, this may eventually be included by using the stress based expression for the elastic energy density in the limit of small deformations [16]

$$E_{\mathrm{el}} = A_{\mathbf{s}}z_{\mathrm{cm}}\boldsymbol{\sigma}\boldsymbol{\epsilon}, \quad (19)$$

being $\boldsymbol{\sigma}$ the stress tensor and $\boldsymbol{\epsilon}$ the strain tensor.

vi Dissipated energy

In previous works [6,7], we have shown that the power dissipated during a silo discharge can be calculated by assuming that the flow is consistent with a quasistatic shear flow in the framework of the $\mu(I)$-rheology model [17]. Here, we present a derivation of W_{d} that considers in detail the effect of the convergent flow close to the orifice and the effect of the forcing piston.

According to the $\mu(I)$-rheology model, in a plane shear configuration where a granular sample is sandwiched between two plates, the average tangential stress σ_{xz} required to move the plates at constant relative velocity v can be written as $\sigma_{xz} = \mu(I)\sigma_{zz}$. Here, σ_{zz} is the confining stress pressing the two plates that constrain the particles, and $\mu(I)$ is an "effective friction coefficient" that accounts for all the complex dissipative interactions in the granular sample between the plates. This effective friction depends on the inertial number $I = vd\sqrt{\rho}/(L\sqrt{\sigma_{zz}})$, with ρ being the density of the material of the grains, d the mean size of the grains, and L the distance between the plates. The inertial number suffices to characterize the flow as long as the grains are stiff and the plate-to-plate distance is much larger than the grain size.

The power W_{d} dissipated during the motion of the plates in plane shear is simply given by $W_{\mathrm{d}} = \sigma_{xz}Av$, where A is the total area of a plate. Hence, the dissipated power can be written as $W_{\mathrm{d}} = \mu(I)\sigma_{zz}Av$. In this expression, the effect of the properties of the granular material on dissipation comes only through $\mu(I)$ since σ_{zz}, A and v are set externally in the experiment.

For spherical grains, it has been shown that $\mu(I)$ follows a universal curve as long as the particle–particle friction coefficient μ is high enough (roughly $\mu > 0.4$) [17]. However, this curve depends on geometrical factors such as dimensionality [18]. Interestingly, in the quasistatic limit (i.e., $I \ll 1$, with $I < 0.01$ being a typical criterion used for practical purposes), the value of $\mu_o \equiv \mu(I = 0)$ converges to a value that is fairly insensitive to material properties such as μ (provided that $\mu > 0.4$), restitution coefficient and Young's modulus. Therefore, the dissipated power becomes fairly
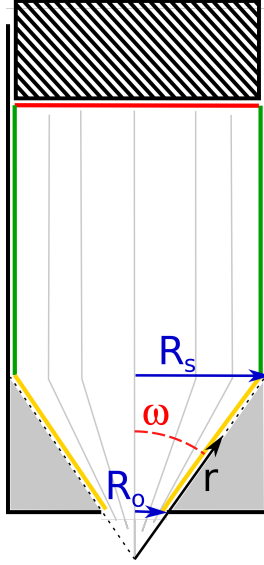
Figure 9: Streamlines during a forced discharge with a stagnant zone and convergent flow. The dissipation due to the shear on the three surfaces highlighted in red, green and yellow are calculated as separated contributions.

independent from the details of the particle–particle interactions for spherical grains.

We postulate that the flow inside a discharging silo can be modeled in a similar way to the simple plane shear geometry. Let us consider the concentric shear due to the velocity gradient in the radial direction between the side vertical walls, highlighted in green in Fig. 9, and the vertical axis of the silo. In the cylindrical silo geometry, the confining pressure can be taken as σ_{rr}, the characteristic driving velocity along the silo axis as the velocity of the free surface of the granular column $v = 2v_{cm}$ and the "plate area" as the area of contact $\pi D_s 2z_{cm}(t)$ between the grains and the lateral silo walls. Therefore,

$$W_d^{\text{side wall}}(t) = \mu_o \langle \sigma_{rr} \rangle(t)[\pi D_s 2z_{cm}(t)]2v_{cm}(t). \tag{20}$$

Here, $\langle \sigma_{rr} \rangle(t)$ is the mean normal wall pressure along the column of grains.

The region close to the base of the silo presents a convergent flow whose dissipation is not taken into account in Eq. (20). A stagnant cone is apparent with a slope that forms an angle ω with the vertical (see Fig. 9). We can calculate the power dissipated due to the shear induced by the conical surface highlighted in yellow in

Fig. 9 as

$$W_d^{\text{cone}}(t) = \int_{R_o/\sin\omega}^{R_s/\sin\omega} v_r(r)\sigma_{r\theta}(r)2\pi r\sin\omega\, dr. \tag{21}$$

Here, we have used a spherical coordinate system with r denoting the distance from the vertex of the cone (which lies below the orifice, outside the silo, see Fig. 9). The velocity in the r-direction is denoted by $v_r(r)$ and the shear stress is denoted by $\sigma_{r\theta}(r)$. Due to the conical flow, mass conservation implies that $v_r(r)$ grows quadratically with decreasing r (see [13] §10.3). Hence,

$$v_r(r) = \frac{-\dot{M}(t)}{\Omega \rho_b r^2}, \tag{22}$$

where $\Omega = 2\pi(1-\cos\omega)$ is the solid angle subtended by the cone.

The shear stress $\sigma_{r\theta}(r)$ should decrease from a finite value σ_o, at the upper part of the cone, next to the wall, down to the orifice. As proposed by Neddermann ([13] §10.3), we assume a linear decrease of the form

$$\sigma_{r\theta}(r) = \sigma_o \frac{r\sin\omega}{R_s}. \tag{23}$$

Plugging (22) and (23) into (21) and integrating, we obtain

$$W_d^{\text{cone}}(t) = -\sigma_o \frac{\pi\sin\omega}{1-\cos\omega}R_s(R_s - R_o)\frac{\dot{M}(t)}{\rho_b A_s}. \tag{24}$$

It can be shown that σ_o is proportional to the vertical stress σ_{zz}^o at the bottom of the silo and proportional to μ_o (see Appendix A). Therefore, we can collect the proportionality constant for σ_o and the factor $\pi\sin\omega/(1-\cos\omega)$ into a single constant α to write

$$W_d^{\text{cone}}(t) = -\alpha\mu_o\sigma_{zz}^o R_s(R_s - R_o)\frac{\dot{M}(t)}{\rho_b A_s}, \tag{25}$$

with α being a constant that depends on the angle of the stagnant cone, the internal angle of friction, and the wall angle of friction (see Appendix A).

In addition to the dissipation due to the lateral walls and the lower conical flow, there is a contribution due to the contact between the forcing piston and the granular material (see the surface highlighted in red in Fig. 9). In this case, the confining pressure due to the piston is gM_p/A_s, with A_s the "plate area". The grains under the piston move tangential to it with a relative velocity

that can be taken as $\gamma 2 v_{\mathrm{cm}}(t)$, with γ being a constant of order one. Therefore,

$$W_{\mathrm{d}}^{\mathrm{piston}}(t) = -\mu_o \gamma g M_{\mathrm{p}} \frac{\dot{M}(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}}. \tag{26}$$

Summing Eqs. (20), (25) and (26) and using Eq. (4), we obtain

$$
\begin{aligned}
W_{\mathrm{d}}(t) =& W_{\mathrm{d}}^{\mathrm{side\ walls}} + W_{\mathrm{d}}^{\mathrm{cone}} + W_{\mathrm{d}}^{\mathrm{piston}} \\
=& -\mu_o \langle \sigma_{rr} \rangle(t) 2\pi R_{\mathrm{s}} \frac{M(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}} \frac{\dot{M}(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}} \\
& -\mu_o \alpha \sigma_{zz}^0(t) 2\pi R_{\mathrm{s}} (R_{\mathrm{s}} - R_{\mathrm{o}}) \frac{\dot{M}(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}} \\
& -\mu_o \gamma g M_{\mathrm{p}} \frac{\dot{M}(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}}. \tag{27}
\end{aligned}
$$

vii Differential equation for $\dot{M}(t)$

Collecting all terms for the energy balance from the previous section, i.e., plugging (12), (16), (18) and (27) into Eq. (5), one can obtain a differential equation for the mass flow rate $-\dot{M}(t)$. As we mentioned before, some of these terms are, in fact, negligible. In particular, we take $\dot{K}_{\mathrm{in}} = 0$ and $\dot{E}_{\mathrm{el}} = 0$. Finally, Eq. (5) can be written as

$$
\begin{aligned}
0 =& -\frac{g}{\rho_{\mathrm{b}} A_{\mathrm{s}}} M(t) \dot{M}(t) - \frac{g}{\rho_{\mathrm{b}} A_{\mathrm{s}}} M_{\mathrm{p}} \dot{M}(t) + \frac{\dot{M}^3(t)}{2\rho_{\mathrm{o}}^2 A_{\mathrm{o}}^2} \\
& + \left[\langle \sigma_{rr} \rangle(t) \frac{M(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}} + \alpha \sigma_{zz}^o(t)(R_{\mathrm{s}} - R_{\mathrm{o}}) + \gamma \frac{g M_{\mathrm{p}}}{2\pi R_{\mathrm{s}}} \right] \\
& \times \mu_o 2\pi R_{\mathrm{s}} \frac{\dot{M}(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}}. \tag{28}
\end{aligned}
$$

Solving for $-\dot{M} \equiv Q(t)$ [19]

$$
\begin{aligned}
-\dot{M}(t) =& \quad \tfrac{\pi \sqrt{2}}{4} \quad \rho_{\mathrm{o}} \sqrt{g} D_{\mathrm{o}}^2 \left[\frac{M(t) + M_{\mathrm{p}}}{\rho_{\mathrm{b}} A_{\mathrm{s}}} \right. \tag{29} \\
& - \quad \frac{\mu_o 2\pi R_{\mathrm{s}}}{g \rho_{\mathrm{b}} A_{\mathrm{s}}} \left(\langle \sigma_{rr} \rangle(t) \frac{M(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}} \right. \\
& + \quad \left. \left. \alpha \sigma_{zz}^o(t)(R_{\mathrm{s}} - R_{\mathrm{o}}) + \frac{\gamma g M_{\mathrm{p}}}{2\pi R_{\mathrm{s}}} \right) \right]^{1/2}.
\end{aligned}
$$

Equation (29) is a first-order differential equation for $M(t)$ that can be closed with an initial condition such as $M(t=0) = M_0$. To solve this equation, it is necessary to know $\langle \sigma_{rr} \rangle(t)$, $\sigma_{zz}^0(t)$, μ_o, α, and γ. In the next section, we review Walters' model for the pressure in discharging silos to close Eq. (29).

It is worth mentioning at this point that Eq. (28) reduces to the equation for an inviscid fluid if the last term that accounts for the dissipated power is neglected and $M_{\mathrm{p}} = 0$. In this case, we obtain

$$
\begin{aligned}
-\frac{\dot{M}(t)^2}{\rho_{\mathrm{o}}^2 A_{\mathrm{o}}^2} &= 2g \frac{M(t)}{\rho_{\mathrm{b}} A_{\mathrm{s}}}, \\
v_{\mathrm{out}} &= \sqrt{2gh(t)}, \tag{30}
\end{aligned}
$$

where $h(t) = M(t)/(\rho_{\mathrm{b}} A_{\mathrm{s}})$ and $v_{\mathrm{out}} = \dot{M}(t)/(\rho_{\mathrm{o}}^2 A_{\mathrm{o}}^2)$.

viii Pressure from Walters model

Walker [20] and Walters [21] have developed models for the pressure during silo discharge (including an overweight) following a similar approach to that of Janssen [22]. These developments went further, in the sense that the Janssen's redirection constant was put in terms of the internal and wall effective friction. Recently, Ferreyra et al. [9] have experimentally validated this model and discussed its shortcomings. The expressions for σ_{zz} and σ_{rr} at a given depth h in the moving column are

$$
\begin{aligned}
\sigma_{zz}(h) =& \frac{g \rho_{\mathrm{b}} D_{\mathrm{s}}}{4B} [1 - \mathrm{e}^{-4Bh/D_{\mathrm{s}}}] \\
& + \frac{4g M_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2} \mathrm{e}^{-4Bh/D_{\mathrm{s}}}, \tag{31}
\end{aligned}
$$

$$\sigma_{rr}(h) = \frac{B}{\tan \phi} \sigma_{zz}(h), \tag{32}$$

where $\tan \phi$ is the effective friction corresponding to the wall yield locus [13], and $B/\tan \phi$ is the equivalent to the Janssen's force redirection factor k, which for discharging silos is

$$
\begin{aligned}
B &= \frac{\tan \phi \cos^2 \delta}{(1 + \sin^2 \delta) - 2y \sin \delta}, \tag{33} \\
y &= \frac{2}{3c} [1 - (1 - c)^{3/2}], \\
c &= \frac{\tan^2 \phi}{\tan^2 \delta}.
\end{aligned}
$$

Here, δ is the effective friction angle for the internal yield locus. As discussed by Nedderman (see [13] §3.7), the wall yield locus ϕ should not be set simply as $\mathrm{atan}(\mu_{\mathrm{wall}})$. Instead, the Jenike's rule should be applied:

$$\tan \phi = \begin{cases} \mu_{\mathrm{wall}} \text{ if } \sin \delta \geq \mu_{\mathrm{wall}} \\ \sin \delta \text{ if } \sin \delta < \mu_{\mathrm{wall}} \end{cases}. \tag{34}$$
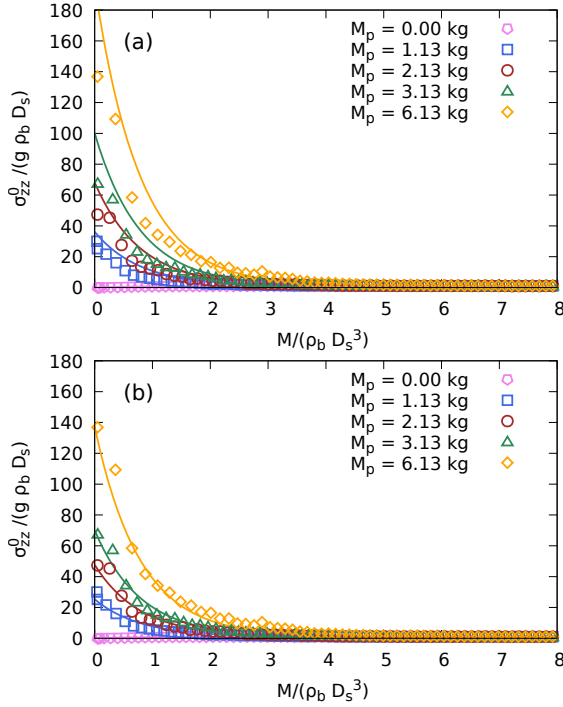
Figure 10: Normal pressure at the bottom of a discharging silo for different forcing overweight M_{p}. Data points correspond to DEM simulations and solid lines correspond to: (a) the Walters model Eq. (31) and (b) corrected formula Eq. (35) to account for the effect of the orifice by setting $\beta = 0.7$. Simulations correspond to $D_{\mathrm{s}} = 30d$, $D_{\mathrm{o}} = 6d$ and $\rho = 2500$ kg/m^3.

Figure 10(a) shows the Walters model along with our simulation data. The particle–wall friction was set $\mu_{\mathrm{wall}} = 0.5$ (as in the simulations). The curves have been fitted by setting $\tan \delta = 0.204$ and $\tan \phi = \sin \delta$, following Eq. (34). This corresponds to $\delta = 11.5°$ and $\phi = 11.3°$ and leads to $B = 0.25$. As we can see, the agreement is not perfect. However, as discussed in [9], we have to bear in mind that the Walters model does not take into account the presence of the orifice, which diminishes the pressure locally, affecting the mean pressure on the base. To account for this effect, we add a correction factor β as

$$\sigma_{zz}(h) = \frac{g \rho_{\mathrm{b}} D_{\mathrm{s}}}{4B} [1 - \mathrm{e}^{-4Bh/D_{\mathrm{s}}}]$$
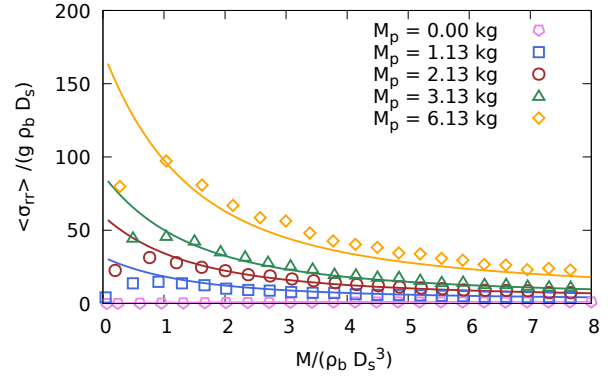$$+ \frac{\beta 4 g M_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2} \mathrm{e}^{-4Bh/D_{\mathrm{s}}}. \tag{35}$$

Figure 11: Mean normal pressure against the side walls as a function of the mass of grains in the silo during discharge with different forcing overweight. Symbols correspond to DEM simulations and solid lines to the modified Walters model Eq. (36) with $\beta = 0.7$. Simulations correspond to $D_{\mathrm{s}} = 30d$, $D_{\mathrm{o}} = 6d$ and $\rho = 2500$ kg/m^3.

By setting $\beta = 0.7$ we obtain a better fit to the data (see Fig. 10(b)).

Since in Eq. (29) we also require the average $\langle \sigma_{rr} \rangle$ over the entire column of grains, we average the expression in Eq. (35) from $h = 0$ to the total depth $h = z$ of the column, and use Eq. (32) to obtain

$$\langle \sigma_{rr} \rangle(z) = \frac{g \rho_{\mathrm{b}} D_{\mathrm{s}}}{4 \tan \phi} \tag{36}$$
$$+ \frac{D_{\mathrm{s}} \left[1 - \mathrm{e}^{-4Bz/D_{\mathrm{s}}} \right]}{4z \tan \phi} \left(\frac{\beta 4 g M_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2} - \frac{g \rho_{\mathrm{b}} D_{\mathrm{s}}}{4B} \right).$$

In Fig. 11 we show $\langle \sigma_{rr} \rangle$ obtained from DEM simulations along with the theoretical predictions from Eq. (36). As we can see, the agreement is very good with the exception of the last part of the discharge, when $M(t) \lesssim 1.5 \rho_{\mathrm{b}} D_{\mathrm{s}}^3$. When the height of the material remaining in the silo falls below $\approx D_{\mathrm{s}}$ the radial pressure decays to zero. This effect is not captured by the Walters model, even after including the correction factor β.

ix Asymptotic pressure

In section III, we have focused on the initial and final flow rates during discharge. These limiting values can be predicted using Eq. (29) by replacing the asymptotic limit of Eqs. (35) and (36). At the beginning of the discharge, for high columns (i.e., $z \gg D_{\mathrm{s}}$), these asymptotic expressions are

$$\sigma^0_{zz}(z \gg D_{\mathrm{s}}) = \frac{g\rho_{\mathrm{b}}D_{\mathrm{s}}}{4B}, \tag{37}$$

$$\langle\sigma_{rr}\rangle(z \gg D_{\mathrm{s}}) = \frac{gD_{\mathrm{s}}}{4\tan\phi}\left[\rho_{\mathrm{b}} + \frac{1}{z}\left(\frac{\beta 4M_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2} - \frac{\rho_{\mathrm{b}}D_{\mathrm{s}}}{4B}\right)\right]$$

At the end of the discharge, when $z \to 0$, we have

$$\sigma^0_{zz}(z \to 0) = \frac{\beta 4gM_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2}, \tag{38}$$

$$\langle\sigma_{rr}\rangle(z \to 0) = \frac{B}{\tan\phi}\frac{\beta 4gM_{\mathrm{p}}}{\pi D_{\mathrm{s}}^2}.$$

We note that the asymptotic value for $\langle\sigma_{rr}\rangle$ does not agree with the simulation results as discussed in Fig. 11. However, we will use this theoretical estimate due to the lack of alternative models in the literature.

V Initial flow rate: the Beverloo scaling

In this section, we consider the flow rate Q_{ini} when the column height is very high ($z \gg D_{\mathrm{s}}$) at the beginning of discharge. By replacing the asymptotic Eq. (37) in Eq. (29), using $z = M(t)/(\rho_{\mathrm{b}}A_{\mathrm{s}})$ and rearranging terms, we obtain

$$Q_{\mathrm{ini}} = -\dot{M}(t) =$$
$$= \frac{\pi\sqrt{2}}{4}\rho_{\mathrm{o}}\sqrt{g}D_0^2\left[\frac{M(t)}{\rho_{\mathrm{b}}A_{\mathrm{s}}}\left(1 - \frac{\mu_o}{\tan\phi}\right)\right.$$
$$+ \frac{M_{\mathrm{p}}}{\rho_{\mathrm{b}}A_{\mathrm{s}}}\left(1 - \frac{\mu_o\beta}{\tan\phi} - \mu_o\gamma\right) \tag{39}$$
$$\left.+ \frac{\mu_o D_{\mathrm{s}}}{B}\left(\frac{1}{4\tan\phi} - \frac{\alpha}{2}\right) + \frac{\mu_o\alpha D_{\mathrm{o}}}{2B}\right]^{1/2}.$$

Equation (39) reduces to the Hagen–Beverloo expression for the flow rate if we select $\mu_o = \tan\phi = 1/(2\alpha)$, $B = 1/4$ and $\gamma = (1-\beta)/\mu_o$. Under these conditions,

$$Q_{\mathrm{ini}} = \frac{\pi\sqrt{2}}{4}\rho_{\mathrm{o}}\sqrt{g}D_0^{5/2}. \tag{40}$$

As we showed in the previous section, the Walters expressions for the pressure fit the DEM data if $\tan\delta = 0.204$. Then, $\tan\phi = 0.20$. Equation (33) then yields $B = 1/4$, which automatically fulfills one of the necessary conditions. Therefore, the well known Hagen's relation will hold if we take $\mu_o = 0.20$ and $\alpha = 2.5$.

Since we set $\beta = 0.7$ to fit the DEM pressure due to the presence of the orifice, the above relations require $\gamma = 1.5$. We note that these values are independent of particular silo conditions, such as the specific values for D_{s}, D_{o}, and the particle–particle or particle–wall interaction. The result of Eq. (40) is equivalent to the one previously found in the case of a free discharge [6].

When calculating our energy contributions, we assumed that the system can be considered as a continuum and the size of the grains d is much smaller than any other length scale in the system. This is one of the assumptions of the local $\mu(I)$-rheology. This approximation is not entirely reasonable when D_{o} is only a few times d. However, the effect of particle size can be introduced by means of a non-local rheology approach [23]. Here, we follow the simplistic correction, in the style of Beverloo, made by replacing the orifice diameter D_{o} by an effective smaller diameter $D_{\mathrm{o}} - kd$. In our simulations, we can fit the data for free discharge using Eq. (40) with $k = 1.72$. We will use this value to correct D_{o} in the rest of the model equations in this work.

As discussed in [6], we can replace $\rho_{\mathrm{o}} \approx \rho_{\mathrm{b}}/2$ in Eq. (29). This is based on measurements carried out on our DEM simulations. Then, in Eq. (29), the non-dimensional prefactor $\frac{\pi\sqrt{2}}{8} \approx 0.56$ can be compared with the constant C in the Beverloo rule. This value deviates less than 4% from the universal value $C = 0.58$ obtained by fitting the experimental data [13].

In Figs. 3 to 6 we have included expression (40) — with the replacements described in the previous paragraph for D_{o} and ρ_{o}— for comparison with the data for the initial flow rate. The agreement is excellent. This is in line with a similar development presented in [6] for freely discharging silos.

VI Final flow rate: inviscid fluid scaling

We now focus on the flow rate Q_{end} at the end of the discharge, when the column of grains vanishes. Replacing the asymptotic Eq. (38) in Eq. (29) we obtain

$$Q_{\mathrm{end}} = -\dot{M}(t) =$$
$$= \frac{\pi\sqrt{2}}{4}\rho_{\mathrm{o}}\sqrt{\frac{gM_{\mathrm{p}}}{\rho_{\mathrm{b}}A_{\mathrm{s}}}}D_{\mathrm{o}}^2 \tag{41}$$
$$\times \left[1 - 2\mu_o\alpha\beta + 2\mu_o\alpha\beta\frac{D_{\mathrm{o}}}{D_{\mathrm{s}}} - \mu_o\gamma\right]^{1/2}.$$

If, as before, we set $\rho_\mathrm{o} \approx \rho_\mathrm{b}/2$ and use the relations and values obtained in the previous section for μ_o, α, β and γ, then

$$Q_\mathrm{end} = \frac{\pi}{8}\sqrt{2\rho_\mathrm{b}P_\mathrm{p}}D_\mathrm{o}^2\left(\beta\frac{D_\mathrm{o}}{D_\mathrm{s}}\right)^{1/2}, \qquad (42)$$

where we defined the pressure applied by the piston as $P_\mathrm{p} = gM_\mathrm{p}/A_\mathrm{s}$. This equation resembles that of an inviscid fluid encountering a plate with an orifice of diameter D_o in a pipe of diameter D_s [15].

Equation (42) predicts that the flow rate depends on the applied piston pressure as a regular fluid. This scaling agrees with the simulation data for high applied pressures (see blue solid line in Fig. 3). However, it is clear that for zero applied pressure, the flow rate cannot drop to zero as suggested by Eq. (42) but must tend to the flow rate of a free discharge. The dependency of Q_end on the density predicted by Eq. (42) is also consistent with the DEM data, as shown in Fig. 4.

Despite the success of Eq. (42) in predicting the pressure and density effect, which contrasts with those observed in free discharges, the dependency on D_o and D_s is not confirmed by the simulation data. While Figs. 5 and 6 show that Q_end scales as D_o^3 and D_s^{-1}, Eq. (42) predicts $D_\mathrm{o}^{5/2}$ and $D_\mathrm{s}^{-1/2}$. The origin of this failure in the model can be related to the actual flow pattern at the end of the discharge. The conical flow pattern depicted in Fig. 9 can be significantly altered when the piston approaches the base (see Fig. 12). Also, as we observed in Fig. 11, the available model for the radial stress fails to predict the behavior at the end of the discharge. Moreover, the approximation that $W_\mathrm{el} = 0$ can be an important missing contribution in the theory, especially for the final stages of the discharge. Developing a fully successful model for the final stages of a forced discharge remains an open challenge.

VII Conclusions

We have presented DEM simulations of forced silo discharges where we vary material density, silo diameter, outlet diameter, and forcing pressure. The flow rate at the beginning of the discharge for a tall column of grains agrees with that of a free discharge. However, at the end of the discharge, the flow rate accelerates and the scaling with the main control variables differs from that of a free discharge. In particular, the final flow rate scales with applied pressure and material density as in a viscous fluid. In addition, the silo diameter becomes
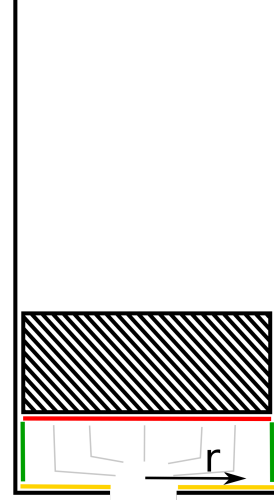

Figure 12: Streamlines when the overweight approaches the base.

a relevant parameter since the flow rate is inverse to it. More surprising yet is the cubic scaling observed with the outlet diameter.

A model based on energy balance and local $\mu(I)$-rheology allowed us to predict the flow rate at the beginning of the discharge. However, this model shows only partial qualitative agreement for the final flow rate. Although the pressure and density scaling are well captured by the model, the effect of silo diameter and discharge outlet diameter is poorly predicted. One may speculate that more sophisticated rheology models may be necessary. However, the usual extension to non-local rheology does not seem to be the natural pathway in these highly forced systems where rapid flow is ubiquitous. These strongly forced granular flows pose new challenges to our understanding of the response of granular materials.

Appendix A

Following Walters [21], the normal σ_{rr} and tangential σ_{rz} stresses in a silo can be put in terms of the vertical stress σ_{zz} according to

$$\sigma_{rr} = \frac{B}{\tan\phi}\sigma_{zz},$$
$$\sigma_{rz} = B\sigma_{zz}. \tag{43}$$

Here, ϕ is the wall friction angle and B is a constant given in Eq. (33) that depends on ϕ and the internal angle of friction δ (see section viii for details).

Since all components of the stress tensor are proportional to σ_{zz}, the shear stress along any plane will be proportional to σ_{zz}. For the plane of the stagnant cone (see yellow line in Fig. 9) that forms an angle ω with the silo axis we obtain

$$\sigma_{r\theta} = \mu_o\sigma_{\theta\theta} \tag{44}$$
$$= \mu_o\sigma_{zz}$$
$$\times \left[\frac{B}{\tan\phi}\cos^2\omega - 2B\sin\omega\cos\omega + \sin^2\omega\right],$$

where we have used the $\mu(I)$-rheology constitutive relation with $\sigma_{\theta\theta}$ that normal stress on the plane of the stagnant cone in spherical coordinates.

Equation (44) is valid next to the silo vertical walls. Hence, σ_{zz} should be taken at the height where the stagnant cone meets the side wall. However, as a proxy, we can simply take the vertical stress σ_{zz}^o at the bottom of the silo and let the factor α defined in the main text to compensate for the deviation.

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